Tomingley Gold Operations Pty Ltd Tomingley Gold Extension Project



Appendix 13 RSF2 Design Report

prepared by

GHD Pty Ltd

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ENVIRONMENTAL IMPACT STATEMENT

Tomingley Gold Operations Pty Ltd Tomingley Gold Extension Project

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TGO RSF 2 Stage 1 Detailed Design Report

Tomingley Gold Operations Pty Ltd

19 January 2022

→ The Power of Commitment



GHD Pty Ltd

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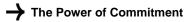
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Executive Summary

Background

GHD has been engaged by Tomingley Gold Operations Pty Ltd (TGO) to undertake the detailed design for the Stage 1 of the Residue Storage Facility 2 (RSF2). This report has been produced to document the Stage 1 embankment design findings.

This report has been prepared to detail the works undertaken as part of the detailed design and construction requirements for the development of the new RSF.

This report is subject to, and must be read in conjunction with, the limitations set out in section 1.5 and the assumptions and qualifications contained throughout the Report.

Consequence Category Review

RSF2 is located to the south of RSF1 and to the west of the Wyoming 1 pit. RSF 2 comprises of two square shaped cells covering a total of 57ha. As part of the detailed design of Stage 1, a detailed dambreak assessment has been undertaken to assess the impacts of a failure of RSF2.

Based on the dambreak assessment, the calculated Population at Risk (PAR) is <1, therefore, with a 'Major' severity of damage and loss to the business and the natural environment due to the release of tailings and contaminated water, the Dam Flood Consequence Category (DFCC) is '**Significant**'. Similarly, the impact of an environmental spill has been assessed as '**Low**' for the Environmental Spill Consequence Category (ESCC) due to the impact extent and the level of damage to the environment.

The consequence category sets the design criteria which is summarised in Section 2.2 this design criteria is based on a '**Significant**' consequence category for both the DFCC and ESCC.

RSF2 Landform Arrangement

Following an Options Assessment to determine the preferred residue storage arrangement, RSF2 has been designed as a paddock facility with a dual cell arrangement to allow for separated tailings deposition between the two cells. Each cell utilises a central decant tower and pump surrounded by drainage rock to allow for the removal of the decant water and rainfall from each of the cells. RSF2 has been developed such that future raises utilise a centreline raise arrangement to optimise stability whilst minimising footprint impacts.

The Stage 1 dam geometry consists of an upstream embankment slope of 2H:1V and a downstream slope of 3H:1V which allows for increased stability in the Stage 1 embankment whilst minimising the storage footprint.

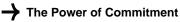
The Stage 1 embankment allows for a 6 m wide crest on the outer and central embankments allowing for 2-way light vehicle traffic, tailings deposition line, safety bunding on each side, and sufficient space to undertake operational and maintenance tasks on the pipeline.

Tailings Management

The dual cell paddock facility allows for the cycling of deposition as required to improve the density of the tailings over the life of the facility. As such the tailings are proposed to be deposited using perimeter discharge with spigots spaced at 50 m centres along the embankment cycling 4 spiggots every 2.5 days to allow for a maximum deposition depth of 300 mm before shifting to the next 4 spiggots. Allowing for the time required for filling of both cells, the 300mm deposition results in a maximum rise rate of 1.95 m/year taking into account the mines maximum throughput capacity.

Water Management

The RSF has been designed to store a 1:200,000 AEP 72 hour storm within the tailings beach and store the Probable Maximum Flood (PMF) within the embankment extents prior to being pumped back into the plant via the decant structure for reuse.



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In addition to this, an emergency spillway has been allowed for on the southern wall of Cell 1 capable of passing a 1:1,000 AEP 72 hour rainfall event from both Cell 1 and Cell 2 with sufficient freeboard for wave run-up during a 1:10 AEP wind event. The spillway consists of a 6 m wide rock-lined trapezoidal spillway which has been designed to convey the design flow event of 1.71 m³/s.

Risk Assessment

A Safety in Design review was carried during the design phase where design improvements were identified to eliminate hazards and improve overall safety. GHD has prepared a summary of the risks identified and mitigation measures adopted or recommended as detailed in the risk register included in Appendix F. Residual risks are presented for TGO's further mitigation in Appendix F which should be reviewed and maintained as a live document prior to construction, throughout construction and operation, and throughout closure of the RSF to ensure the residual risks are addressed.

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- Appendix C RSF2 Geotechnical Investigation Report

- Appendix D RSF2 Slope Stability Analysis
- Appendix E RSF2 Finite Element Modelling
- Appendix F Safety in Design Risk Assessment
- Appendix G RSF2 Technical Specification

1. Introduction

1.1 Overview

Tomingley Gold Operations (TGO) has commenced underground mining in the Wyoming 1 pit at their Tomingley Gold Mine, located approximately 50 km southwest of Dubbo, NSW.

The current active Residue Storage Facility (RSF) has been constructed to Stage 7 height in 2020 with Stage 8 raise planned in 2021/2022 to allow for continuous operation of the mine. GHD Pty Ltd (GHD) has developed the concept designs up to Stage 9 and a concept for a new RSF to cater for further mine development.

Tomingley Gold Operations (TGO) plans to extend their mine production for at least another 7 years of operation to extend their production by 14.9 Mt to FYE2033. GHD have recently completed an initial options assessment and preliminary design for a new Residue Storage Facility (RSF2) immediately south of the existing RSF (herein referred to as RSF1) which was submitted to the NSW Resources Regulator for approval in December 2020. Following this submission and the approval to proceed with the construction of RSF2 Stage 1 and Stage 2, geotechnical investigations were undertaken within the area to the south of RSF1 by GHD to better understand the foundation and construction materials to be used in RSF2. TGO has since engaged GHD Pty Ltd (GHD) to provide engineering services related to the detailed design of RSF2 for the storage of the tailings resulting from the ongoing operations.

This report documents the detailed design of RSF2 Stage 1 to allow for sufficient storage to extend TGO's project life to end of December 2025. This document provides detailed design for Stage 1 of RSF 2 and the conceptual design for Stages 2 to 9 to allow sufficient storage for the proposed extended mining operations. It is GHD's understanding that a further approval application for the Tomingley Gold Extension Project, allowing for the further raising of RSF2 Stage 3 to Stage 9 is in progress.

1.2 Purpose of this report

This report has been prepared to detail the works undertaken as part of the detailed design of the Residue Storage Facility 2 (RSF2) starter dam (herein known as Stage 1).

1.3 Scope of Work

GHD's scope of work for the detailed design of the proposed raise, covered in this report, has comprised:

- Review the available data from the recent geotechnical investigation to confirm design parameters for use in analysis in this report.
- Update the existing Dam Consequence Category in accordance with ANCOLD 2012 Guidelines on the Consequence Categories for Dams (ANCOLD, 2012) and Guidelines on Tailings Dams (ANCOLD, 2019).
- Undertake Stage 1 Embankment Design including:
- Embankment geometry design.
- Foundation lining and drain system design.
- Undertake stability and seepage analysis on the embankment.
- Undertake Finite Element Analysis of the RSF1/RSF2 connecting wall.
- Undertake groundwater analyses.
- Undertake water balance modelling.
- Undertake hydrology and hydraulic design.
- Undertake instrumentation design.
- Undertake conceptual closure design.
- Undertake safety in design risk assessment for RSF 2 Stage 1 embankment.

1.4 Assumptions

The design for RSF 2 Stage 1 assumes the following:

- RSF2 Stage 1 and 2 dam embankment construction material will be won from the dam footprint and existing waste rock dump on site.
- Detailed design of the subsequent stages of the RSF2 embankment will be undertaken prior to the construction of the embankment raise.

1.5 Limitations

This report: has been prepared by GHD for Tomingley Gold Operations and may only be used and relied on by Tomingley Gold Operations for the purpose agreed between GHD and Tomingley Gold Operations as set out in section 1.2 of this report.

GHD otherwise disclaims responsibility to any person other than Tomingley Gold Operations arising in connection with this report. GHD also excludes implied warranties and conditions, to the extent legally permissible.

The services undertaken by GHD in connection with preparing this report were limited to those specifically detailed in the report and are subject to the scope limitations set out in the report.

The opinions, conclusions, and any recommendations in this report are based on conditions encountered and information reviewed at the date of preparation of the report. GHD has no responsibility or obligation to update this report to account for events or changes occurring subsequent to the date that the report was prepared.

The opinions, conclusions, and any recommendations in this report are based on assumptions made by GHD described in this report (refer section(s) 1.4 of this report). GHD disclaims liability arising from any of the assumptions being incorrect.

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Investigations undertaken in respect of this report are constrained by the particular site conditions, such as the location of buildings, services, and vegetation. As a result, not all relevant site features and conditions may have been identified in this report.

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2. Background and Design Basis

2.1 Background

TGO currently produces approximately 1.1Mt of process residue per year, and stores this residue within a twin paddock-style facility known as the Residue Storage Facility (RSF1). RSF 1 is located to the south of the plant area

The existing RSF1 (shown as RSF Cell 1 and RSF Cell 2 in Figure 2.1 below) is approved to Stage 9, Stage 8 is anticipated to be full by June 2022. As a result, RSF2 will be required to be available prior to this to cater for the remaining residue to be stored from future processing. Additional capacity (RSF2 Stages 3 to 9) will be required for the Tomingley Gold Extension Project subject to the relevant approvals being obtained.



The current site layout is presented in Figure 2.1.

Figure 2.1 Current Site Layout

2.1.1 Preferred Site Arrangement

Based on the Options Assessment report outcomes outlined in GHD (2020), Option 1 was identified as the preferred location, as such the new RSF is to be located adjacent to the existing RSF. Due to the topography around the preferred location, the new RSF has been designed as a turkey's nest arrangement minimising hydrological impacts and simplifying depositional and residue management requirements. The facility has been designed such that the southern wall of the existing embankment will be utilised as a part of the new RSF to optimize earthworks and ensure a stable landform arrangement. This will also allow for a single closure landform at the end of mine life. The facility arrangement can be found in Figure 2.2.

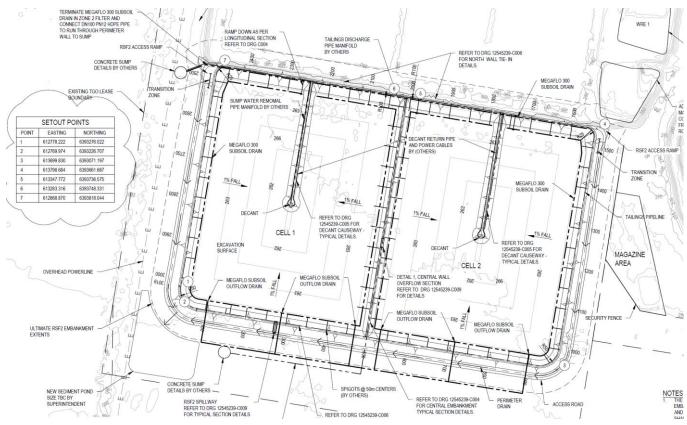


Figure 2.2 Preferred Site Arrangement – Stage 1 Extent Plan

2.1.2 Climate

TGO mine is located on the south-western outskirts of Tomingley town and falls in a warm temperate climate zone. TGO has been recording daily rainfall data at the mine since 2013. Data prior to 2013 have been collected by the Bureau of Meteorology at the Peak Hill Post Office (Station Number 050031) with an average rainfall of 560 mm/year.

2.1.3 Geology

The local area geology map (refer to Figure 2.3 and Table 2.1) was obtained from the NSW Department of Primary Infrastructure and indicates the following:

- A combination of alluvium deposits (Qa), Colluvium (Qv), undifferentiated sediments (Czs), and Ordovician sedimentary rocks (Os) are present.
- Multiple fault zones in close proximity to the mine location.

The map also includes the extents of the existing mining lease as sourced from Planning, Industry & Environment MinView and the proposed future mining lease extents as provided by TGO.

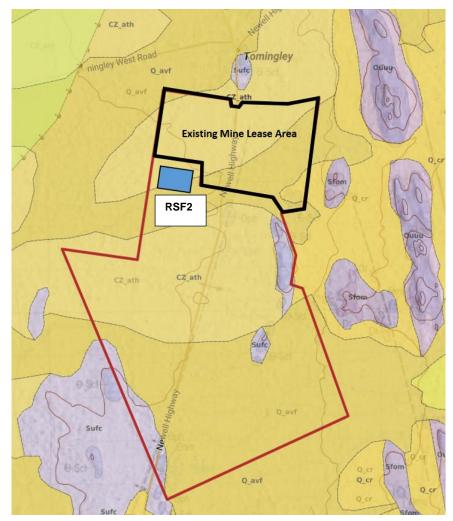


Figure 2.3 Local Geology (Image Sourced from NSW Department of Primary Infrastructure)

Table 2.1	Local Geological Summary
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Unit code	Unit name	Dominant Lithology
Q_avf	Alluvial Terrace deposits-high stand facies	Biogenic sediment
CZ_ath	Alluvial fan deposits	Mud
Sform	Mumbridge Formation	Siltstone
Sufc	Cotton formation	Siltstone
Ouuu	Mugincoble Chert	Chert

A high-level local geological review has been undertaken for the mine site previously as discussed in "Wyoming 3 In-pit Residue Storage Facility Preliminary Design Report". Key findings from this review have been summarised below:

- In the vicinity of the Wyo3 pit (beyond the ore body), alluvial clay / sandy clay and saprolite clay generally extend up to 70 m below ground level (bgl).
- The weathered zone is underlain by fractured siltstone, sandstone, and shale extending beyond 100 m bgl.

2.1.4 Geotechnical Investigations

Previous geotechnical investigations for embankment materials and foundations were undertaken during the initial embankment design and construction phases [ref: Cooper and Associates] as well as the Stage 2-6 raises, which

also included in-situ and laboratory testing for the residue. The previous investigations can be summarized as follows:

- Foundations for the current RSF 1 (cells 1 and 2) generally consisted of stiff clay, beneath 1m thickness of topsoil. The foundations were found to be low permeability, between 2 x 10⁻⁸ and 2 x 10⁻⁹ m/s.
- The residue geochemistry is classified as Non-Acid Forming (NAF).
- The proposed construction material for the embankment raise consists of generally low permeability (less than 1 x 10⁻⁹ m/s as per EPA requirements) sandy clay from the footprint of the TSF area and overburden.
- Investigations of the residue as foundations for upstream raises found that they were generally low strength, low permeability, clayey silt, and the residue is susceptible to liquefaction with low residual shear strength ie, unsuitable.

2.1.5 Hydrogeology and Groundwater Investigations

A groundwater impact assessment report has been previously completed for Wyoming In-pit Tailings Study (GHD, 2016). Key findings from this study related to the existing groundwater conditions at the mine site have been summarized below.

- Perched groundwater occurs within the shallow alluvium, however, it is generally not continuous across the mine site.
- The hydraulic conductivity of the upper clay is generally low to very low.
- A deeper confined saline groundwater system occurs within the fractured sandstone, siltstone, and volcanic materials. This groundwater would be classified as less productive fractured rock groundwater under the NSW Aquifer Interference Policy.
- Coffey (2007) investigated this groundwater system as a potential water supply for the mine and found it to be inadequate in terms of both yield and quality.

2.1.6 Hydrology

Surface water drainage surrounding the mine site typically flows to the southwest. Gundong Creek flows for the north of Tomingley before passing through the north-western section of the mine site. Other named creeks in the vicinity of the Mine Site include Fiddlers and Tomingley Creeks (upstream) to the north and Bulldog Creek to the south. Each of these, together with Gundong Creek, flow to the west and merge with the Bogan River approximately 11 km to the southwest of the mine.

2.2 Design Basis

The design was collated from a number of sources, including TGO input, RSF Stage 6 Raise Design Report (GHD, 2018), Stage 7-9 Concept Report (GHD, 2019), Stage 7 Design Report (GHD, 2021), industry guidelines, GHD's TGO site experience, and GHD's industry experience.

Whilst the majority of parameters and design basis remain consistent throughout the RSF life, it is important to note that some will vary over time and should be checked and verified at each stage of the design, construction and operation. In some cases, various values for the same parameters have been sources and included as an example. In each case, the most recent source should be relied upon for current and future designs.

The assumptions and requirements used as the basis of design supplied and agreed by TGO are presented in Table 2.2.

Table 2.2RSF 2 Basis of Design

Design Aspect	Design Basis	Design Source
Storage Capacity		
Facility Duration	7 years (scalable to 10 years)	TGO Requirement
Ultimate Height	RL 286.0 m (for 10 year storage)	GHD Proposed
Ultimate Capacity	15 Mt	TGO Requirement
Annual Residue Production Rate	1.5 Mtpa (Scaleable to 1.75 Mtpa for short durations)	TGO Requirement
Long-term Residue Stored Dry Density	1.4 t/m ³	Based on density reconciliation of RSF1
Rate of Rise	Maximum 2 m raise/annum (with 700mm freeboard)	Industry experience for upstream and centreline raising
Consequence Category (ANCOLD)	Significant	GHD Assessed
Embankment Arrangement		•
Embankment Type	Centreline	As discussed with TGO
Upstream Face	2:1 (H:V)	GHD Proposed
Downstream Face	3:1 (H:V)	GHD Proposed
Crest Width	6 m	GHD Proposed
Foundation Liner	1 m Compacted Clay Liner (CCL) with permeability of 1x10 ⁻⁹ or better.	Environmental Protection Authority Tailings Dam Liner Policy Letter (2016)
Construction Material	Stage 1 and 2 embankments will be constructed using excavated material from the footprint of the RSF and existing NAF waste rock at the site.	As discussed with TGO
Construction Fleet	Mining fleet where possible, civil construction fleet where geometry does not suit mining fleet.	As discussed with TGO
Mining Fleet	TGO to confirm mining fleet.	TGO to confirm
Closure Arrangement	Water shedding cover with spillway	GHD Proposal
Seismic Loading		1
Operating Basis Earthquake (OBE)	1:475 0.028 PGA	ANCOLD 2019 GHD 2021
Safety Evaluation Earthquake (SEE)	1:10,000 0.180 PGA This PGA is recommended based on the dams susceptibility to the failure from liquefaction of the tailings and foundations.	ANCOLD 2019 GHD 2021
Maximum Credible Earthquake (MCE) (Closure)	1:10,000 0.180 PGA	ANCOLD 2019 GHD 2021
Hydrology		,
Catchment Area	RSF Footprint Area	GHD Assessed
Flood storage requirements	Facility to be designed as a non-release facility, capable of storing a Probable Maximum Precipitation (PMP) event.	GHD Assessed
Decant	Central decant tower. Excess water discharged either to Process Water Dam or Residue Decant Storage Dam.	GHD Assessed
Emergency Spillway	Emergency spillway is designed to pass a 1:1,000 AEP 72 hour Rainfall Event with suitable	ANCOLD (2019)

Design Aspect	Design Basis	Design Source	
	freeboard, following the storage of a PMP rainfall event		
Residue Information and Infrastructure	e		
Residue Classification	Non-Acid Forming (NAF), pumped as a slurry at 45 % solids	RSF1 Stage 7 Pre- Construction Report	
Specific Gravity	2.7		
Beach Angle	1V:140H	Based on a survey of the existing RSF	
Deposition Infrastructure	Deposited from spigots on the perimeter pipe	Based on RSF1 Design	
Deposition Methodology	Sub-aerial deposition	Based on RSF1 Design	
Water Management Philosophy	Decant pond kept as low as possible with water return to Process Water Dam for re-use and excess to Residue Decant Storage Dam	Based on RSF1 Design	
Water Recovery Infrastructure	Central decant tower with a single submersible pump. The submersible pump is to be sized in accordance with the following Normal Operations and Emergency Operations.	Based on RSF1 Design	
	Minimum Normal Operational: 22L/s: Minimum Emergency Operations: 50ML/Day		
Seepage Recovery Infrastructure	Liner graded to the central decant tower at 1% to capture seepage runoff. Underdrainage located at the upstream toe of the stage 1 embankment.	GHD Proposed	
Monitoring	Automated monitoring system where practicable.	GHD Proposed	
Dam Break Modelling			
Failure Modes	Sunny Day Failure (SDF): Piping Flood Failure (FF): Piping	GHD Assessed	
Storm Event (AEP) for Flood Failure Scenario	1% AEP, 0.01% AEP and PMP-DF	GHD Assessed	
Breach Locations	East, South, and West Walls	GHD Assessed	
Water Depth in Impoundment	SDF: Max Operating Level FF: Peak Water Level due to Flood Run-Off	GHD Assessed	

2.3 Design Criteria

The design criteria have been developed and agreed with TGO for the RSF Stage 1 embankment and have been based on the following:

- Currently accepted practices for dam engineering in Australia;
- Australian National Committee on Large Dams (ANCOLD) Guidelines;
- NSW Environmental Protection Authority, Tailings Dam Liner Policy Letter (2016); and
- Dam Safety NSW regulations 2019.

3. Consequence Category Review

3.1 Overview

The consequence category assessments (CCA) for RSF2 Stage 1 have been undertaken in accordance with the ANCOLD Guidelines on the Consequence Categories for Dams (ANCOLD, 2012) and ANCOLD Guidelines on Tailings Dams (ANCOLD, 2019a). The consequence category assessment is used to categorise dams based on the consequences of a potential dam break which sets the basis for design criterion based on risks and determines dam safety management requirements.

The previous consequence category assessment was based on the high-level dam break consequence assessment undertaken during the RSF2 Concept Design. The Population and Risk (PAR) was assessed to be <1 for both Sunny Day and Flood Failure.

The highest severity of damage and loss due to the dam failure was assessed as being 'Major' based on the impact on the business and the natural environment due to the release of tailings and contaminated water. Based on a damage and loss level of 'Major' and a PAR <1, the DFCC was determined to be 'Significant' for both Sunny Day and Flood failure scenarios.

The ESCC was conservatively assessed to be 'Significant' due to the release of contaminated decant water stored in the RSF 2 that would have a 'Major' impact on the environment with PAR<1.

3.2 Dam Break Modelling

3.2.1 Failure Modes

The failure modes considered were piping failure through the embankment causing a breach of the dam down to the foundation elevation which are shown in Figure 3.1. These locations are as follows:

- Cell 1 West wall,
- Cell 1 South wall, and;
- Cell 2 East wall.

The level of ponded water in RSF2 Stage 1, Cell 1 and Cell 2 at the time of failure has been adopted as follows:

- Sunny day Failure (SDF): Pool elevation at normal operating level (RL 268.3 m AHD) with piping failure leading to a breach of the embankment.
- Flood Failure (FF) condition, which includes representation of different storm event: The pool elevation at each flood level with piping failure leading to a breach of the embankment are listed below:
- 1:100 AEP: RL 268.9 m AHD.
- 1:10,000 AEP: RL 269.4 m AHD.
- PMP-DF: RL 269.6 m AHD.

3.2.2 Breach Locations and Scenarios

Three breach locations were selected to assess the potential differences to downstream impacts based on height and volume of tailings and pondwater released as shown in Figure 3.1.

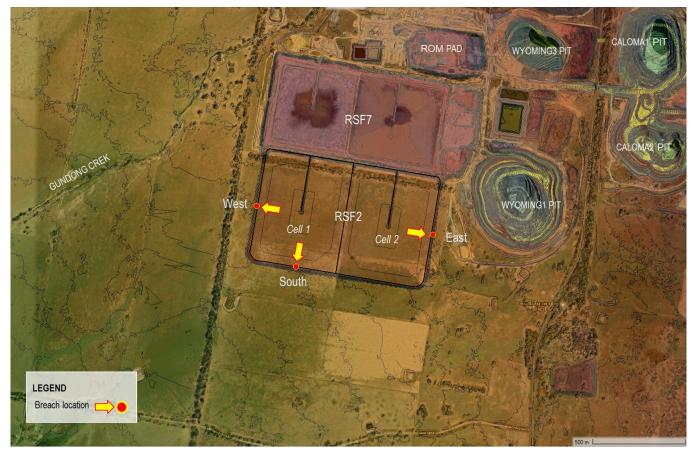


Figure 3.1 RSF2 Breach Locations

The scenarios considered in the analysis are summarized in Table 3.1.

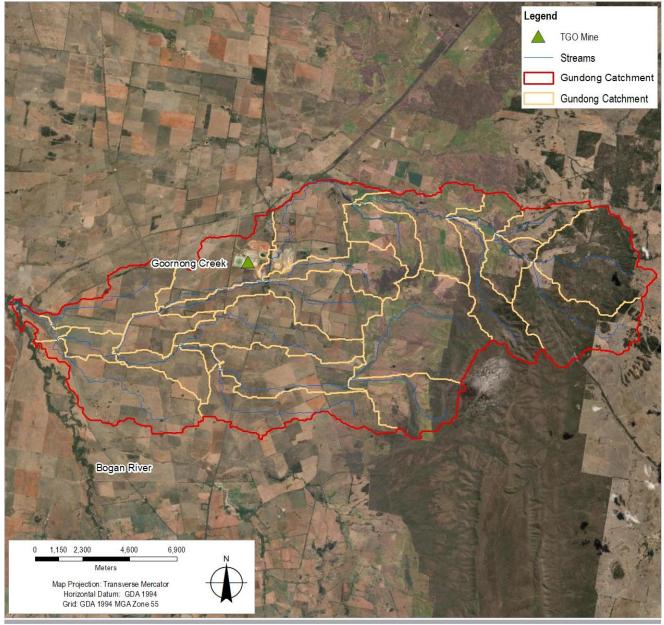
Scenario	Breach location	Failure event/mode	Pool water level at failure (m AHD)	Maximum tailings level (m AHD)
1	Cell 2 East Wall	SDF, Piping	268.3	269.3
2	Cell 2 East Wall	FF – 1:100 AEP, Piping	268.9	
3	Cell 2 East Wall	FF – 1:10,000 AEP, Piping	269.4	
4	Cell 2 East Wall	FF – PMP-DF, Piping	269.6	
5	Cell 1 South Wall	SDF, Piping	268.3	
6	Cell 1 South Wall	FF – 1:100 AEP, Piping	268.9	
7	Cell 1 South Wall	FF – 1:10,000 AEP, Piping	269.4	
8	Cell 1 South Wall	FF – PMP-DF, Piping	269.6	
9	Cell 1 West Wall	SDF, Piping	268.3	
10	Cell 1 West Wall	FF – 1:100 AEP, Piping	268.9	
11	Cell 1 West Wall	FF – 1:10,000 AEP, Piping	269.4	
12	Cell 1 West Wall	FF – PMP-DF, Piping	269.6	

Table 3.1	Dam Break Assessment Scenarios
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3.2.3 Catchment Hydrology of Receiving Environment

Hydrologic modelling was undertaken to determine required design event hydrographs using the XP-RAFTS model developed as part of the TGO RSF Stage 7 Dam Break Study (GHD, 2021). A brief overview of the hydrologic model is provided below. For a more detailed description of the model used, refer to the 2021 RSF Stage 7 Dam Break Study.

Figure 3.2 presents the Gundong Creek catchment boundary and its sub-catchments. The hydrologic model was intended to determine the concurrent flooding within the mine site and its downstream receiving environment. The antecedent flood conditions considered are 1 in 100 AEP, 1 in 10,000 AEP and PMPDF for the Rainy Day Failure scenarios. No hydrologic modelling was considered for the Sunny Day Failure scenarios.



Data source: Source: Esri, Maxar, GeoEye, Earthstar Geographics, CNES/Airbus DS, USDA, USGS, AeroGRID, IGN, and the GIS User Community Created by ccsalang

Figure 3.2 Gundong Creek Sub-Catchments

3.2.4 Dam breach estimation

3.2.4.1 Release Volume

The proportion of tailings released was estimated based on the post-liquefied shear strength of the tailings and assessment of the breach width due to various failure modes and correlated against empirical equations.

The three-dimensional (3D) software package Civil3D was used to estimate the tailings failure discharge volume based on average beach slope for the critical static liquefaction failure modes. The volume estimation in the 3D model is based on a "cone" formed by static gravitation forces at the angle of repose for saturated tailings material.

With the breach slope of 1.72 degrees and empirical equation-derived breach width, failure cones were modelled in Civil3D for estimation of the tailings volume in a runout event. These slopes/cones were projected from the external toe of the starter dam using recent LiDAR data provided by TGO. The 3D surface model for different breach locations is illustrated in Figure 3.3.

Water and tailings have been assumed to mix in a single volumetric release for the hydrographs modelled. The summary of the discharge volume estimated is given in Table 3.2.

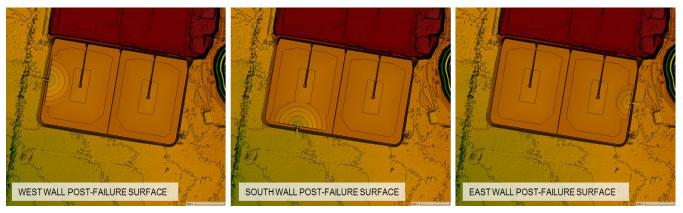


Figure 3.3 Post-Failure Surface Model of the Three Breach Locations

Table 3.2 Estimated Breach Discharge Volume

Scenario	Breach Iocation	Failure event/mode	Discharge Water Volume (x10 ⁶ m ³)	Discharge Tailings Volume (x10 ⁶ m ³)	Total Breach Volume (x10 ⁶ m ³)	Sediment Concentrati on by Volume, Cv
1	Cell 2 East Wall	SDF, Piping	0.015	0.029	0.044	33%
2	Cell 2 East Wall	FF – 1:100 AEP, Piping	0.122	0.029	0.151	10%
3	Cell 2 East Wall	FF – 1:10,000 AEP, Piping	0.336	0.029	0.365	4%
4	Cell 2 East Wall	FF – PMP-DF, Piping	0.438	0.029	0.467	3%
5	Cell 1 South Wall	SDF, Piping	0.015	0.113	0.128	44%
6	Cell 1 South Wall	FF – 1:100 AEP, Piping	0.122	0.113	0.235	24%
7	Cell 1 South Wall	FF – 1:10,000 AEP, Piping	0.337	0.113	0.449	13%
8	Cell 1 South Wall	FF – PMP-DF, Piping	0.438	0.113	0.551	10%

Scenario	Breach location	Failure event/mode	Discharge Water Volume (x10 ⁶ m ³)	Discharge Tailings Volume (x10 ⁶ m ³)	Total Breach Volume (x10 ⁶ m ³)	Sediment Concentrati on by Volume, Cv
9	Cell 1 West Wall	SDF, Piping	0.015	0.079	0.094	42%
10	Cell 1 West Wall	FF – 1:100 AEP, Piping	0.122	0.079	0.202	20%
11	Cell 1 West Wall	FF – 1:10,000 AEP, Piping	0.337	0.079	0.416	10%
12	Cell 1 West Wall	FF – PMP-DF, Piping	0.438	0.079	0.517	8%

All references within this report to sediment concentration or the terms: water flow, mud flood, mud flow and landslide have their understanding based on the CDA's recent technical bulletin on tailing dam breach analysis (CDA, 2019). In this publication the authors refer to the different states that tailings leave the TSF upon failure. Refer to Figure 3.4 for these flow types as a function of solids concentration, including graphical examples of each.

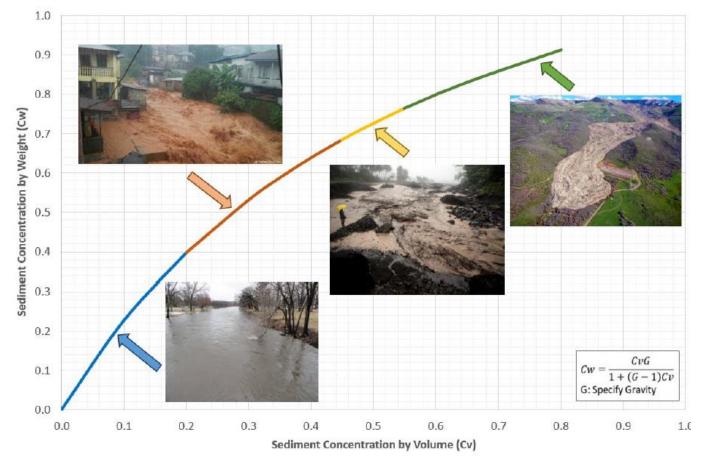


Figure 3.4 Flow Types as a Function of Solids Concentration (CDA, 2019)

Analysis of the geotechnical data, specifically the dry density and specific gravity of the tailings (data provided by TGO), suggested that the in-situ sediment concentration of the TSF varied between around 50% percent, without the influence of an elevated decant pond and 33% to 44% considering the amount of decant water for the SDF scenarios. This sediment concentration is typically representative of a mud flow case. However, for the FF scenarios, the volume of water run-off in the RSF2 is relatively high, resulting in a lower sediment concentration, Cv that ranges from 3% to 24% which can be considered as Newtonian fluid based. Therefore, a mudflow case is considered a highly credible scenario for the SDF scenarios.

3.2.4.2 Dam Breach Parameters

The breach parameters were estimated adopting the average results from five different empirical equations such as MacDonald et al, Froelich (1995), Froelich (2008), Von Thun & Gillete and Xu & Zhang. A summary of the breach parameters derived for all scenarios is provided in Table 3.3.

Scenario	Breach location	Failure event/mode	Pool water level at failure (m AHD)	Estimated Breach Volume (x10 ⁶ m ³)	Breach Formation Time (hours)	Breach bottom width (m)	Side Slopes (1:X)
1	Cell 2 East Wall	SDF, Piping	268.3	0.044	0.30	5.20	0.67
2	Cell 2 East Wall	FF – 1:100 AEP, Piping	268.9	0.151	0.43	7.60	0.67
3	Cell 2 East Wall	FF – 1:10,000 AEP, Piping	269.4	0.365	0.57	10.40	0.67
4	Cell 2 East Wall	FF – PMP- DF, Piping	269.6	0.467	0.62	11.60	0.67
5	Cell 1 South Wall	SDF, Piping	268.3	0.128	0.34	7.00	0.66
6	Cell 1 South Wall	FF – 1:100 AEP, Piping	268.9	0.235	0.40	8.60	0.66
7	Cell 1 South Wall	FF – 1:10,000 AEP, Piping	269.4	0.449	0.49	11.20	0.66
8	Cell 1 South Wall	FF – PMP- DF, Piping	269.6	0.551	0.52	12.20	0.66
9	Cell 1 West Wall	SDF, Piping	268.3	0.094	0.32	6.40	0.66
10	Cell 1 West Wall	FF – 1:100 AEP, Piping	268.9	0.202	0.42	8.20	0.67
11	Cell 1 West Wall	FF – 1:10,000 AEP, Piping	269.4	0.416	0.50	11.00	0.67
12	Cell 1 West Wall	FF – PMP- DF, Piping	269.6	0.517	0.54	11.80	0.67

Table 3.3 Summary of Breach Parameters

3.2.4.3 Breach Results

Breach hydrographs were estimated in HEC-RAS which provided the parameters presented in Figure 3.5, Figure 3.6 and Figure 3.7 while Table 3.4 summarizes the breach peak discharge and volume.

Scenario	Breach location	Failure event/mode	Peak breach discharge (m³/s)	Breach volume (m ³)
1	Cell 2 East Wall	SDF, Piping	40.46	43,877
2	Cell 2 East Wall	FF – 1:100 AEP, Piping	72.53	150,980
3	Cell 2 East Wall	FF – 1:10,000 AEP, Piping	108.74	365,187
4	Cell 2 East Wall	FF – PMP-DF, Piping	126.37	466,653
5	Cell 1 South Wall	SDF, Piping	130.5	127,582
6	Cell 1 South Wall	FF – 1:100 AEP, Piping	169.73	234,776

 Table 3.4
 Summary of Breach Flows

Scenario	Breach location	Failure event/mode	Peak breach discharge (m³/s)	Breach volume (m ³)
7	Cell 1 South Wall	FF – 1:10,000 AEP, Piping	231.55	449,164
8	Cell 1 South Wall	FF – PMP-DF, Piping	251.98	550,717
9	Cell 1 West Wall	SDF, Piping	92.54	94,308
10	Cell 1 West Wall	FF – 1:100 AEP, Piping	127.31	201,502
11	Cell 1 West Wall	FF – 1:10,000 AEP, Piping	187.5	415,890
12	Cell 1 West Wall	FF – PMP-DF, Piping	208.16	517,443

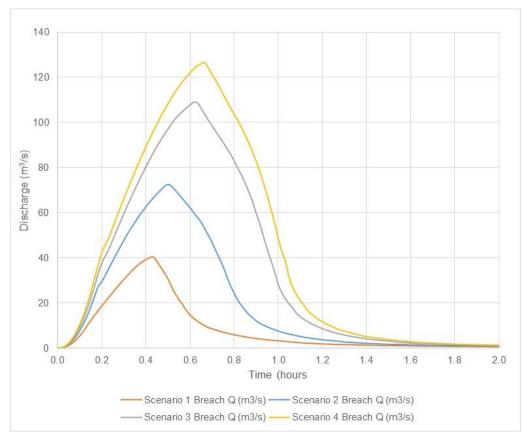


Figure 3.5 East Wall Breach Hydrographs

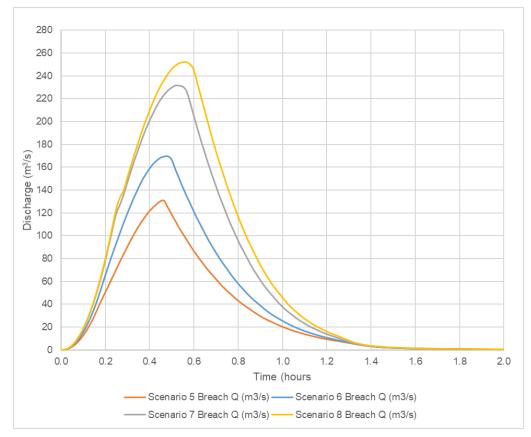


Figure 3.6 South Wall Breach Hydrographs

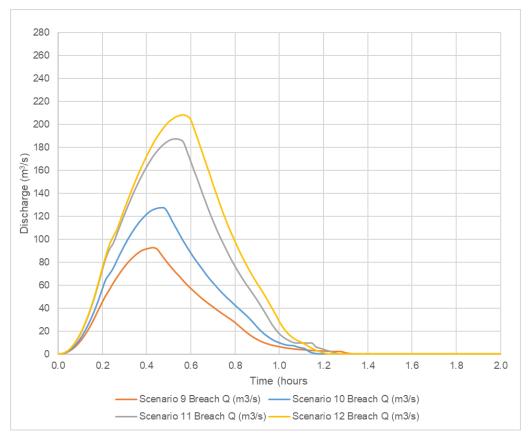


Figure 3.7 West Wall hydrographs

3.2.5 Hydraulic modelling

3.2.5.1 Modelling Objectives

The purpose of the dam hydraulic (flood routing) modelling was to estimate flood levels and extents resulting from a possible failure of RSF2 Stage 1. The modelling simulated a number of breach scenarios with varying model parameters, breach conditions, and breach locations. The modelled scenarios are described in detail in the previous sub-sections.

3.2.5.2 Modelling Software

The same HEC-RAS models used to estimate the breach hydrographs were utilized in routing the flood to the receiving environment. HEC-RAS capability of simulation mudflows were used for the SDF scenarios, and the FF scenarios were simulated as Newtonian fluid based on the assessment of the tailings and water volume and densities.

3.2.5.3 Model Extent

The two-dimensional hydraulic model extends from TGO mine site down to the confluence of Gundong Creek and Bogan River as presented in Figure 3.8. The model consists of over 149,000 composite cells ranging from 2mx2m to 20x20m cell sizes. The LiDAR supplied by TGO and the downloaded 5-m resolution DEM from ELVIS were combined to build the surface terrain for the hydraulic model.

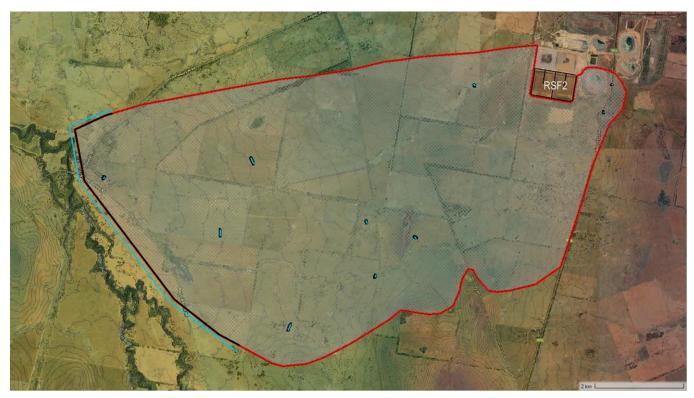


Figure 3.8 Hydraulic Model Extent

Roughness

The HEC-RAS model uses Manning's 'n' values to represent hydraulic resistance (notionally channel or floodplain roughness). Discrete regions of continuous vegetation types and land uses were mapped, and an appropriate roughness value assigned to each region.

The model domain covers the mining plant extending to the agricultural plains past the Tomingley West Road up to the Bulgandramine Road just before Bogan River. Visual inspection of available aerial imagery (Google Earth, ESRI Basemap) showed that the land within the model domain is mostly agricultural plains. In lieu of any other

information, the main channel was assumed to be clean with minimal debris. Therefore, the Manning's roughness coefficient was set to 0.035 all throughout the model domain.

3.2.5.4 Boundary Conditions

Inflow hydrographs developed from the XP-RAFTS model were input at the upstream end of each waterway. These were the inflow hydrographs for the 1:100 AEP, 1:10,000 AEP and PMPDF design flood events.

The "normal flow" outflow boundaries were assigned where water slope (based on the model topography).

3.2.6 Modelling results

The full set of inundation mapping flood depths and velocities is provided in Appendix B. Based on the hydraulic model, among the three Flood Failure events, the 1:100 AEP Flood Failure has the larger incremental flood extent, thus 1:100 AEP FF event is considered to be critical among the three FF scenarios. It should be note that for all modelled scenarios, no dwellings experienced flooding directly from the dambreak flood. Also, there is no flood impact observed in Wyoming1 Pit which is approximately 200m from the East wall.

The summary of flood depth, velocity and flood run-out distance from each breach location is tabulated in Table 3.5.

Scenario	Breach location	Failure event/mode	Depth (m)	Velocity (m/s)	Run-out distance (m)
1	Cell 2 East Wall	SDF, Piping	0.5 - 1.0	<1.0	900
2	Cell 2 East Wall	FF – 1in100 AEP, Piping	0.3 – 1.0*	<1.0	700*
3	Cell 2 East Wall	FF – 1:10,000 AEP, Piping	0.2 - 0.6*	<1.3	340*
4	Cell 2 East Wall	FF – PMP-DF, Piping	0.1 – 0.3*	<1.1	160*
5	Cell 1 South Wall	SDF, Piping	0.5 – 2.0	0.5 – 1.5	950
6	Cell 1 South Wall	FF – 1in100 AEP, Piping	0.3 – 1.0*	0.5 – 2.0	1,800*
7	Cell 1 South Wall	FF – 1:10,000 AEP, Piping	0.2 - 0.8*	<2.6	1,750*
8	Cell 1 South Wall	FF – PMP-DF, Piping	0.1 – 0.7*	<2.8	830*
9	Cell 1 West Wall	SDF, Piping	0.5 – 2.0	0.5 - 1.5	1,200
10	Cell 1 West Wall	FF – 1in100 AEP, Piping	0.5 – 1.0*	0.5 – 2.0	760*
11	Cell 1 West Wall	FF – 1:10,000 AEP, Piping	0.2 - 0.8*	0.5 – 2.6	1,500*
12	Cell 1 West Wall	FF – PMP-DF, Piping	0.1 – 0.1*	0.5 – 2.8	1,000*

 Table 3.5
 Summary of Depth, Velocity, and Run-Out Distance

*based on incremental depth

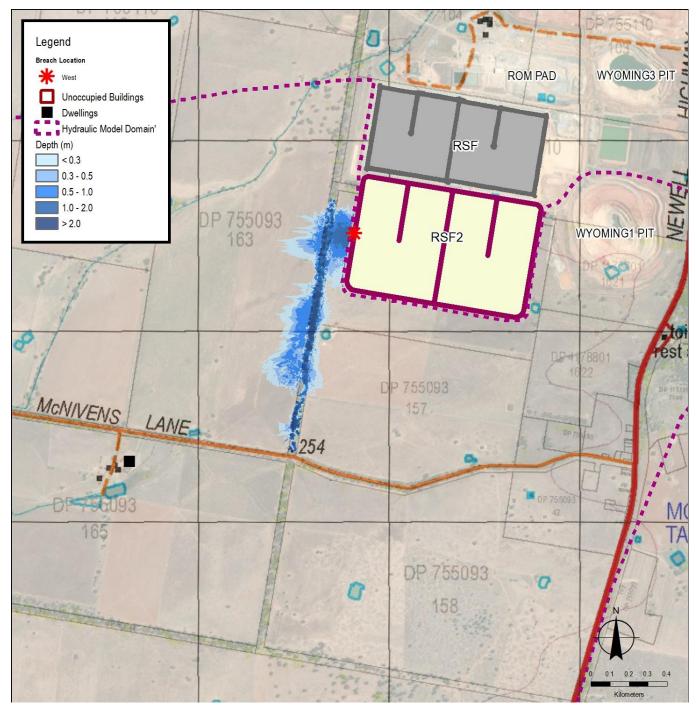


Figure 3.9 West Wall SDF Flood Extent

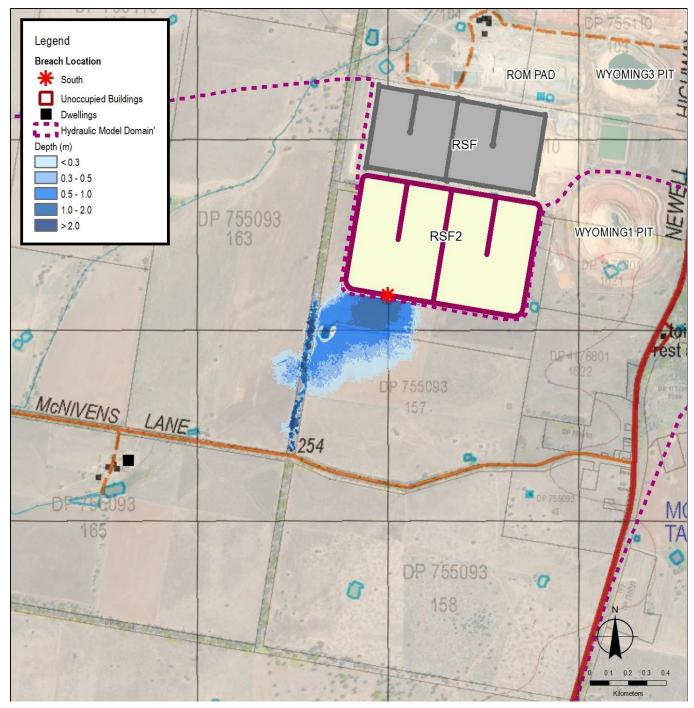


Figure 3.10 South Wall SDF Flood Extent

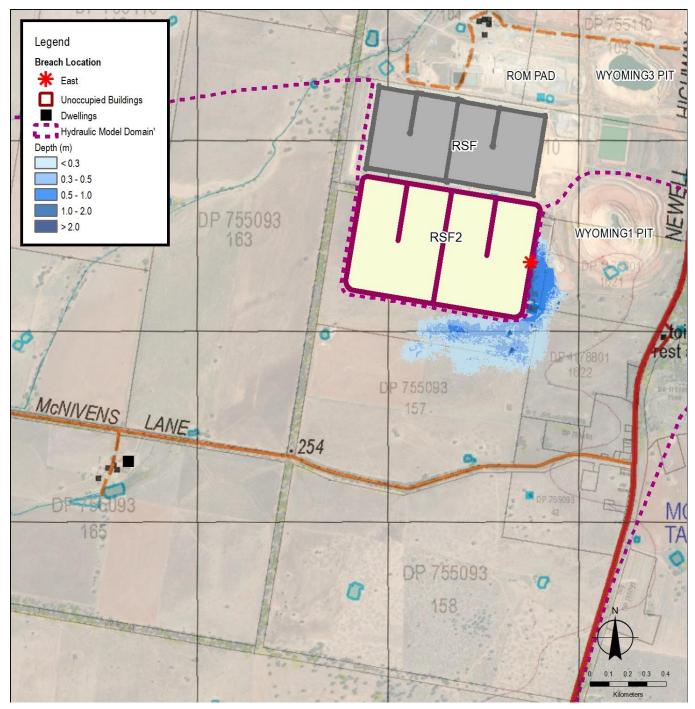


Figure 3.11 East Wall SDF Flood Extent

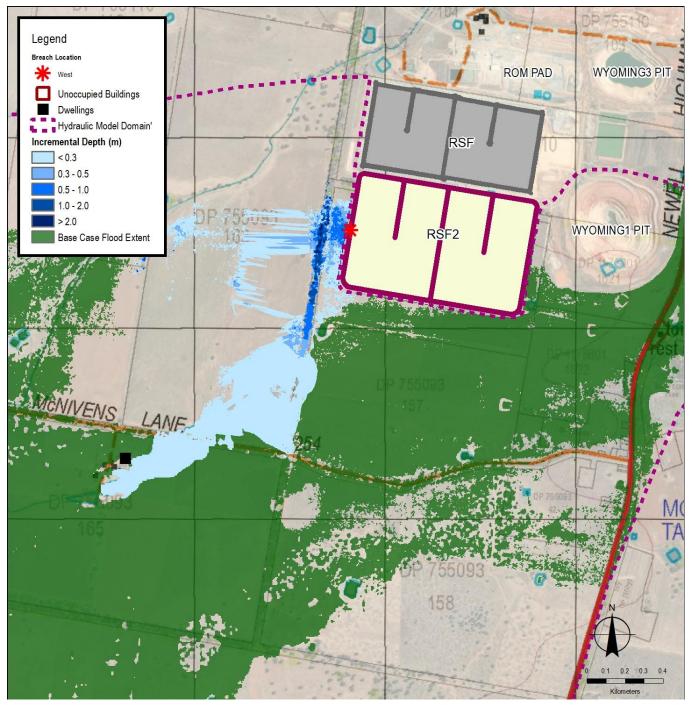


Figure 3.12 West Wall 1:10

West Wall 1:100 AEP FF Flood Extent

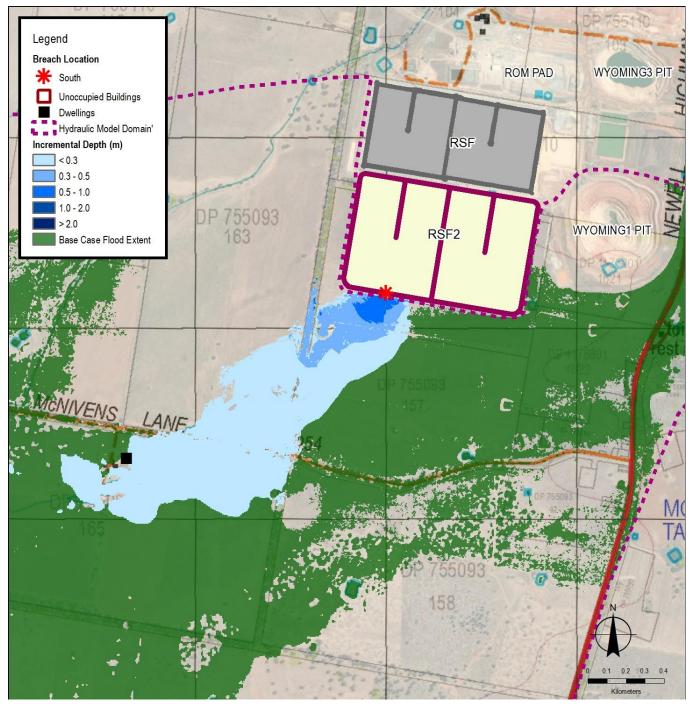


Figure 3.13 South Wall 1:100 AEP FF Flood Extent

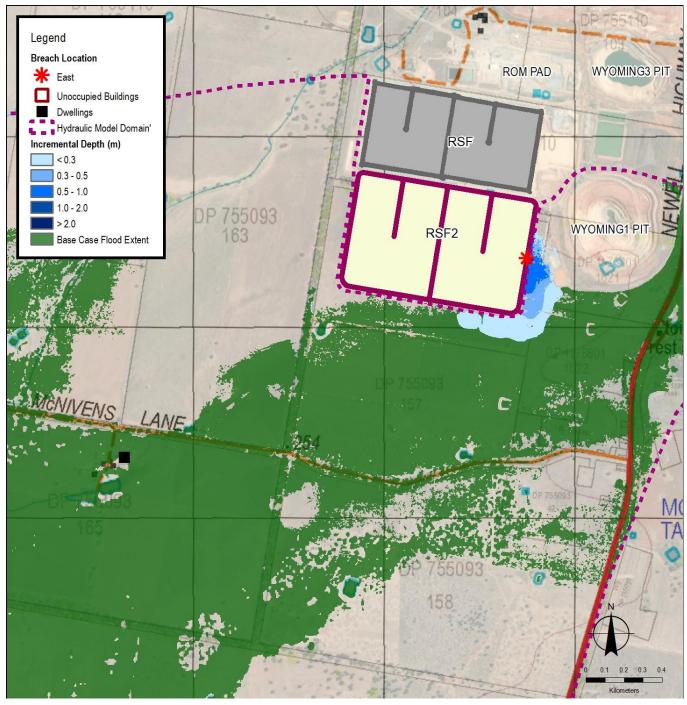


Figure 3.14 East Wall 1:100 AEP FF Flood Extent

3.3 Dam Failure Consequence Category (DFCC)

The Tomingley Gold Mine is surrounded by farming land which generally falls to the southwest. Both RSF1 and RSF2 are located to the south of the plant site.

The underground mining is currently operating from a portal within the Wyoming 1 pit which is to the east of the RSF2 Cell 2. It is estimated that there will be 1 Mm³ of tailings in each cell when they are filled to Stage 1/Stage 2 design level.

The highest severity of damage and loss due to the dam failure remains as being 'Major' based on the impact on the business and the natural environment due to the release of tailings and contaminated water. The population at risk (PAR) is estimated below based on a high-level dam break assessment.

3.3.1 Sunny Day Failure

The most probable failure mode on a sunny day for RSF2 is piping through the embankment which would require water against the embankment wall.

Base on the simulations for the three breach locations, the Population at Risk (PAR) has been assessed to be <1 resulting in a Consequence Category of '**Significant**' based on '**Major**' impacts to the business.

3.3.2 Flood Failure

The flood failure scenario would provide adequate warning time for any site personnel to be evacuated in the event of a catastrophic failure as the plant and underground would likely be stopped and evacuated due to the flood.

Base on the simulations for the three breach locations and three rare and extreme storm events such as 1:100 AEP, 1:10,000 AEP and PMPDF, the population at Risk (PAR) has been assessed to be <1.

The critical flood failure event would consist of poor operational management at the site, i.e. the storage being overfilled with tailings and a large ponded area combined with a 1:100 AEP event occurring causing overtopping. However, this can be avoided with good operational management, including minimizing the pond level, not filling beyond the design limits, daily inspections by trained personnel, and consistent monitoring.

The DFCC for Flood Failure is considered to be 'Significant'.

3.4 Environmental Spill Consequence Category (ESCC)

The Environmental Spill Consequence Category has been reviewed for the RSF2 Stage 1 and Stage 2 dam and assessed to be '**Low**' due to the release of decant water stored in the RSF that would have a "**Medium**" impact on the environment with PAR<1. The main contributing factors to the ESCC are related to the potential contamination of water supplies used by stock and fauna with no anticipated health impacts and the short-term impacts on the local ecosystems.

3.5 Summary of CCA

GHD has reviewed the consequence category of the RSF2 Stage 1 as summarised in Table 3.6.

Assessment	The severity of Damage and Loss	PAR	Consequence Category
Dam Failure CC	Major	(SDF)<1	Significant
		(FF)<1	Significant
Environmental Spill CC	Medium	<1	Low

 Table 3.6
 Summary of Consequence Category Assessment

3.6 Implications of Consequence Category Assessment

The RSF 2 should be designed to meet the standards in accordance with ANCOLD guides for a '**Significant**' DFCC and '**Low**' ESCC, as outlined below.

3.6.1 Design Parameters

Key design parameters based on the Consequence Category Assessment are detailed below.

Design Flood and Freeboard

ANCOLD gives guidance on flood design parameters based on both the Dam Failure Consequence Category and the Environmental Spill Consequence Category of the dam. The design hydraulic performance parameters adopted based on ANCOLD Guidelines on Tailings Dams (ANCOLD, 2019) is as follows:

- Minimum wet season water storage allowance: 1:10 AEP wet season runoff.
- Minimum extreme storm storage: 1:1,000 AEP, 72 hour flood event¹.
- Contingency freeboard: 300 mm.
- Emergency spillway design capacity: 1:1,000 AEP flood event with sufficient freeboard for wave run-up during a 1:10 AEP wind event¹.

Note 1: Minimum extreme storm storage design event is only suggested in ANCOLD as 1:100 AEP 72 hour, however, given the importance of the structure to the business moving forward and the impacts of a failure through the external embankment this minimum storage requirement has been increased as outlined in Section 7.

Design Earthquake Loading

For a '**Significant**' Sunny Day Failure Consequence Category dam, ANCOLD (2019a) suggests a 1:475 AEP Operating Basis Earthquake (OBE) and 1:1,000 AEP Safety Evaluation Earthquake (SEE).

ANCOLD (2019a) highlights that if the structure is susceptible to liquefaction or has components that will fail at ground motions only a little greater than those recommended, to check the design for the critical ground motion and assess the adequacy of the design using risk assessment methods. Where liquefaction of tailings is likely to affect the stability of the dam, the design may need to consider a 1:10,000 AEP earthquake, unless an appropriate risk assessment is undertaken to confirm that an adequate risk profile is achieved with a lower earthquake loading. ANCOLD (2019b) also recommends 1:10,000 AEP Maximum Credible Earthquake (MCE) be used in the design of the closure arrangement.

3.6.2 Monitoring, Surveillance, and Reporting Requirements

ANCOLD's Guidelines on Dam Safety Management (ANCOLD, 2003) gives guidance on the frequency of dam monitoring and surveillance for different consequence category tailings dams. For a '**Significant**' Dam Failure Consequence Category dam, ANCOLD suggests the inspection and monitoring types and frequencies as shown in Table 3.7 and Table 3.8, respectively. The recommended surveillance and monitoring requirements are detailed in Section 15.

Inspection Type	ANCOLD Recommended Frequency
Routine Visual	Twice Weekly to Weekly
Intermediate	Annual to 2-Yearly
Comprehensive	On Commissioning then 5-Yearly
Special	As Required

Table 3.7 Inspection Types and Frequencies

Table 3.8 Monitoring Types and Frequencies

Monitoring Type	ANCOLD Recommended Frequency
Rainfall	Twice Weekly to Weekly
Storage Level	Twice Weekly to Weekly
Seepage	Twice Weekly to Weekly
Chemical Analysis of Seepage	Consider
Pore Pressure	3-Monthly to 6-Monthly
Surface Movement Control	Consider
Surface Movement Normal	Consider
Internal Movement / Stresses	Consider
Seismological	Consider

3.6.3 Operation Maintenance and Surveillance Manual

An Operations Maintenance and Surveillance (OMS) Manual has been prepared as part of the RSF2 Detailed Design. The principal objective of this manual is to provide a documented operation procedure to assist in the safe and efficient management of tailings and water around the material stockpile.

The OMS manual has been developed in accordance with minimum regulatory requirements (ANCOLD, 2003) and includes:

- Roles and responsibilities.
- Design intent.
- Regular operations and inspections.
- Water and tailings management procedures.
- Operational requirements for mechanical equipment and instrumentation.
- Maintenance schedules and procedures.
- Surveillance requirements.
- Examples of potential damages and associated repair works.
- Trigger Action Response Plans.

The OMS manual outlines key monitoring activities which will include:

- Routine monitoring of RSF.
- Routine monitoring of water levels and process plant return water rates.
- Routine monitoring of groundwater level fluctuations.
- Underdrainage system return rates and volume.

The OMS Manual has been provided to TGO as a separate deliverable.

3.6.4 Dam Safety Emergency Planning

ANCOLD (ANCOLD, 2003) states a Dam Safety Emergency Plan (DSEP), otherwise known as an Emergency Response Plan (ERP), be prepared where any persons, infrastructure, or environmental values could be at risk if the dam were to fail. An ERP is therefore required for both the construction and operation of the RSF2 impoundment. There is an existing DSEP for RSF1, which has been updated to include RSF2 also as a combined facility.

4. Geotechnical Data

4.1 Overview

An intrusive geotechnical ground investigation was carried out between 17 November and 5 December 2020 to confirm the foundation characteristics and geotechnical design parameters at the RSF2 location. The geotechnical investigation comprised of the following:

- A desktop review of publicly available information and previous geotechnical investigations nearby the proposed site.
- A program of geotechnical fieldwork, consisting of test pit investigations and borehole investigations.
- Laboratory tests on samples obtained from test pitting and boreholes.

The geotechnical investigation focused on foundation soil up to depths of 20 m. The geotechnical characteristics and shear strength parameters adopted for materials such as tailings or borrows have been adopted as similar to those characterized as part of RSF 1 Stage 8 (GHD 2021), as later summarized in Section 4.7.

4.2 Geotechnical Site Investigation

The geotechnical investigation comprised of the following activities:

- Mechanical excavation of 20 test pits (TP01 to TP20) across the proposed reservoir storage area. Test pits were excavated to a maximum depth of 4.0 m below ground level.
- Drilling of 10 boreholes (BH01 to BH10) within the alignment of the proposed embankment foundation. The boreholes were advanced using sonic drilling techniques from the ground surface to achieve a maximum depth of 20 m below ground level.

The location of these test pits and boreholes are shown in Figure 4.1.

During test pitting and drilling, the following in-situ testing was undertaken:

- Pocket penetrometer testing to assess the in-situ consistency of the shallow soils and aid in the identification of any potentially soft or low strength zones.
- Dynamic Cone Penetrometer (DCP) testing was undertaken at select locations to assist in determining the density of the near-surface material.
- Standard Penetration Tests (SPT) were undertaken within the boreholes at approximately 1.5 m intervals (where appropriate) to assess the consistency of the soils and to recover disturbed samples.



Figure 4.1 Test Pits and Boreholes Location

4.3 Laboratory Testing

Laboratory testing was undertaken on samples collected from the test pits and boreholes to confirm field logging and assess material properties. The following tests were undertaken according to the relevant Australian Standard test methods:

- Particle Size Distribution, including Hydrometer.
- Atterberg Limits.
- Field Moisture Content.
- Bulk Density.
- Emerson Class (dispersion).
- Standard Compaction Tests (98% MDD).
- Constant Head Permeability (Triaxial Cell).
- X-ray Computed Tomography (XCT) scan.
- Single Stage Triaxial Consolidated Undrained (CU) Strength Tests with Pore Pressure Measurement.
- One-dimensional consolidation (Oedometer) tests.

Testing was conducted in the NATA accredited Fugro AG Laboratory in Perth, and GHD's NATA accredited facility in Sydney.

4.4 Foundation

4.4.1 General

The area is dominated by alluvial sequences of clays, sands, and gravel of Quaternary to Tertiary age. The subsurface conditions encountered within the test pits and boreholes generally comprised of topsoil overlying alluvial (sandy) clays with varying proportions of gravels. The topsoil encountered typically comprises of low plasticity, sandy clay with rootlets. The alluvial material encountered comprises of very stiff to hard, sandy clay/clay with varying proportions of gravels. The clay was typical of low to medium plasticity with a field moisture condition generally equal to, or less than, the plastic limit (i.e. dry to moist).

4.4.2 PSD, Atterberg Limits and Permeability

The foundation material corresponds to clay to sandy clay material with traces of gravel. The fines content is highly variable and within a range of 13% to 95%. The PSDs for the samples recovered during the 2021 geotechnical investigation are shown in Figure 4.2. The fine fraction generally classifies as a low to intermediate plasticity clay (CL - CI) following the Casagrande Plasticity Chart, as shown in Figure 4.3. Standard compaction tests conducted in samples collected from test pits show maximum dry densities between 1.62 (t/m³).and 1.87 (t/m³). Optimum moisture content was found between 14.5% and 20% with field moisture contents typically between 10% to 17.4%.

The in-situ unit weight for the materials in the upper 10 m is approximately 1.40 t/m^3 to 1.85 t/m^3 . The specific gravity of the material is approximately 2.6. Oedometer tests conducted on samples collected from boreholes at depths greater than 3 m show Compression Index (C_c) in the range of 0.09 to 0.12.

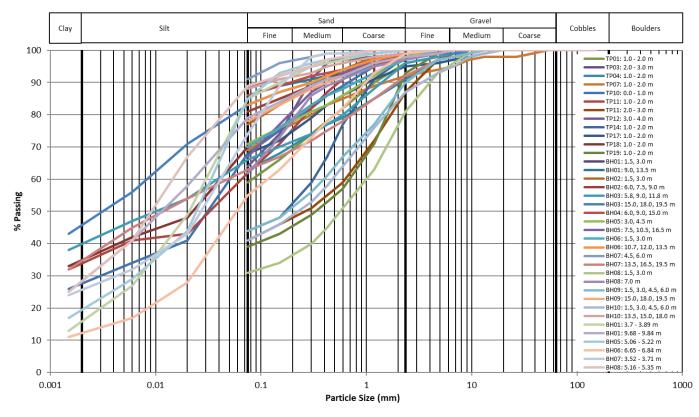


Figure 4.2 Particle Size Distribution Plot

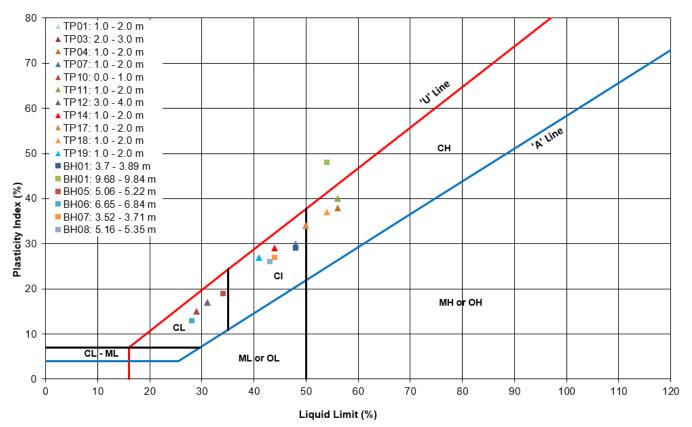


Figure 4.3 Atterberg Limits Plot

The clayey foundation material show, after Constant Head Permeability testing, a coefficient of permeability between 10⁻¹⁰ m/s to 10⁻¹¹ m/s, Table 4.1.

 Table 4.1
 Laboratory Permeability Testing

Test ID	Depth (m)	Description	Coefficient of permeability, k (m/sec)
TP03	2.0 - 3.0	CLAY trace sand	2 x 10 ⁻¹⁰
TP10	0.0 – 1.0	Sandy CLAY, trace gravel	2 x 10 ⁻¹⁰
TP11	1.0 – 2.0	Sandy CLAY	4 x 10 ⁻¹¹
TP18	1.0 – 2.0	Sandy CLAY, trace gravel	1 x 10 ⁻¹⁰

Tabulated summaries of the respective soil laboratory test results are presented in Table 4.2.

Table 4.2 Soil Classification Testing

Test ID	Depth (m)	Soil Description	MC	Plastic	city (%)	LS	F	Particle Size	Distribution	(%)	Emerson
Testid			(%)	LL	PI	(%)	Clay	Silt	Sand	Gravel	Class
TP01	1.0 - 2.0	Sandy CLAY, trace gravel	10.0	31	17	—		59	37	4	2
TP03	2.0 - 3.0	CLAY with sand, trace gravel	16.9	50	34	—	32	56	11	1	2
TP04	1.0 - 2.0	CLAY with sand, trace gravel	13.9	56	38	—	30	33	28	9	2
TP07	1.0 - 2.0	CLAY with sand, trace gravel	17.7	48	30	-		69	23	8	2
TP10	0.0 - 1.0	Sandy CLAY, trace gravel	11.5	29	15	_	28	40	31	1	2
TP11	1.0 - 2.0	Sandy CLAY	17.4	56	40	_	35	27	36	2	2
TP11	2.0 - 3.0	Sandy CLAY with gravel	-	-	-	—		41	46	13	-
TP12	3.0 - 4.0	Sandy CLAY	14.1	31	17	—		64	36	0	2
TP14	1.0 - 2.0	Sandy CLAY, trace gravel	14.1	44	29	—		44	54	2	2
TP17	1.0 - 2.0	Sandy CLAY, trace gravel	14.4	50	34	-		66	29	5	2
TP18	1.0 - 2.0	CLAY with sand, trace gravel	14.7	54	37	-	35	35	27	3	2
TP19	1.0 - 2.0	Sandy CLAY, trace gravel	8.1	41	27	—		39	54	7	-
BH01	1.5 and 3.0*	Sandy CLAY trace gravel	-	-	-	—		95	35	5	-
BH01	3.7 - 3.89	CLAY with sand	22.7	48	29	9.0	17	68	15	—	-
BH01	9.0 and 13.5*	CLAY, trace sand & gravel	-	-	-	—		86	12	2	-
BH01	9.68 - 9.84	CLAY with sand	17.0	54	48	10	31	48	19	2	
BH02	1.5 and 3.0*	CLAY with sand, trace gravel	-	-	-	_		77	22	1	-
BH02	6.0, 7.5, and 9.0*	CLAY with sand, trace gravel	-	-	-	—		81	17	2	-
BH03	5.8, 9.0, and 11.8*	Sandy CLAY, trace gravel	-	-	-	_	40	26	30	4	-
BH03	15.0, 18.0 and 19.5*	CLAY with sand, trace gravel	-	-	-	—	45	38	14	3	-
BH04	6.0, 9.0, and 15*	CLAY trace sand	-	-	-	_		86	13	1	-
BH05	3.0 and 4.5*	CLAY with sand, trace gravel	-	-	-	—		71	27	2	-
BH05	5.06 - 5.22	CLAY with sand	19.1	34	19	6	20	65	15	—	-
BH05	7.5, 10.0, and 16.5*	Sandy CLAY, trace gravel	-	-	_	_		67	32	1	—
BH06	1.5 and 3.0*	CLAY with sand, trace gravel	-	-	_	—		70	27	3	—
BH06	6.65 - 6.84	Sandy CLAY	12.2	28	13	6	13	42	44	_	—
BH06	10.7, 12.0, and 13.5*	CLAY with sand, trace gravel	-	-	_	_		83	16	1	—
BH07	3.52 - 3.71	CLAY with sand	17.5	44	27	10	26	48	26	_	—
BH07	4.5 and 6.0*	CLAY trace sand	-	-	_	_		91	9	0	—
BH07	13.5, 16.5, and 19.5*	CLAY with sand, trace gravel	-	-	_	_	34	29	29	_	_
BH08	1.5 and 3.0*	Clayey SAND with gravel	-	-	_	_		31	50	19	—
BH08	5.16 - 5.35	CLAY trace sand	20.1	43	26	11.5	30	59	11	—	—
BH08	7.5 - 7.95	Sandy CLAY trace gravel	-	-	_	_		62	35	3	—
BH09	1.5, 3.0, 4.5, and 6.0*	Sandy CLAY trace gravel	-	-	_	_		44	46	10	—
BH09	15.0,18.0, and 19.5*	CLAY with sand, trace gravel	-	-	_	_		78	20	2	—
BH10	1.5, 3.0, 6.0, and 9.0*	Sandy CLAY trace gravel	-	-	-	—		41	46	13	-

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BH10	13.5,15.0, and 18.0*	CLAY trace sand and gravel	—	_	_	_	88	11	1	_
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Note: MC = Moisture Content; LL = Liquid Limit; PI = Plasticity Index; LS = Linear Shrinkage, * Multiple depths indicate blended samples

4.4.3 Shear Strength

Four multi-stage and three single-stage isotropically consolidated triaxial (CIU) tests with measurement of pore water pressure, have been completed on undisturbed samples. Samples were tested with the consideration of the future consolidation pressure. The test results are summarised in Table 4.3. Effective shear strength parameters show a mean friction angle of 34° and zero-effective cohesion. Similarly, the undrained shear strength ratio (s_u/p_o) is generally greater than 0.6.

Sample Locations	Consolidation Pressure, po' (kPa)	Maximum Stress Ratio (M=q/p')	Effective Peak Friction Angle, Φ (°)	Peak Undrained Shear Strength, S _u (kPa)	Shear Strength Ratio, s _u /po'
D U 04	130	1.61	39.40	126	0.97
BH01 9.68 – 9.84 m	330	1.24	30.92	234	0.71
5.00 5.04 m	530	1.09	27.47	318	0.6
	120	1.39	34.35	102	0.85
BH05 5.06 – 5.22 m	300	1.21	30.23	180	0.6
5.00 – 5.22 m	480	1.15	28.85	269	0.56
	120	1.51	37.10	119	0.99
BH06 6.65 – 6.84 m	300	1.38	34.12	240	0.8
0.03 – 0.04 m	480	1.33	32.98	379	0.79
	120	1.66	40.55	134	1.12
BH08 5.16 – 5.35 m	300	1.33	32.98	228	0.76
3.10 – 3.33 m	480	1.18	29.54	302	0.63
BH01 3.51 – 3.70 m	100	1.5	36.87	119	1.19
BH01 3.70 – 3.86 m	200	1.39	34.35	160	0.8
BH07 3.52 – 3.71 m	400	1.23	30.69	240	0.6

Table 4.3 Triaxial Test Results

These CIU results were also compared against the outcomes of the geotechnical investigations completed as part of the RSF1 detailed design for Stage 7 and Stage 8. The investigations were undertaken during 2017 and 2020 geotechnical investigation programs and explored the undrained shear strength of the foundation soil. The peak undrained shear strength parameters were obtained from single-stage triaxial testing and the final stage of the multistage triaxial tests. The shear strength was then reduced to account for direct simple shear conditions (intermediate strength between triaxial compression and extension) which is expected along the failure surfaces of both RSF1 and RSF2.

Figure 4.4 and Figure 4.5 show the undrained shear strength functions for both peak and the residual conditions for both the RSF2 and RSF1 data. The ultimate residual strength of the foundation was generally not attained even at 15% axial strain. As such, the response at 15% was adopted in the analysis (this strain well exceeds RSF serviceability criteria). The resulting S_{u}/σ'_{vo} functions were adjusted using a polynomial function considering the minimum values between drained and undrained shear strength.

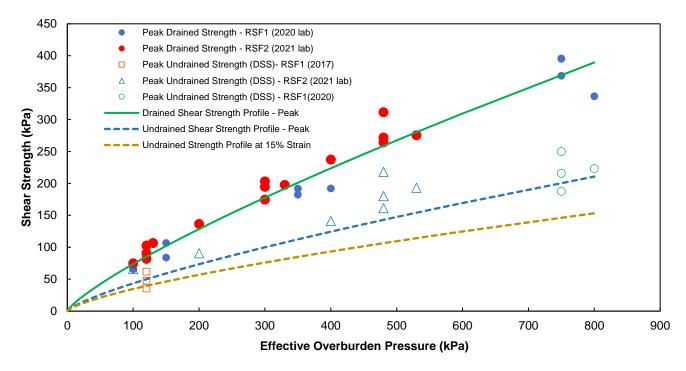


Figure 4.4 Peak Undrained Shear Strength.

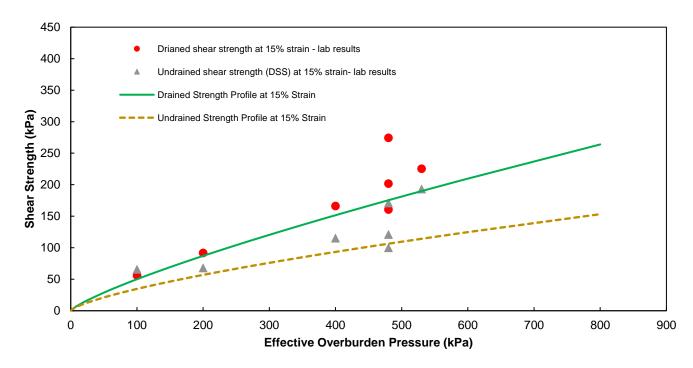


Figure 4.5 Peak Undrained Shear Strength.

4.4.4 SPTs

Standard Penetration Tests (SPT) were undertaken within the boreholes at approximately 1.5 m intervals (where appropriate) to assess the consistency of the soils and to recover disturbed samples. The uncorrected results of the SPTs are plotted in Figure 4.6. The N value increases monotonically with depth with a minimum value of N \approx 16 at z \approx 6 m. Refusal was observed at different depths. The corrected N value is later used to evaluate the liquefaction susceptibility of the foundation materials.

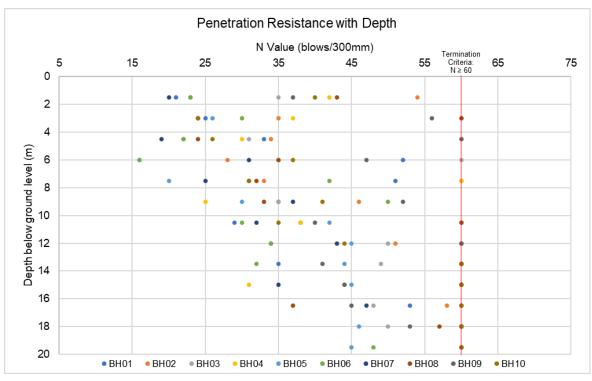


Figure 4.6 Relationship between SPT N value and depth

4.5 Tailings

4.5.1 General

A thorough characterisation of the physical properties of the tailings materials can be found in Section 13.1. The characterisation considered 5 CPTu holes taken through the Stage 2/3 embankments, and 4 CPTu holes on the tailings beaches.

The CPTu - SBT charts show that the upper section of the RSF1 tailings (from surface to 5 m deep) are likely to respond similarly to sand-like material in accordance with Jefferies and Davies (1993). At greater depths, the tailings are highly interbedded with clay-like and very thin layers of sand-like material. Interlayering and segregation in the RSF is likely a result of variable settlement rates of tailings, environmental effects, over-consolidation of material, and changes in ore characteristics/processing over time.

4.5.2 Index properties

The particle size distributions (PSDs) for the tailings show that the fines content for the "clay-like" and "sand-like" tailings range between 60% to 85%, and between 40% to 55%, respectively. The material mostly classifies as low plasticity silty (ML), according to the Unified Soil Classification System (USCS). The fine portion (finer than 0.075 mm) of the "clay-like" and "sand-like" material are classified, according to Casagrande Plasticity Chart, as low to intermediate plasticity Clay (CL-CI) and low plasticity silt (ML), respectively. The plasticity index of the tailings is generally below 10% with an average liquid limit of 28%. The specific gravity (G_s) of the material was found to be equal to 2.55.

The tailings unit weight is variable within RSF1. Based on the 2018 density testing on the beach surface, the unit weight of the tailings has been estimated to be 1.50 t/m³ with an in-situ moisture content between 8% and 18%. These results give a saturated unit weight of 1.95 t/m³ and a bulk unit weight of 1.8 t/m³ for tailings at shallow depths.

4.5.3 Shear strength

The sand-like tailings, which are conservatively adopted as representative for materials in the RSF2, show CPTU tip resistance between 1 MPa and 3 MPa in the RSF1. The Soil Behaviour Type (Ic) value is typically between 2 and 2.6, and CPTu excess pore pressure ratio (B_q) is between 0 and 0.1.

The shear strength parameters were defined from CPTs and consolidated drained and undrained triaxial tests undertaken on re-constituted RSF1 tailings samples (GHD, 2021). The inspection of CPTu-SBTn charts presented shows that the "sand-like" portion is near the dilation-contraction contour and deposited in a loose state. As such, the undrained shear strength is expected to be significantly lower than the drained shear strength. As a reference, this type of tailings shows an effective friction angle generally greater than 25 degrees and yields a strength versus effective overburden ratio of no less than 0.46.

A $S_{u,peak}/\sigma'_{vo}\approx 0.20$ for tailings was inferred from the CPTu tests adopting the first quartile for S_u values as a reasonable criterion for LE analysis. Similarly, the residual shear strength ratios ($S_{u,r}/\sigma'_{vo}$) were obtained using the Critical State Line (CSL) and the state parameter ψ (defined as the difference in void ratios between the in-situ state and the Critical State under the same effective confining stress) for the 'sand-like' tailings. The in-situ state parameters (ψ_0) were obtained from 2020 CPTu tests and the expression proposed by Jefferies and Been (2016) which relates ψ and normalised CPTu tip resistance as shown in Figure 4.7. The 'sand-like' tailings shown in Figure 4.7, show that the upper quartile of the data points is $\psi_{Q3} \ge -0.05$, and thus slightly dilative. Based on this, a residual undrained shear strength ratio of 0.2 is adopted for the tailings in the RSF1.

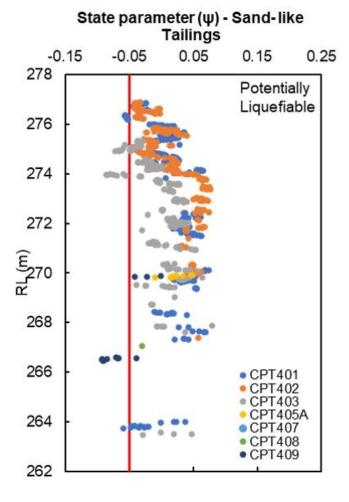


Figure 4.7 State Parameter for Tailings

4.6 Embankment Clay

4.6.1 General

The clayey material has been extensively used in the construction of the RSF1. This material will be also used in the RSF2. Previous as-constructed reports for Stages 6 and 7 (GHD, 2019; GHD, 2020) and previous geotechnical investigation provide an adequate characterisation of the material.

4.6.2 Index properties

The embankment material corresponds to a sandy silty clay or silty clay with sand. The fine portion of the material (finer than 0.075 mm) shows low to medium plasticity with PI between 13% and 33%. Fines content is generally greater than 60% with a coefficient of permeability of approximately 10⁻⁹ m/s. The material is generally compacted to a target density of 98% of Standard Maximum Dry Density (SMDD).

4.6.3 Shear strength

Two material samples were collected during both the Stage 3 and Stage 5 construction stages in 2016 and 2018, respectively. Multistage Consolidated Undrained (CU) triaxial tests were conducted on the Stage 3 embankment material at effective confining pressures of 50, 100, and 200 kPa. Single Stage CU triaxial tests were conducted on remoulded materials (prepared at 98% SMDD) on the Stage 5 materials at effective confining pressures of 40, 80, and 160 kPa.

For long-term loading conditions, the drained shear strength parameters have been obtained from the effective stress paths (p'-q) as described in GHD (2021) and shown in Figure 4.8. Considering the lower bound envelope, an effective friction angle of 30° and zero cohesion is found for the embankment material.

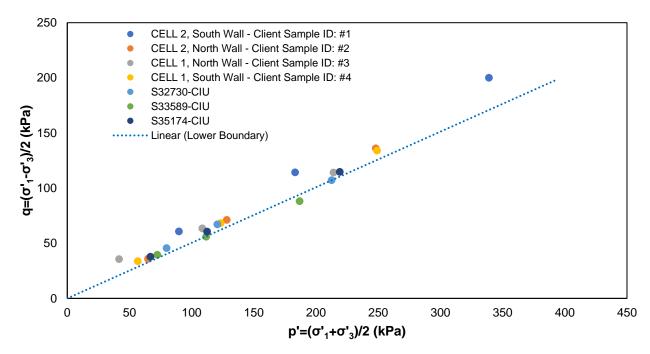


Figure 4.8 Drained Shear Strength – Embankment Clay

Similarly, the undrained shear strength ratio was obtained from CU tests. The resistance has been reduced by a factor of 0.57 to represent in situ direct simple shear (DSS) conditions. Figure 4.9 below shows the lower bound envelope (in the τ - σ'_{vo} plane) for the equivalent DSS shear strength, which yields $S_u/\sigma'_{vo}\approx$ 0.23 for the material.

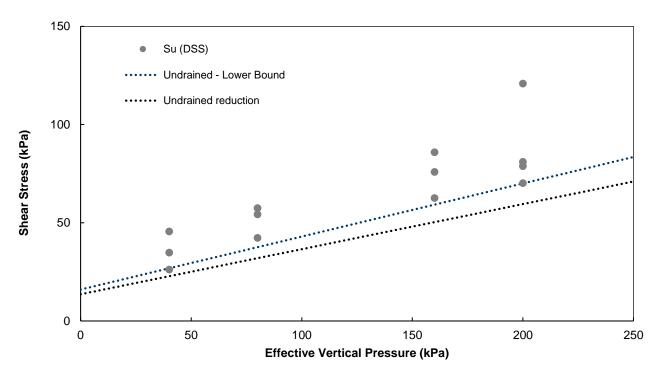


Figure 4.9 Undrained Shear Strength – Embankment clay

4.7 Adopted Material Strength Parameters

Material strength parameters have been developed based on both previous parameters and the results of the geotechnical investigation carried out in the RSF2 footprint. Table 4.4 summarises materials shear strength parameters adopted in the stability assessment of the facility. Liquefied shear strength parameters have been estimated from the post-liquefaction analysis as discussed in Section 13.4.

Material	Bulk Unit Weight	Drained Shear	Undrained Shear Strength	(kPa)
Description	(kN/m ³)	Strength (kPa)	Pre-seismic Peak	Post-seismic Strength
Sand-like Tailings	18	Refer Note 1	$S_u/\sigma'_v = 0.20$	$S_{u(liq)}/\sigma'_v = 0.038$
Embankment Clay	20	c' = 0 kPa ¢' = 30°	Normal/Shear function (see Figure 4.9)	Normal/Shear reduction function (see Figure 4.9)
Foundation	21	c' = 0 kPa ¢' = 26°	Normal/Shear function	Normal/Shear function
General Fill	20	c' = 0 kPa φ' = 35°	Same as the left	Same as the left
Zone 2	18	c' = 0 kPa φ' = 33°	Same as the left	Same as the left
Zone 3	22	c' = 0 kPa ¢' = 40°	Same as the left	Same as the left

 Table 4.4
 Adopted Material Strength Parameters

5. Tailings Deposition Options Assessment

5.1 Overview

During the preliminary design and leading into the detailed design of RSF2, a number of deposition options and RSF arrangements were considered for the RSF2 storage, in order to determine the best solution for the site. The following options were considered:

- Upstream Constructed RSF
- Downstream Constructed RSF
- Centreline Constructed RSF
- In-Pit Deposition RSF
- Dry Stacking Deposition RSF
- GeoTube RSF

Each of these options was assessed on a range of criteria including:

- Footprint impacts.
- CAPEX.
- OPEX.
- RSF Stability.
- Suitability for Closure.

This assessment was undertaken as a desktop assessment involving both GHD and TGO and incorporated readily available information and high-level conceptual arrangements for each of the options to inform the discussions.

5.2 Options Assessment

5.2.1 Upstream Constructed RSF

Upstream constructed TSF's have traditionally been constructed across the world as the preferred methodology for raising tailings facilities due to their low cost and efficient approach. However, recent developments within the industry, specifically large-scale failures of these types of facilities and the need for additional buttressing on the downstream face, have raised questions as to the suitability of this type of dam. At TGO, RSF1 has been constructed as an upstream raised facility with additional buttressing to meet stability requirements. However, when assessing the suitability of this option, the following was identified as key considerations:

- Low baseline option. However, there it is a high likelihood that costs may increase significantly, due to buttressing requirements.
- Reduced footprint area required under the base case scenarios which consists of purely upstream raises when compared with downstream raise arrangements. However, this is expected to increase with future buttressing requirements on site.
- Increased operational requirements to closely monitor and manage tailings deposition and pond extents such that the decant is far enough away from the external embankments to manage the risk of embankment instability.
- Reliance on non-engineered tailings materials of potentially variable consistency for the foundation/stability of the embankments, leading to significantly increased risk for the RSF.
- It is expected that significant work will be required for the stabilisation and encapsulation of the RSF prior to closure, including the effective compaction of tailings, capping, and development of long-term stable buttressing due to the risk of liquefaction of the tailings post closure when compared with alternative options.

Whilst this option was identified as the lowest potential cost option for the construction of a new embankment with no change to the process plant and minimal sustaining CAPEX required with each raise, the increased risk of instability both during operation and to satisfy closure requirements was not considered by TGO to be justifiable.

5.2.2 Downstream Constructed RSF

Downstream constructed RSF's are traditionally rarer than upstream constructed dams, largely due to the increased initial and ongoing capital costs. Because the embankment is constructed using the same geometry as a traditional water storage embankment, utilising natural foundations and is not dependent on the strength of the tailings for stability. The stability of these structures is thus inherently greater than for other embankment types. In particular, compared to upstream construction, which relies on the strength of the tailings for stability.

When assessing the suitability of this option, the following points were identified as key considerations:

- High baseline and ongoing cost option requiring significant capital input.
- Larger footprint area for the embankment than other options considered resulting in increased land clearance and preparation requirements.
- Potential for reduced operational requirements due to the greater stability of the structure, however the reduced operational requirements are considered to be marginal when considering the centreline option discussed below.
- Increased stability when compared with upstream raised tailings dams due to founding on the embankments (other than starter embankment) on natural ground rather than tailings.
- It is expected that a moderate amount of work (less than upstream for upstream) will be required for encapsulation of the RSF, prior to closure including the effective compaction of tailings and capping. However, buttressing is not anticipated, subject to confirmation by stability analysis of the embankment.

This option was identified as the highest potential cost option for the construction of the RSF. The stability and decreased operational risk benefits were seen to be outweighed by the magnitude of sustaining CAPEX and the footprint extents required for this arrangement. Accordingly, this was not considered to be a justifiable option for TGO, over say a Centreline Raise.

5.2.3 Centreline Constructed RSF

Following some recent failures of upstream constructed embankments, centreline constructed embankments are becoming more common in the industry due to increased safety of the structures. When compared with the downstream constructed RSF's, previous projects have shown up to a 50% reduction in embankment volume requirements resulting in reduced sustaining CAPEX costs. Additionally, because the centreline arrangement doesn't rely solely on the strength of the tailings for foundational strength this option is considered to be more stable than the upstream arrangement considered previously.

When assessing the suitability of this option, the following points were identified as key considerations:

- Moderate baseline and ongoing cost option requiring moderate capital input for the life of the facility.
- Similar initial footprint area to that of the upstream raise arrangement however smaller than downstream requirements, thus resulting in relatively similar land clearance and preparation requirements to the base case upstream raise arrangement.
- Potential for reduced operational requirements due to the increased stability of the structure, however the reduced operational requirements are considered to be similar to that of the downstream option discussed above.
- Increased stability when compared with upstream raised tailings dams due to founding 50% of each raise on natural ground rather than entirely on tailings.
- It is expected that a moderate amount of work (less than upstream construction) will be required for encapsulation of the RSF, prior to closure including the effective compaction of tailings and capping.

This option was identified as a moderate potential cost option for the construction of the RSF. The significant benefits for the centreline option are; increased stability, decreased operational risk benefits and the reduced sustaining CAPEX when compared with the downstream raise arrangement.

5.2.4 Dry Stacking Arrangement RSF

Dry stacking of the tailings, a more recent option adopted by mine owners in Australia, in conjunction with a transition to paste storage, which was considered as an alternative to the embankment tailings storage by TGO. The benefits of dry stacking are seen to be increased stability, reduced CAPEX for embankment construction decreased risk, and decreased cost of closure compared with the embankment options.

When assessing the suitability of this option, the following were identified as key considerations:

- 1. Low embankment CAPEX and sustaining earthworks CAPEX. However, high CAPEX associated with the plant required, OPEX of the mechanical dewatering operations, and of ongoing placement/compaction of the tailings.
- Additional storages for traditional tailings deposition during plant maintenance and runoff water management were identified as being essential.
- Whilst the overall footprint of the dry stack was considered to be reduced when compared to embankment options, the footprint allowance for the water management dam and the slurry RSF was considered to increase the land clearance and preparation requirements, as such the footprint area was comparable if not greater to that of the centreline arrangement.
- Due to the facility location, and the increased risk of wind-blown and eroded tailings (when compared with conventional deposition arrangements) this option was considered to be a significant environmental risk.
- This option was identified as potentially resulting in decreased inspection requirements. However, there are
 increased operational costs with regard to surface drainage management and avoiding high pore pressures
 forming in the deposit which can adversely affect stability.
- It was expected that this option would result in the most cost-effective closure arrangement with the ability to cap and close the facility progressively during operation and the ability to use inert waste rock as a buttressing material on the external slopes.
- Dry stacking can enable progressive capping of the RSF with ongoing placement of tailings. Also, the
 potential for use of waste rock for confining embankments and co-disposal of both tailings and waste rock.

Whilst this option was identified as the lowest potential CAPEX cost option for the storage of tailings in the dry stack earthworks formation, the additional requirement for the construction of two additional storages for wet tailings during filter maintenance and surface runoff partially negates these saving in CAPEX. Additionally, the OPEX associated with the dewatering and placement of the tailings and the identified environmental risks meant that this was not considered to be a justifiable option for TGO at this stage of the mine. However incorporation of additional filtering technology is being considered by the mine for use in the future to maximise capacity and stability of the existing arrangements.

5.2.5 In-Pit Deposition

In-pit deposition of the tailings is generally considered to be a cost-effective solution for deposition of tailings as it eliminates the risk of a tailings dam failure. However, there are numerous elements that need to be considered with respect to this option. The benefits of in-pit deposition are the elimination of the risk of dam failure, reduced CAPEX and sustaining CAPEX, and improved visual amenity following closure and no need for embankment strengthening works during closure. When assessing the suitability of this option, the following were identified as key considerations:

- No embankment CAPEX and sustaining CAPEX, however, previous assessments have indicated there is a risk of groundwater impacts resulting in a need to line the pit using a geosynthetic lining system.
- No additional facility such as a water management dam or emergency storage (as per the requirements for a dry stacking facility) will be required with respect to this option.
- Due to the facility arrangement, there are risks associated with the storage of tailings within the pit due to groundwater impacts, whilst this can be mitigated with the lining of the facility, the use of geosynthetic liners would likely pose a construction difficulty.
- This option was identified as potentially resulting in decreased inspection requirements. However, there will be increased operational costs with regard to dewatering from the pit due to increased pump head. Plus requirements for environmental management of groundwater.

- Due to the storage of tailings within the pit shell, it was determined that there is no risk of embankment failure either during operation or closure.
- This option would result in the elimination of capping external embankments. However, the management of the tailing surface during closure is expected to be difficult due to reduced densities, the closure timeframes are expected to be longer due to dewatering, longer ongoing maintenance of the capping is expected due to ongoing consolidation and creep of the tailings, as well as the long-term risks of environmental impacts to groundwater.

Whilst this option was identified as one of the lowest potential CAPEX cost options for the storage of the tailings, the additional requirement for the lining of the pit negates any savings made in not having to construct an embankment for the storage of the residue. Additionally, the risk of impacting groundwater even with the installation of a geosynthetic lining system was considered too high for TGO as such this option was not considered to be justifiable.

5.2.6 GeoTube Storage

The final option considered was a geotube storage, whereby the tailings are pumped into 'dewatering bags' which passively decant the water out of the tailings through the bag, whilst leaving the tailings solids inside. This deposition arrangement is a relatively new development within the industry and one that allows for effective passive separation of the tailings from the decant water resulting in reduced CAPEX and Sustaining CAPEX throughout the life of the facility and increased stability during closure. When assessing the suitability of this option, the following were identified as key considerations:

- Low embankment CAPEX and sustaining CAPEX are required throughout the life of the facility.
- The overall footprint of the geotubes stack was considered to be similar or greater to that of a centreline raise arrangement due to loss of storage capacity due to the voids between the bags and a maximum limit on the geotube stacking height.
- Additional infrastructure including deposition pipelines will be required for the effective filling of the geotubes.
- This option was identified as potentially resulting in decreased inspection requirements. However, increased
 operational costs were anticipated with regard to managing the storage.
- This option was identified as providing increased stability in the structure when compared with other storage options as the bags are self-containing and effectively unable to fail in the same way a typical RSF would fail. However, it was identified that failures could still occur due to functional failure of the bags themselves or failure of the deposition pipeline.
- It was expected that this option would result in the second most cost-effective closure arrangement with the ability to immediately cap and close the facility following the operation.

Whilst this option was identified as the second lowest potential CAPEX cost option for the storage of the tailings, the additional requirement for the construction of the water management facility, storage area required and the OPEX costs for the deposition management negates these saving in CAPEX. Ultimately this was seen as a good option for potential consideration for future expansion but with the need for more experience in use prior to adoption by TGO eg, further case studies etc.

5.3 Preferred Solution

Based on the analysis, the preferred solution identified throughout this process was the use of traditional slurry storage, using centreline construction for future raises with a compacted clay liner placed across the floor of the storage area of RSF2. This option was selected as the preferred arrangement due to the expected CAPEX and sustaining CAPEX associated with this option, which minimised the footprint of the facility, the anticipated long-term stability of RSF2, and the ease of capping the facility when required. As stated in Section 5.1, taking into account the preliminary nature of the work preparatory leading into the RSF2 detailed design, a number of potential options were considered. However, these options were deemed unsuitable on grounds of cost, environmental risks, and increased delivery risk due to uncertainty in performance. The preferred arrangement is detailed in Section 6 and is in accordance with the project approval conditions.

6. Landform Design

6.1 RSF Arrangement

Based on the Options Assessment detailed above, RSF2 has been designed as a paddock facility with a dual cell arrangement to allow for separated tailings deposition between the two cells to improve the drying capability of the tailings. Whilst a single cell would also allow for adequate drying and rate of rise in the TSF, the use of a central embankment mitigates impacts of an external embankment failure by allowing for an alternative deposition location and aids in construction staging. Each cell of RSF2 utilises a central decant tower and pump surrounded by drainage rock to allow for the removal of the supernatant water and rainfall from each of the cells as required. Further staging of the RSF2 has been developed using a centreline arrangement to reduce reliance on the strength of the tailings for construction whilst minimising the impact of the footprint.

A detailed description of the proposed landform including external and internal embankment geometry and the connection to the existing RSF1 embankment buttress is provided in the following sections.

6.2 External Embankment Geometry

The RSF 2 Stage 1 (starter embankment) has been designed to allow the Stage 2 embankment and all future raises to be constructed as centreline embankments. The RSF2 Stage 1 dam geometry will consist of upstream embankment slopes of 2H:1V which allows for an increased depositional capacity whilst allowing for sufficient stability on the upstream side of the embankment. The downstream slope of the Stage 1 embankment is 3H:1V which allows for increased stability in the Stage 1 embankment whilst minimising the storage footprint.

The design of the Stage 1 embankment allows for a 6 m wide crest on the outer and central embankments to allow for potentially 2-way light vehicle traffic, a tailings deposition line, safety bunding on each side, and sufficient space to undertake operational and maintenance tasks on the pipeline.

The decant access embankments for each cell will be formed also with a 6 m wide crest with 1.5H:1V embankment batters.

6.3 Embankment Zoning

Embankment zoning of the external embankments will consist of Zone 1 (Core and Liner), Zone 2 (filter material), and Zone 3 (General Fill).

6.3.1 Zone 1 Core

The Zone 1 core material will consist of low permeability clay material (as specified in the Technical Specification in Appendix G), nominally 5 m wide, sourced from the storage footprint which shall be placed on the upstream side of the RSF 2 Stage 1 embankment.

The upstream Zone 1 slope will be 2H:1V and the internal slope will be 1H:1V. The Stage 1 Zone 1 material will be keyed into the foundation and connected to the storage liner to minimise the risk of a seepage path forming beneath the embankment.

6.3.2 Zone 1 Liner

Zone 1 Liner will comprise of 1 m thick zone of in-situ clay material (as specified in the Technical Specification in Appendix G), which will be ripped and recompacted to form a low permeability layer (of 1 x 10⁻⁹ m/s minimum) across the foundation of RSF2 to limit seepage through the foundation during operation and closure of the facility.

6.3.3 Zone 2 Transition Material

Zone 2 transition material will be used as a filter layer between the RSF2 northern wall and the existing buttressing on the downstream side of Cell 1 and Cell 2 of RSF1 as specified in the Technical Specification in Appendix G.

6.3.4 Zone 3 General Fill

Zone 3 material will consist of general fill material as specified in the Technical Specification in Appendix G. The foundation will need to be stripped to nominally 300 mm depth to remove topsoil and organic matter and ensure a strong competent foundation material. The Zone 3 material will be selectively placed such that the finer material will be placed against the Zone 1 core and the coarser material will be on the downstream face. This placement methodology is to reduce the risk of fines migrating into the Zone 3 material and assist in the stability and erosion protection of the embankment.

The Zone 3 material is also proposed to be used on the upstream face of the embankment and foundation liner to protect the Zone 1 material from desiccation cracking increasing susceptibility to seepage.

6.3.5 Crest Wearing Course

The wearing course material to form the embankment pavement will likely consist of course durable material found onsite.

6.4 Embankment Connections

RSF 2 eastern and western embankments will abut to the existing RSF 1 Cell 1 and Cell 2 southern embankment and utilize the existing embankment to form RSF 2 northern embankment. A Zone 1 clay liner will be placed against the existing embankment following treatment of the existing buttressing.

RSF 1 includes buttressing on the downstream side of the south embankments which comprises of uncompacted material. The buttressing material contains a matrix of dispersible highly weathered to moderately weathered material, containing boulders up to an approximate size of 600 mm. This uncompacted material when loaded will likely settle resulting in cracking of the Zone 1 clay liner and presenting a risk for piping failure or internal erosion.

To mitigate this risk, a filter (Zone 2) will be required between Zone 1 and the existing buttressing, and Finite Element Modelling (FEM) has been undertaken to model these deformations and assess the impacts on the storage.

The transition filter zone to the existing buttressing will consist of placing a layer of compacted General Fill against the existing buttress and retaining the existing batter slope. Due to the existing geometry of buttressing for Cell 1 being different to Cell 2, additional General Fill will be placed against the buttress as required to create a consistent Northern Wall Profile for RSF2. The General Fill would be required to comprise of appropriately graded material to allow for the construction of a filter zone to be placed against it.

A subsoil drain will be placed in the filter zone to collect and prevent any seepage from RSF1/2 from building up within the embankment and pass any seepage to the external surface toe drains for measurement.

7. Tailings and Water Management

7.1 Overview

This section outlines the proposed deposition arrangement for RSF2 and the water management requirements for the facility to allow for suitable drying to increase the density of the tailings and maximise the capacity within the TSF.

7.2 Tailings Deposition/ Filling Schedule

RSF2 has been designed as a Paddock type facility with a dual cell arrangement to allow for the cycling of deposition as required to improve the density of the tailings over the life of the facility. The tailings will be deposited using perimeter discharge with spigots spaced at 50 m centres along the embankment to allow for even coverage of tailings across the RSF. The following deposition cycle is proposed which is shown graphically in Figure 7.1 and described below.

- Tailings will be deposited from the western wall of Cell 1 of RSF2 cycling 4 spigots every 2.5 days to achieve a deposition depth of 300 mm.
- Following the completion of deposition on the western wall, deposition of tailings will continue around the cell at 4 spigots every 2 days until all sides of the RSF have been deposited in.
- Move deposition to the western wall of Cell 2 of RSF2 and continue cycling 4 spigots every 2.5 days around the entire cell to achieve a deposition depth of 300 mm across the cell.

This deposition cycling, dependent on the variability in the throughput and management of the decant water level, should result in a maximum tailings depth of approximately 300 mm during each cycle optimising the dewatering and increasing the dry density of the materials.

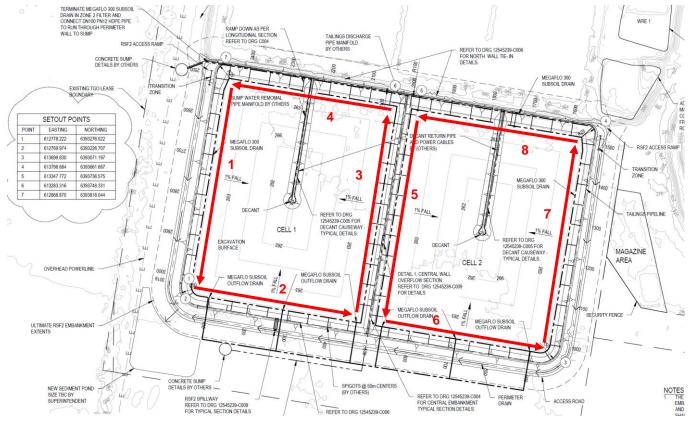


Figure 7.1 Proposed tailings deposition

7.3 Hydrology

As a '**Significant**' Consequence Category dam, the RSF has been designed to have a minimum capacity to store the following rainfall events in accordance with ANCOLD guidelines *Guidelines to Tailings Dams* (ANCOLD, 2019):

- Minimum wet season water storage allowance: 1:10 AEP wet season runoff
- Minimum extreme storm storage: 1:1,000 AEP, 72 hour flood event¹
- Contingency freeboard: 300 mm

In addition to this, the RSF has been designed to store a 1:200,000 AEP storm within the tailings beach and store the PMF within the embankment extents such that the flood water will be temporarily stored in RSF2 to be pumped back into the plant via the decant structure for reuse or to the Residue Storage Dam (RSD) used for excess flood water storage.

Note 1: The minimum extreme storm storage design event is suggested in ANCOLD as 1:100 AEP 72 hour, however, given the importance of the structure to the business moving forward and the impacts of a failure through the external embankment this minimum storage requirement has been increased.

In addition to this, an emergency spillway has been designed for in the southern external embankment of Cell 1 capable of passing a 1:1,000 AEP rainfall event from both Cell 1 and Cell 2 with sufficient freeboard for wave runup during a 1:10 AEP wind event.

The design rainfall events have been estimated by using *Very Rare to Extreme Flood Estimation* (ARR 2019) as shown in Table 7.1.

AEP (years)	1:100	1:1,000	1:2,000	1:10,000	1:200,000	PMP
Design Rainfall (mm) 72 hr event	192	310	347	470	676	996
RSF 2 Storage Volume	109 (ML)	177 (ML)	198 (ML)	268 (ML)	385 (ML)	568 (ML)

 Table 7.1
 Storm events for various AEP (2 cells)

It should be noted that the storage requirements estimations are indicative only and are based on 57 Ha footprint

Given the internal cell dimensions of 475 m x 600 m for each cell, and assuming a beach slope of 1V: 140H based on the previous survey undertaken on RSF 1 Cell 1 and 2, and assuming the decant pond is kept to a maximum level with a 150 m of beach, and the crest height is 0.7 m above the tailings beach, the storage available for each separate cell is summarised in Table 7.2.

Table 7.2RSF2 storage capacity

Storage Area	Single Cell Volume	e (ML) Dual Cell Volume (ML)
With 150 m beach (Maximum Decant Extents)	11	22
With 50 m Beach	73	146
Top of the tailings beach	193	386
Between embankment crest and top of tailings beach including 300 mm freeboard	114	228
TOTAL	307	607

7.4 Flood Storage

The catchment area for RSF 2 is the surface area of the facility, estimated at 57ha which comprises of 28.5 ha for each Cell. The runoff coefficient has been assumed as 1.0. For a 'Significant' dam, the RSF is required to store a minimum of 568ML without spilling to cater for the 72 hour Probable Maximum Flood (PMF).

7.5 Spillway

An emergency spillway has been allowed for in accordance with ANCOLD guidelines *to Tailings Dams* (ANCOLD, 2019), the spillway has been designed allowing for a 1:1,000 AEP 72 hr rainfall event immediately following the PMP. The maximum flood level following the 72 hr PMP event has been assumed as 269.70 m. The 1:1,000 AEP rainfall depth has been calculated by the Bureau of Meteorology and obtained from their website. The temporal pattern that has been adopted from the ARR 2019 Datahub. XPRAFTS has been used to calculate the runoff from storage and route the runoff through the storage prior to discharge through the spillway.

A 6 m wide rock-lined trapezoidal spillway has been designed to convey the design flow event of 1:1,000 AEP event based on a discharge coefficient of 1.56, this allows for a maximum flow rate of 1.71 m³/s occurring between a 4.5-hour and 6-hour event. Based on these flows the critical flow depth through the spillway is 0.285 m.

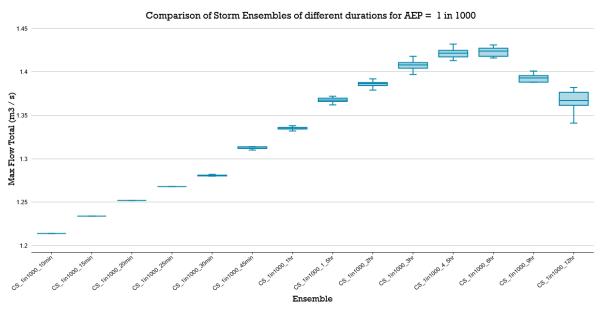


Figure 7.2 Peak outflow for 1:1000 AEP storm following PMP, 6 m Spillway

The spillway crest will consist of a rock-lined earthen approach channel with a concrete crest block, flowing into a geofabric lined, rip rap protected, trapezoidal chute on the downstream face to convey flows to the downstream toe. The erosion protection through the spillway chute has been designed to accommodate the 1.71 m³/s flow as detailed below.

7.5.1 Spillway Erosion Protection

The rip rap sizing through the spillway has been calculated using CHUTE software developed by the Cooperative Research Centre for Catchment Hydrology, utilising the model arrangement shown in Figure 7.3 and model parameters detailed in Table 7.3.

Drawing of a typical chute with explanations of terms

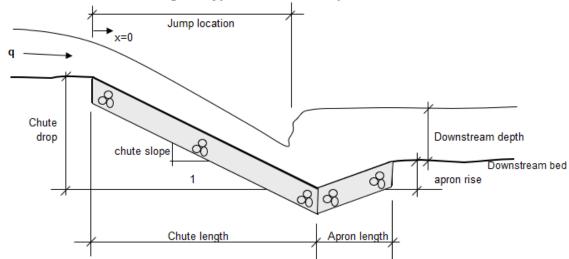


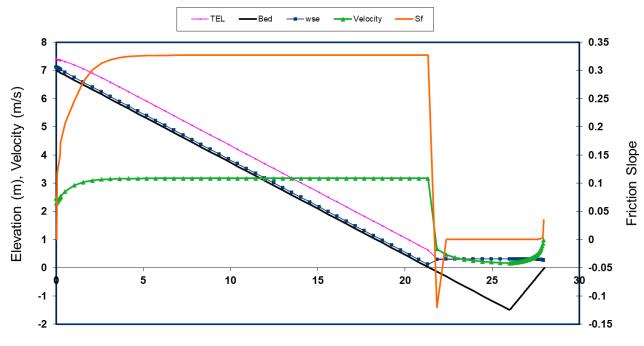


Table 7.3 CHUTE Modelling Parameters

Input Parameter	Value	Unit
Chute Drop	8.5	m
Chute Length	26	m
Apron Rise	1.5	m
Apron Length	2	m
Maximum Flow Rate	1.713	m³/s
Chute Width	6	m
Rock Angle of Repose	42	Degrees
Specific Gravity	2.65	
Factor of Safety	1.5	
Mannings Roughness	0.04	

Based on these model parameters, the following spillway profile in Figure 7.4 was developed to determine the water surface elevation through the chute, expected velocities, and the location of the hydraulic jump which indicates the hydraulic jump occurs within the chute effectively dissipating velocities.

RSF2 Spillway Profile



Distance from Upstream (m)

Figure 7.4 RSF2 Spillway Hydraulic Profile

Based on the proposed spillway arrangement, the rip rap erosion protection requirements are outlined in Table 7.4.

Table 7.4	RSF2 Spillway Erosion Prote	ction Requirements
Table 7.4	KSFZ Spillway Elusion Flote	cuon requirements

Rip Rap Sizing	Value	Unit
D10	0.25	m
D50	0.55	m
D90	0.75	m
Thickness	1	m

7.6 Decant System

The RSF decant area will capture residue bleed water and incidental run-off from the catchment area associated with each of the cells. This water will be returned to the plant for reuse as required.

The pumps required for the decant tower have been sized to allow for the expected runoff from the initial settlement of the material allowing for expected evaporation and seepage into the deposited residue profile as per Table 7.5 below.

Table 7.5	Decant Management Requirements
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Parameter	Value	Unit
Production Rate	1.5	Mtpa
% solids in the slurry	45	%
Particle density/ SG	2.7	t/m ³
The volume of water in the slurry	1.83	Mm³pa
The volume of solids in the slurry	0.56	Mm³pa

Parameter	Value	Unit	
The dry density of the slurry	0.63	t/m ³	
Residue dry density	1.40	t/m ³	
Saturated tails density	1.88	t/m ³	
Water retained in residue	0.48	t/m ³	
Water retained in the residue (volume)	er retained in the residue (volume) 0.52 Mm ³ pa		
Water released (volume)	1.32	Mm³pa	
Water released (volume)	3,609	m³pa	
Water released (volume) 42 I/s		l/s	
Minimum estimated Required Pump Capacity Range for a Single Pump (Refer Note 1)	60 - 80	m ³ /hour	
Minimum estimated Required Pump Capacity Range for a Single Pump (Refer Note 1)	22 L/second		

Note 1 – The required pump capacity does not currently take seepage, evaporation, or emergency dewatering requirements into account.

During normal operations, the decant pumps are only required to flow at 22 L/s, in order to manage the pond level in the RSF, however, a pumping allowance of 50 ML/day per cell should be allowed for by the mine using additional pumping as required. An additional 50ML/day pump in each cell allows for enough capacity to remove ponded water from a 1:10,000 AEP, 72 hr rainfall event in 3 days, and a 1:200,000 AEP, 72 hr rainfall event in 4 days which minimises the inundation of the beach.

Decant towers in each of the cells will be accessed via a causeway constructed from readily available fill materials. Throughout the life of the RSF, the decant causeway will be raised utilising a centreline raise arrangement to meet the crest level of each raise.

The proposed decant tower will be a slotted concrete ring type of arrangement whereby ponded water decants through the slots in the concrete ring tower which will be raised incrementally to remain elevated above the rising residue. The variable speed submersible pump will be installed at the base of the tower for water return to the plant.

In order to remove excess decant and rainfall water from the RSF, a decant pipeline to the existing Residue Storage Dam (RSD) is required. The proposed pipeline arrangement from each decant pump consists of a DN180 HDPE PN12.5, running approximately 1,800 m from each decant pump to the RSD.

8. Water Balance

A site water balance model has previously been prepared for TGO (GHD, 2019). The model was simulated from 1 July 2021 to 1 July 2029, with an initial inventory of process water volume of 18 ML and a mining water volume of about 636 ML. The water balance has been updated by including the catchment of the proposed RSF 2 commencing in May 2022 and reflecting the proposed production schedule with a nominal 1.5 Mtpa production commencing in May 2022 and continuing until July 2029. All other model parameters and inputs were as documented in GHD (2019).

8.1 Process Water

The range of total process water inventory under various potential rainfall conditions are shown in Figure 8.1.

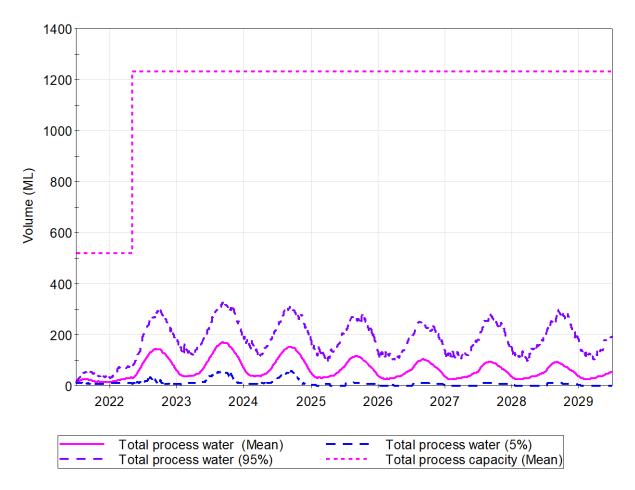


Figure 8.1 Total Process Water Inventory

Figure 8.1 shows that process water inventory is likely to continue to vary, potentially exceeding the 150 ML capacity of the RSD in above average rainfall conditions. However, the total process water inventory is not expected to exceed the combined physical water storage capacity of RSD and the RSFs, especially following the commissioning of RSF 2. Figure 8.1 shows that the potential range of total process water inventory is expected to be increase compared to current conditions. This reflects that the additional process water from runoff of the additional RSF catchment area across RSF 1 and RSF 2.

A sensitivity run was performed, with minimal production demand of 0.5 Mtpa. The results are shown in Figure 8.2.

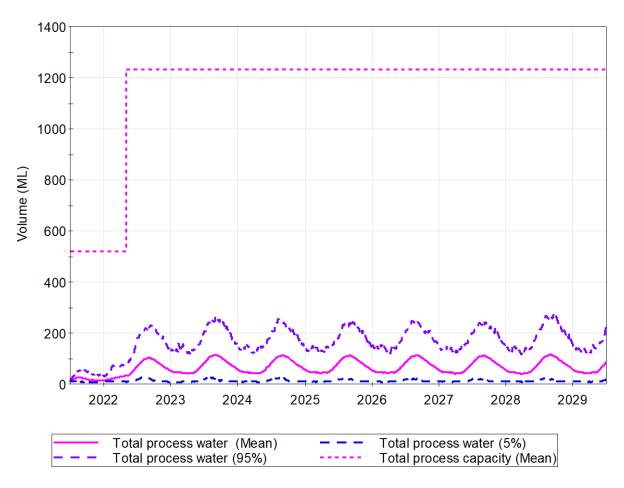


Figure 8.2 Total Process Water Inventory – Minimal Production Case

Figure 8.2 shows a similar process water inventory compared to Figure 8.2, which remains well within the total water storage capacity of the combined process water storage. Therefore, operating the proposed RSF 2 while the existing RSF1 remains uncapped is not expected to result in an unmanageable process water excess.

8.2 Mine Water Inventory and Water Security

The overall forecast site water inventory is shown in Figure 8.3.

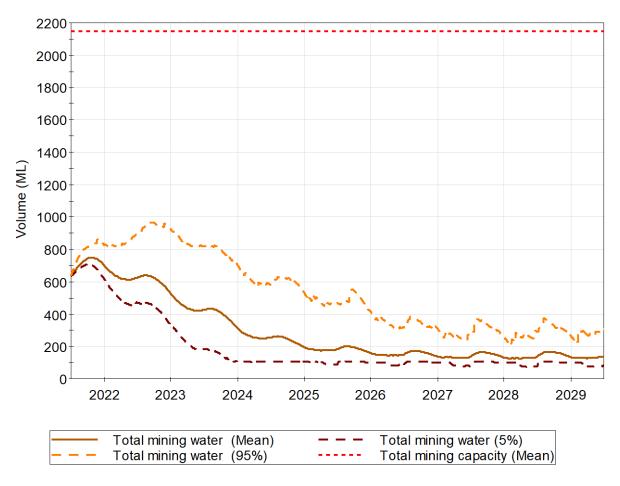


Figure 8.3 Forecast Total Mining Water Inventory

Figure 8.3 shows that mining water inventory is likely to increase until 2022, corresponding to the peak in predicted groundwater inflows, then starts to decline until mid-2026. Total mining inventory remains within the total water storage capacity. The average annual water balance for the year 2026 is summarised in Table 8-1 which is expected to be typical over the remainder of operations at TGO.

Table 8.1 Average Annual Water Balance for TGO

Water management feature	The year 2026 (ML)
Inputs	
Direct rainfall and catchment runoff	678
Supplied from woodlands borefield	960
External water delivery	3
Moisture in ore	75
Secondary release from residue	22
Groundwater inflows	3
Total Inputs	1741
Outputs	
Evaporation from water storages	168
Discharge from sediment dams	5
Potable use	1

Water management feature	The year 2026 (ML)
Water in residue	692
Evaporation from active residue	174
Losses from rewetting of inactive residue	493
Dust suppression	493
Losses from underground workings	7
Total Outputs	2033
Change in Storage	-294

Table 8.1 shows that, on average, the largest inflow into the site water balance is coming from water supplied from Woodlands borefield, accounting for about 55% of total inflows. This may be attributable to the decline in groundwater inflows after the year 2022. With a higher production rate, the largest site demand is ore processing, where water ultimately remains entrained in residue or is lost to evaporation. Table 8.1 indicates an overall decrease in site water inventory on average.

The site water balance model simulated the available borefield allocation over the prediction period. A plot of allocation is shown in Figure 8-4.

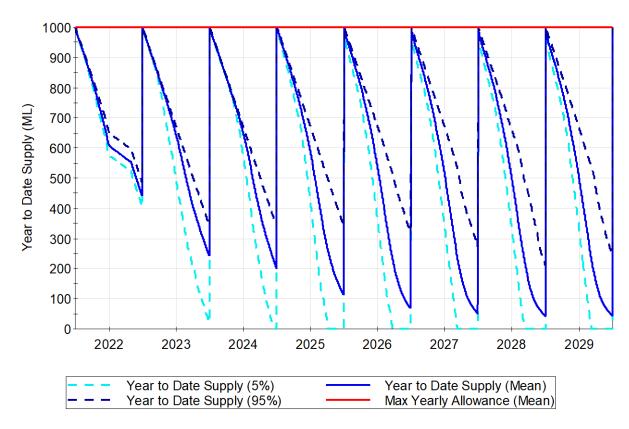


Figure 8.4 Forecast Borefield Allocation

Figure 8.4 shows that with a higher production rate and lower forecast groundwater inflows, water demand from borefield allocation has the potential to be almost completely utilized in the last four years of operations at TGO.

9. Groundwater Impact

9.1 Local Hydrogeology

There are three distinct groundwater systems within the vicinity of TGO's mining leases, as identified by The Impax Group (2011) and Jacobs (2021):

- Shallow alluvium discrete, shallow alluvium (less than 10-20 m deep) dissects the plains surrounding the mine site along creek flow paths. These aquifers are believed to be recharged from rainfall infiltration. Groundwater within these systems is of relatively good quality, however, yields are relatively low and dependent on rainfall. Perched groundwater occurs within the shallow alluvium underlying the RSF, however, it is generally not continuous across the mine site. Shallow groundwater appears to be more permanent along Gundong Creek to the northwest of the RSFs.
- Deep alluvium up to 100 m deep and located approximately 10 km to the northwest and west of TGO.
 Groundwater yields are believed to be low and of poor quality. These systems may have some interaction with underlying bedrock however are believed to be primarily recharged from rainfall.
- Fractured rock the area surrounding Tomingley is underlain by a confined saline groundwater system within the fractured sandstone, siltstone, and volcanics at a depth of greater than 80 m. Groundwater yields range from 0-3 L/s, generally less than 1.5 L/s, and water quality is poor with high salinity (average electrical conductivity (EC) exceeds 20,000 µS/cm). Coffey (2007) investigated this groundwater system as a potential water supply for the mine and found it to be inadequate in terms of both yield and quality.

The hydraulic conductivity of the clay, which comprises the foundation of the existing RSF is generally low to very low. Falling head tests on clayey strata between 1.55 and 42.5 m bgl at the existing RSF area indicate hydraulic conductivities of 0.0002 to 0.002 m/d or 2.3 x 10^{-8} to 1 x 10^{-9} m/s (DEC, 2011). In addition, for the recent RSF 2 geotechnical investigation (GHD 2021a), permeability testing was undertaken on four samples (TP03, TP10, TP11, and TP18) collected from the site. The results of the permeability testing are summarized in Table 9.1.

Test ID	Depth (m)	Description	Coefficient of permeability, k (m/sec)	
TP03	2.0 - 3.0	CLAY trace sand	2 x 10 ⁻¹⁰	
TP10	0.0 - 1.0	Sandy CLAY, trace gravel	2 x 10 ⁻¹⁰	
TP11	1.0 – 2.0	Sandy CLAY	4 x 10 ⁻¹¹	
TP18	1.0 – 2.0	Sandy CLAY, trace gravel	1 x 10 ⁻¹⁰	

 Table 9.1
 Laboratory Permeability Testing (GHD 2021a)

Shallow perched groundwater, where it occurs throughout the TGO site, is typically at a depth of less than 10 m bgl. For the RSF 2 geotechnical investigation, 10 groundwater bores (BH01 to BH10) were installed within the alignment of the proposed embankment foundation to a target depth of 20 m. Groundwater was encountered in BH01, BH04, and BH08 at 7.5 m (RL 257.2 m), 8.0 m (RL 255.8 m), and 7.5 m (RL 258.4 m) below ground level, respectively (GHD 2021a).

The water-bearing zone within the deep confined fractured rock groundwater system occurs at an elevation below 190 m AHD, based on observed groundwater inflows into the WYO3 pit and groundwater monitoring bore data. The groundwater is under pressure, as indicated by the monitoring bore data showing groundwater levels ranging from approximately 200 m and 240 m AHD.

Groundwater usage is limited in the vicinity of the mine site. The closest active production bores (i.e. non-test or monitoring bores) are over 3 km to the north of the mining lease area within shallow alluvium (GW034897, GW037395, and GW803148) with all reported yields less than 1.5 L/s. These bores are registered for stock and domestic, irrigation use, and town water supply, respectively.

9.2 Analysis of Foundation Seepage

Although seepage through the foundation from the existing RSF has not been detected by the existing shallow groundwater monitoring bore network, a conceptual analysis of potential seepage through the foundation of RSF2 Stage 1 and 2 has been undertaken.

9.2.1 Methodology

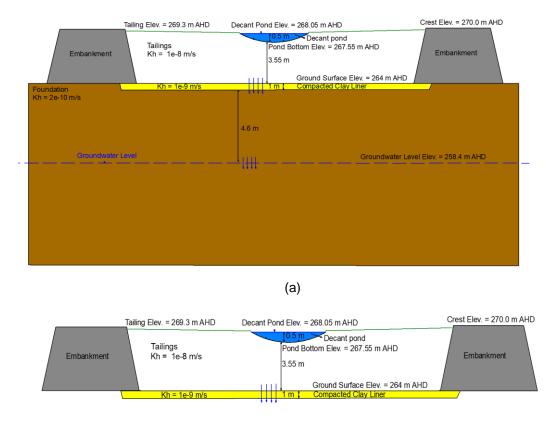
The seepage analysis involved a one-dimensional calculation of vertical advective flow from the RSF 2 decant pond into the underlying foundation. The calculation was based on the Darcy flow equation. For both RSF 2 stage 1 and stage 2, the rate of seepage and time for seepage to occur were calculated under two scenarios:

- Scenario 1: seepage to shallow strata through the residue and CCL only.
- Scenario 2: seepage to shallow groundwater through the residue, CCL, and foundation.

The conceptual analysis of the seepage through the foundation of RSF 2 stage 1 and stage 2 are shown schematically in Figure 9.1 and Figure 9.2.

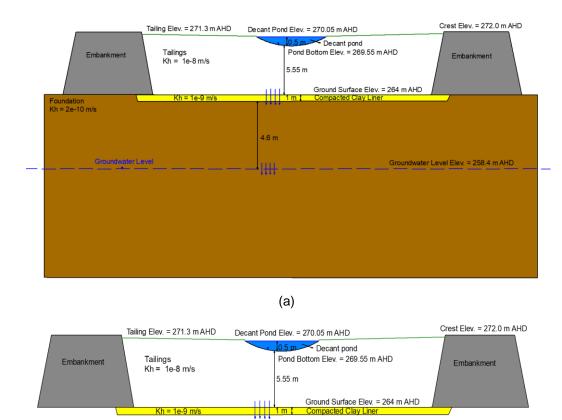
The following inputs and assumptions were applied:

- For Stage 1, the decant pond water level was assumed to be managed at RL 268.05 m and 0.5 m deep, while for Stage 2 the decant pond water level was assumed to be managed at RL 270.05 m and 0.5 m deep.
- Maximum decant pond area of 0.49 ha (70 m x 70 m).
- The base of the residue was assumed to be RL 264.0 m.
- The residue has a permeability of 1×10^{-8} m/s (DEC, 2011).
- The CCL is 1 m thick and has a permeability of 1×10^{-9} m/s (refer to Table 2.2).
- The influence of the embankment on seepage was not considered.
- Negligible water pressure at the base of the tailings.
- Foundation permeability of 2×10^{-10} m/s was adopted (based on GHD 2021a).
- Groundwater level at the RSF 2 of RL 258.4 m. This is based on the groundwater level recorded at BH08 at 7.5 m (GHD 2021a). It should be noted that this is generally a conservative approach since regional groundwater occurs within a confined groundwater source at elevations below RL 190 m, and recorded groundwater levels reflect water under pressure. However, it is noted that groundwater levels at WYMB06 are also influenced by water levels within the old McPhail workings.
- Seepage rates are calculated under steady-state conditions.



(b)

Figure 9.1 Stage 1 Conceptual Seepage Cross-Section Through the (a) Residue, CCL and Foundation, and (b) Residue and CCL



(b)

Figure 9.2 Stage 2 Conceptual Seepage Cross-Section Through the (a) Residue, CCL and Foundation, and (b) Residue and CCL

9.2.2 Results

The results of the seepage analysis are shown in Table 9.2. The calculated seepage rates below the CCL to shallow strata are 1.6 kL/day for Stage 1 and 1.9 kL/day for Stage 2. Seepage is calculated to occur after approximately 40 years. The calculated seepage rate to shallow groundwater is approximately 0.2 kL/day and is predicted to occur after a timeframe of approximately 900 years. These calculations are subject to the assumptions outlined in Section 9.2.1, however, they indicate a low risk of seepage of RSF decant through the CCL and RSF foundation throughout the current 7-year life of the facility. Once the RSF is closed and rehabilitated, and the residue is dewatered, the risk of foundation seepage is further reduced.

The analysis suggests a negligible incremental change in seepage rate to shallow groundwater and seepage time between Stage 1 and Stage 2 rise.

Scenario	Stage 1		Stage 2	
	Calculated seepage volume (kL/day)	Seepage time (yrs)	Calculated seepage volume (kL/day)	Seepage time (yrs)
Scenario 1	1.6	39	1.9	46
Scenario 2	0.154	796	0.182	823

 Table 9.2
 RSF2 Stage 1 and Stage 2 Seepage Analysis Results

9.3 Analysis of Embankment Seepage

Seepage through RSF2 has been undertaken for Stage 1 and Stage 9 raises. The analysis is consistent with the results and input data adopted for the foundation seepage analysis, as previously presented. The seepage and hydraulic gradients found are later used as input in the LEM slope stability assessment.

9.3.1 Model Characteristics

Seepage analysis was executed only for the perimeter embankment as there are no major water gradients between the two residue storage facilities. The Rocscience finite element groundwater package Slide2 was used in this analysis. The upstream boundary condition was defined by a water table representing the pond at its maximum operational level, 90 m from the beach edge. Similarly, the downstream boundary condition corresponds to the encountered groundwater level as found at BH08 (GHD 2021). The upstream toe drain has been considered to be clogged, thus modelling a conservative scenario for Stage 1, while for Stage 9 this is the most likely condition to be present. In addition to the isotropic hydraulic conductivities described for the RSF materials, a $k \approx 5 \cdot 10^{-7}$ m/s has been adopted for the general fill material.

9.3.2 Results

The seepage model results for the RSF embankment are shown in Figure 9.3 and Figure 9.4 for Stage 1 and Stage 9, respectively. From the results, it can be seen that the resulting water table, shown as the light blue line, extends from the decant pond through tailings, abruptly drops in the embankment clay, and further extends within the general fill as the groundwater level is attained. The phreatic surface in the embankment generally remains near the foundation soil with average hydraulic gradients close to 3 across the embankment clay and below 0.3 across the general fill material. Therefore, piping or internal erosion risk is minor.

Based on the input parameters, the model returned seepage results of $4.5 \cdot 10^{-4} \text{ m}^3/\text{d/m}$ for Stage 1, which is equivalent to 226.5 L/d for a cell side equivalent to 500 m length. As expected, the seepage for Stage 9 is significantly larger than that for Stage 1, with maximum seepage of $3.5 \cdot 10^{-3} \text{ m}^3/\text{d/m}$, equivalent to 1,750 L/d.

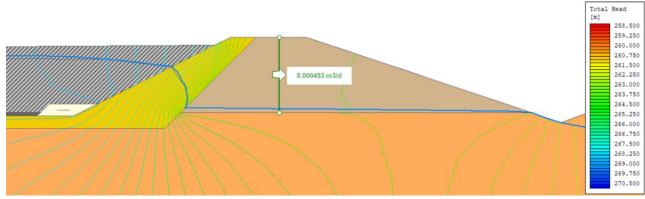


Figure 9.3 Perimeter Embankment Stage 1 Seepage Modelling Results

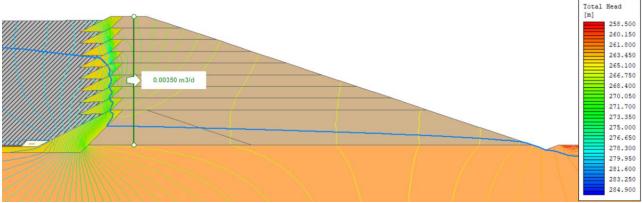


Figure 9.4

Perimeter Embankment Stage 9 Seepage Modelling Results

9.4 Monitoring Program

It is recommended that shallow monitoring bores be installed around the outside of the embankment of RSF 2 to detect seepage through the foundation. Monitoring bores should be installed to depths of 10 m below ground level (bgl) and at an interval of approximately 250 m around the perimeter of the RSF. Bores should be screened between 2 m depth and the base of the bore. The new RSF monitoring bores should be incorporated into the monthly groundwater monitoring program at TGO. Ongoing monitoring of the regional bores near the site - WYMB01 and WYMB06 – should also occur.

RSFs can influence local groundwater pressure, which can be detected by increasing groundwater levels, however, this does not necessarily mean that seepage is occurring. A line of evidence approach is necessary for interpreting monitoring data whereby the spatial and temporal trends in both groundwater levels and groundwater chemistry should be assessed to determine whether seepage is occurring.

10. Slope Stability Analysis

10.1 General

Slide2 software package was used to undertake static slope stability analysis of the RSF2 for Stage 1 and up to Stage 9. Two-dimensional analyses were performed in two representative cross-sections, the southern perimeter embankment, and common wall embankment through the RSF. The Morgenstern-Price and Spencer methods were adopted in calculating the Factor of Safety (FoS) against non-circular failures. The geotechnical parameters adopted correspond to those summarised in Section 4. Groundwater conditions were obtained from seepage analysis presented in previous Section 9.3. For the common wall, a water surface line has been included to represent phreatic conditions when the pond is at its maximum design extends (90 m beach).

10.2 Geometry

The adopted cross-sections for Stage 1 and Stage 9 are shown in Figure 10.2. Stage 1 and Stage 9 embankment crest levels are at RL 270 m and RL 286 m, respectively.

The embankment shows a 6 m wide crest with a downstream slope of 3H:1V, and an upstream slope of 2H:1V, constructed using centreline construction as shown in Figure 10.1. Tailings management strategy and geotechnical properties for tailings are expected to remain similar to those found at RSF1 during 2020 investigations. Tailings beach profile has been considered as 140H:1V in accordance with the RSF1 beach profile.

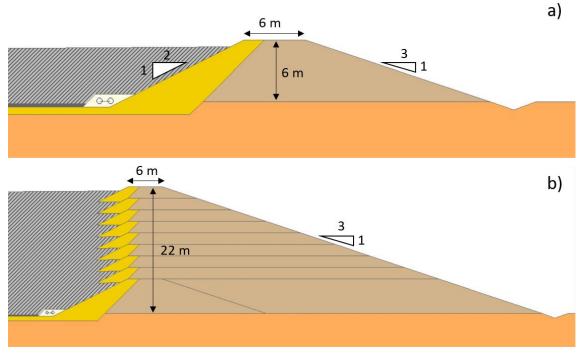


Figure 10.1 Perimeter Embankment Cross-Section. a) Stage 1. b) Stage 9

The internal embankment, located through the centre of RSF2, has been designed as a divider to create two cells to allow for better tailings management. This embankment will be raised using a 'centreline' construction method. The embankment geometry consists of a 6 m wide crest and 1.5H:1V slopes on both the upstream and downstream face with crest level at RL 270 m for Stage 1 and RL 286 m for Stage 9. The internal embankment cross-section is shown in Figure 10.2. \u78y

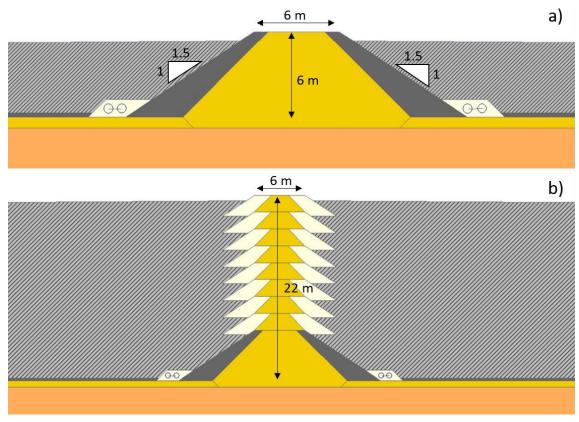


Figure 10.2 Common Wall Cross-Section. a) Stage 1. b) Stage 9

10.3 Loading Conditions

The loading conditions assessed were based on the recommendations provided by ANCOLD Guidelines on Tailings (ANCOLD, 2019) and are shown below in Table 10.1.

Loading Condition	Recommended minimum safety factor for tailings dams	Shear strength to be used for evaluation
Long-term drained	1.5	Effective strength
Short-term undrained (potential loss of containment)	1.5	Consolidated undrained strength
Short-term undrained (no potential loss of containment)	1.3	Consolidated undrained strength
Post-seismic	1.0-1.2 ²	Post seismic shear strength ³

Note 1 See Section 6.1.3 of ANCOLD (2019) of the description of loading conditions.

Note 2 To be related to the confidence in the selection of residual shear strength. 1.0 may be adequate for use with lower bound results.

Note 3 Cyclically reduced undrained/drained shear strength and/or liquefied residual shear strength for potentially liquefiable materials.

From these loading conditions and RSF2 design, short-term undrained (potential loss of containment) applies to the perimeter embankment and short-term undrained (no potential loss of containment) applies to the common wall and upstream failure of the perimeter embankment.

10.4 Stability Analysis Results

10.4.1 Stage 1 Stability Results

The FoS and slip surfaces at Stage 1 for the long-term drained condition for perimeter and internal embankment are shown in Figure 10.3 and Figure 10.4, respectively. For the former, deep slip surfaces, extending from the upper third of the buttress to toe, are found. The failure occurs within the downstream shoulder of the embankment not extending within tailings mass. The obtained FoS is 1.674, which is above the minimum required of 1.5 for this loading case. For the common wall, a high FoS is obtained due to the confining stress imposed by tailings in most of the embankment height. The slip surface is surficial and located near the crest with limited development within the tailings.

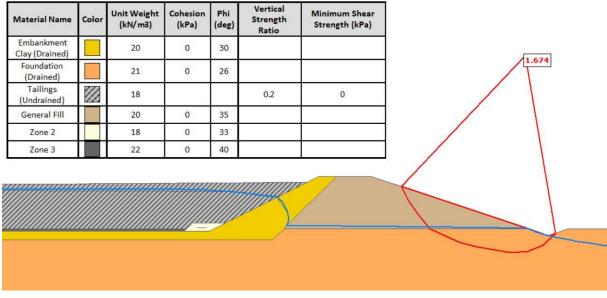


Figure 10.3 Slip Surface in Perimeter Embankment, Long-term Drained Loading Condition

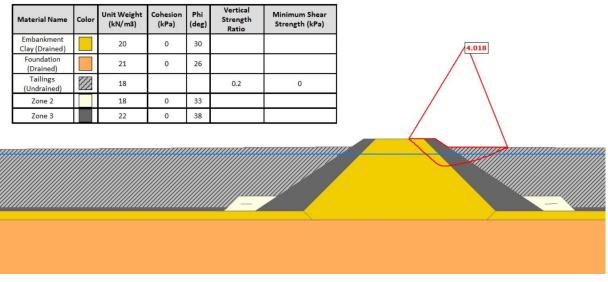


Figure 10.4 Slip Surface in Common Wall, Long-term Drained Loading Condition

The other modelled scenarios show FoS well above minimum design criteria, for both perimeter embankment and common wall, the results for which are summarised in Table 10.2. All the failure surfaces for the different scenarios modelled as well as the corresponding FoS are summarised in Appendix D.

10.4.2 Stage 9 Preliminary Stability Results

The FoS and slip surfaces at Stage 9 for the short-term undrained condition for perimeter and internal embankment are shown in Figure 10.5 and Figure 10.6, respectively.

The failure surface in the perimeter embankment extends from the upstream batter near the crest toward and elongating through general fill shoulder and foundation soil. The slip surface is contained within the general fill and foundation materials with negligible effects on the embankment clay and tailings. The FoS for this scenario is 1.62 which meets ANCOLD 2019 design criteria. The slip surface in the common wall develops within the upper raises, extending from the crest, Zone 2 filter, and tailings storage. The FoS is 1.42 which meets the criteria for a short-term undrained condition with no potential loss of containment. The FoS results for all the scenarios considered are shown in Appendix D.

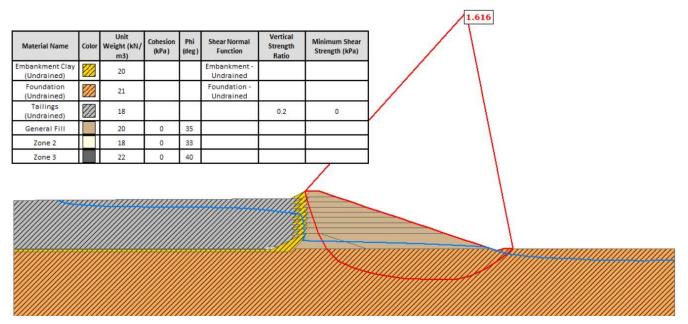


Figure 10.5 Slip Surface in Perimeter Embankment Downstream Batter Slope, Short-term Undrained Loading Condition

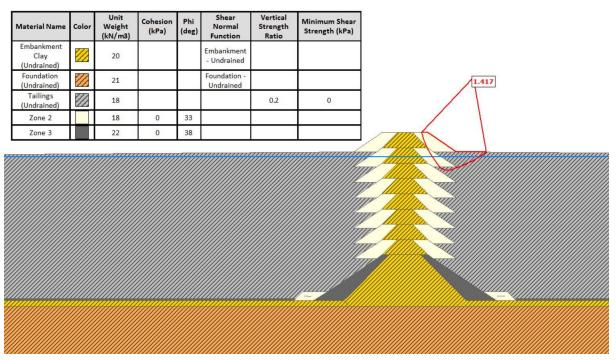


Figure 10.6 Slip Surface in Common Wall, Short-term Undrained Loading Condition

For the post-seismic analysis, where residual strength parameters are adopted for the tailings, a minimum FoS of 1.176 is obtained as shown in Figure 10.7, which meets the criteria for a post-seismic conditions. The failure surface develops within the downstream shoulder and foundation soil.

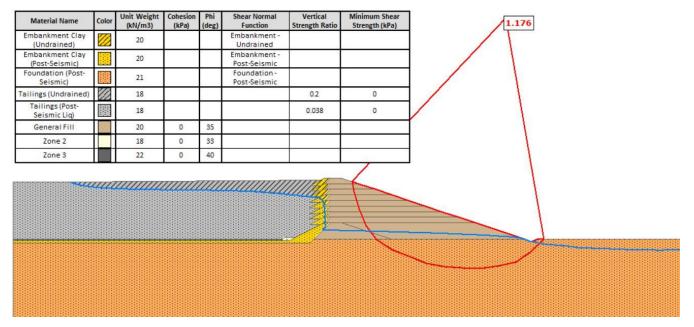


Figure 10.7 Slip Surface in Perimeter Embankment Downstream Batter Slope, Post-Seismic Loading Condition

10.4.3 Summary

A summary of FoS for LEM analysis is shown in Table 10.2. The FoS shows that ANCOLD design criteria are met for all the considered scenarios.

	Long-term drained condition	Short-term undrained condition (potential loss of containment)	Short-term undrained condition (no potential loss of containment)	Post-seismic
Required FoS	1.5	1.5	1.3	1.0 – 1.2
Perimeter embankment – Stage 1	1.674	1.826	1.819	1.490
Perimeter embankment – Stage 9	1.702	1.616	1.704	1.176
Common wall – Stage 1	4.018	-	1.613	3.498
Common wall – Stage 9	3.312	-	1.417	2.770

Table 10.2 Obtained FoS from Stability Analysis

11. Static Deformation Analysis

11.1 General

This section presents information, methodology, and results related to static deformation analysis using Finite Elements Method (FEM) software package Plaxis 2D. The model was developed as a two-dimensional representation of the RSF1/RSF2 interaction zone, this being the RSF1 southern embankment and RSF2 northern embankment. The FEM analysis is useful to simulate the complex soil stress-strain behaviours and deformations experienced during the construction and tailings deposition stages.

The main objective of the deformation analysis is to assess deformations and displacements under RSF2 filling. The specific goals of the deformation analysis include:

- To assess the deformations and crest settlements on RSF1 embankment caused by RSF2 tailings deposition.
- To assess the deformations of the filter and clay materials at the drainage system located at the RSF1 toe.

11.2 Geometry and Meshing

The analysis has been undertaken on a typical cross-section at the southern embankment of RSF1 and the northern embankment of RSF2.

The geometry for RSF1 has been adopted from the RSF1 Stage 8 design (GHD, 2021b). The main geometrical features of the RSF1 embankment consists of a starter dam followed by 8 upstream lifts, including a 6 m wide embankment crest, 3H:1V downstream batter slope, 1.5H:1V upstream batter slope, downstream buttress constructed in 5 phases (starting from the fifth embankment raise stage) and tailings distribution as shown in Figure 11.1.

RSF2 tailings have been modelled considering the 9 stages, with a freeboard and beach slope similar to that adopted for RSF1. RSF2 tailings have been considered to be a sand-like type of tailings, as this is the most observed tailings type in the existing RSF1. A transition zone and a zone 1 blanket have been also modelled downstream from the buttress. These can be observed in Figure 11.2.

The phreatic conditions were adopted from seepage analyses executed for RSF1 at Stage 8. The intermediate stages and RSF2 phreatic conditions were estimated considering the expected groundwater level.

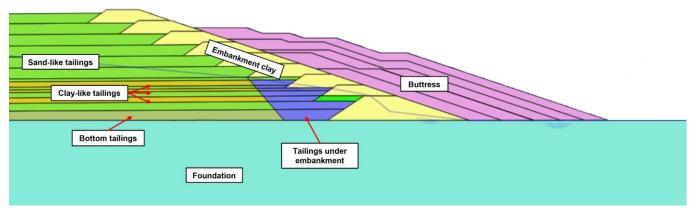


Figure 11.1 RSF 1 Extents following Stage 9 Embankment

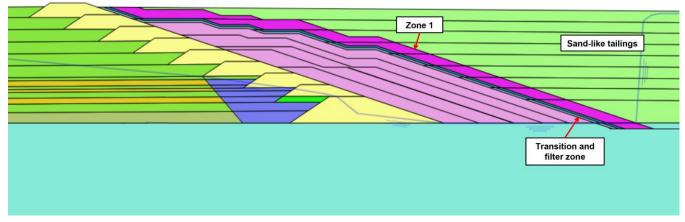


Figure 11.2 Conceptual RSF 1 Stage 9 and RSF 2 Stage 9 Connection Arrangement

The model was discretised in a mesh formed by 10,514 triangular finite elements. Each element contains 15 nodes and 9 stress points. Finite elements size varies according to material type and model geometry, but in general terms, bigger elements are presented in the foundation soil while tailings and embankments are discretised with smaller elements. More than 90% of finite elements have a quality index greater than 0.7 (1 representing the maximum value for perfectly equilateral triangles), which implies a good discretisation of the model. The phreatic surface is shown as an indicative representation assuming underdrainage remains 100% operational over the life of the facility, this will be assessed over the life of the facility based existing operations.

11.3 Material Properties

Two different constitutive models were used to model the expected behaviour of materials under static loading conditions.

The buttress, embankment clay, foundation, and zone 1, zone 2, and zone 3 materials were simulated considering a Mohr-Coulomb (M-C) constitutive model. The adopted M-C parameters are, in general, the same as those used in the limit equilibrium stability analyses, except for the effective Poisson ratio and effective Young's modulus values which were not required in the stability analyses. These parameters are detailed in Table 11.1. Most of the soil parameters adopted are consistent with previous geotechnical analyses which had already been subjected to a peer RSF1 review process.

Material	Bulk Unit Weight (kN/m³)	Effective Friction Angle (deg)	Effective Cohesion (kPa)	Effective Poisson's Ratio	Effective Young's Modulus (MPa)
Buttress	18	28	0	0.3	30
Embankment Clay	21	30	0	0.2	20
Foundation	21	29	0	0.3	50
Zone 1: Clay	21	20	0	0.3	20
Zone 2: Filter	21	32	0	0.3	50
Zone 3: Transition	21	32	0	0.3	55

Table 11.1 Mohr-Coulomb Parameters

The tailings were simulated with the Hardening Soil (HS) constitutive model to better capture their non-linear behaviour and expected consolidation due to the dissipation of pore pressure after loading. Plaxis uses a permeability coefficient to carry out consolidation calculations. Adopted permeability values are presented in Table 4.1. The HS parameters are summarized in Table 11.2. These values are consistent with the parameters used in Stability Modelling and Liquefaction Triggering Analyses for Stage 7 RSF1 (GHD, 2021c), which has previously been subjected to a peer-review process.

Material	Bulk Unit Weight	Saturated Unit Weight	Effective Friction angle	Dilatancy angle	Effective Cohesion	E _{50-ref}	Eeod-ref	Eur	Power	Reference Pressure
	(kN/m³)	(kN/m³)	(deg)	(deg)	(kPa)	(kPa)	(kPa)	(kPa)	(m)	(kPa)
Sand-like tailings	18	19.5	31	0	0	29,000	29,000	87,000	0.5	100
Clay-like tailings	18	19.5	26	0	0	600	600	1,800	0.7	100
Tailing under embankment	18	19.5	26	0	0	400	400	1,200	0.7	100
Bottom tailings	18	19.5	26	0	0	1,100	1,100	3,300	0.7	100

11.4 Methodology and Modelling Sequence

11.4.1 Methodology

Plaxis 2D is a software package designed to specifically deal with highly nonlinear geotechnical problems such as deformations in complex geometries, stress-strain generation, and consolidation. A few specific features are worth noting about the deformation analysis:

- Staged construction: The construction of RSF1 and tailings deposition sequences in RSF1 and RSF2 have been modelled in stages, in order to capture as realistic as possible the stress distribution, strains generation, and excess pore pressure of the embankment and tailings materials during the life of the facility.
- Drained and undrained behaviour of tailings: The timeframe of RSF1 construction was not considered because this part of the analysis was conducted considering drained behaviour of tailings and no excess of pore pressure dissipation was modelled. For the second part of the analysis, where RSF2 tailings deposition were modelled, tailings in RSF1 were considered with an undrained behaviour to capture pore pressure generation and effective stresses. Because the focus of the analysis is RSF1 deformations, RSF2 tailings are modelled with a drained behaviour. A period of time was considered for consolidation between each RSF2 tailings deposition stage.
- Zero displacement setting: After RSF1 construction was finished, displacements were reset to zero to focus
 on the displacement generated only as the result of RSF2 tailings deposition. However, the stress-strain state
 of the different materials was maintained as calculated.

11.4.2 Modelling Sequence

Although the focus of this analysis is the deformations due to RSF2 tailings deposition, the construction and tailings deposition sequences of RSF1 have been modelled in order to realistically capture the stress distribution of the embankment and tailings materials prior to the RSF2 tailings deposition. The following sequences have been adopted in the RSF1 construction process:

- Stage 1: Set up the initial stress stage of the in-situ field prior to the construction of the Starter Dam (See Figure 11.3).
- Stage 2: Construction of the Starter Dam. Tailings deposition to RL 269.85 m.
- Stage 3: Construction of embankment raise 2. Tailings deposition to RL 271.85 m.
- Stage 4: Construction of embankment raise 3. Tailings deposition to RL 273.85 m.
- Stage 5: Construction of embankment raise 4. Tailings deposition to RL 275.85 m.
- Stage 6: Construction of embankment raise 5 and buttress 1. Tailings deposition to RL 277.85 m.
- Stage 7: Construction of embankment raise 6 and buttress 2. Tailings deposition to RL 279.85 m.
- Stage 8: Construction of embankment raise 7 and buttress 3. Tailings deposition to RL 281.85 m. (See Figure 11.4).
- Stage 9: Construction of embankment raise 8 and buttress 4. Tailings deposition to RL 283.85 m.

- Stage 10: Construction of embankment raise 9 and buttress 5. Tailings deposition to RL 285.85 m.

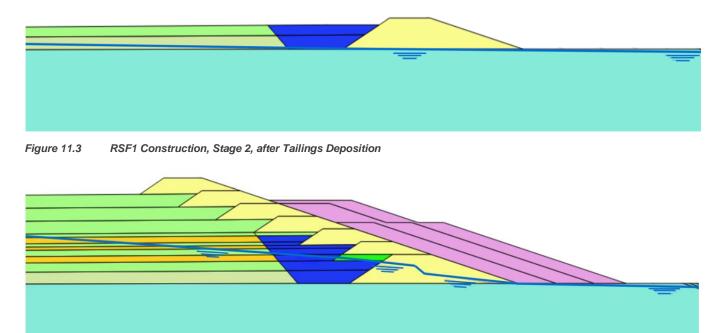


Figure 11.4 RSF1 Construction, Stage 8, before Tailings Deposition

After completion of the RSF1 construction, RSF2 tailings deposition has been modelled. The RSF2 filling has been carried out in 9 stages consistent with the 9 embankment raises considered for the life of the facility. To account for tailings consolidation and excess pore pressure dissipation, several days have been considered between each 2 m deposition stage to represent the maximum rate of rise. At the end of the RSF2 tailings deposition, the model has been run for long-term drying without any additional load, to simulate excess pore pressure dissipation. The following sequences have been adopted in the RSF2 filling process:

- Stage 11: Blanket excavation at the RSF1 downstream toe (See Figure 11.5). Consolidation phase of 10 days. Blanket construction.
- Stage 12: Drainage system construction. Tailings deposition to RL 269.85 m (RSF2 Starter Dam).
 Consolidation phase of 100 days.
- Stage 13: Drainage system construction. Tailings deposition to RL 271.85 m (RSF2 embankment raise 2).
 Consolidation phase of 100 days.
- Stage 14: Drainage system construction. Tailings deposition to RL 273.85 m (RSF2 embankment raise 3). Consolidation phase of 100 days.
- Stage 15: Drainage system construction. Tailings deposition to RL 275.85 m (RSF2 embankment raise 4). Consolidation phase of 100 days.
- Stage 16: Tailings deposition to RL 277.85 m (RSF2 embankment raise 5). Consolidation phase of 100 days (See Figure 11.6).
- Stage 17: Tailings deposition to RL 279.85 m (RSF2 embankment raise 6). Consolidation phase of 100 days.
- Stage 18: Tailings deposition to RL 279.85 m (RSF2 embankment raise 7). Consolidation phase of 300 days.
- Stage 19: Tailings deposition to RL 279.85 m (RSF2 embankment raise 8). Consolidation phase of 300 days.
- Stage 20: Tailings deposition to RL 279.85 m (RSF2 embankment raise 9). Consolidation phase of 300 days.
- Stage 21: Long-term consolidation phase of 2,074 days.

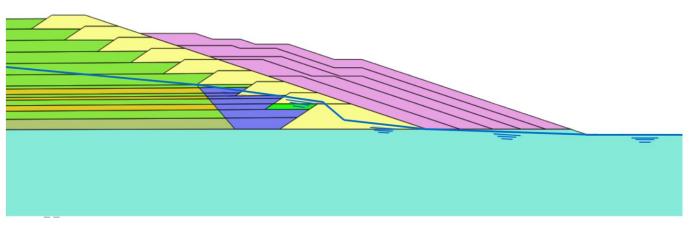


Figure 11.5 RSF2 Filling, Stage 11, Blanket Excavation

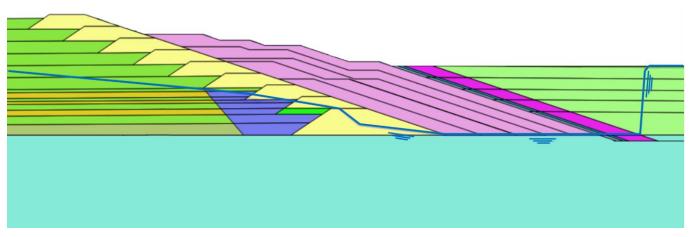


Figure 11.6 RSF2 Filling, Stage 16

11.5 Results

To understand the deformations generated in RSF1 southern embankment resulting from RSF2 tailings deposition results in terms of displacements, shear strains, and excess pore pressure are presented in the following section. Finally, the summary section presents conclusions regarding the key findings.

11.5.1 Displacements

The following results can be observed from the analysis in terms of displacements:

- The maximum total vertical displacements are located just over the tailings under the embankment, associated with the low elastic modulus of this material. Figure 11.7 shows that a large portion of the displacements is developed in the early stages of the RSF 1 construction. It is noted that displacements of 1 m occur for Stage 2 of the model, with a total displacement of 5 m at the end of RSF1 construction. This is a numerical estimation that is not necessarily perceived in the field as construction accommodates deformations over time.
- Consequently, at the beginning of RSF2 filling, these displacements have been reset to zero and no further displacement occurs at the central axis of RSF1 embankment until Stage 19 of the model when RSF2 tailings are deposited at this location. The maximum vertical displacement at the end of RSF2 filling modelling is 0.3 m at the right side of the model where tailings are fully saturated and at the central axis of RSF1 embankment. This is shown in Figure 11.8.
- The total horizontal displacements are significantly smaller than the total vertical displacements. The maximum displacement occurs in the tailings downstream of the drainage system.
- It was observed that during the blanket excavation in Stage 11 of the model, displacements are limited.
 Maximum total displacement occurs at the RSF2 impoundment base with a value of 6 mm The influence of

such change in the structural geometry extends toward the middle of the embankments downstream slope, however it can be seen that the displacements are smaller than 2 mm.

A sensitivity analysis has been conducted for the RSF1 buttress Young's modulus (E'). This material has been placed using end dumped fill and is not compacted. The adopted E' values were 30 MPa, 20 MPa, and 15 MPa. In the RSF1 embankment raises, maximum total vertical displacement increases 2.5% when E' decreases from 30 MPa to 20 MPa and 9% when E' decreases from 30 MPa to 15 MPa. For the same changes in E', maximum total vertical displacement in tailings located at the central axis of RSF1 embankment increase 8% and 16%, respectively. The influence of this parameter on the drainage system is greater than in the previously commented zones of the structure. Maximum total vertical displacement increases 13% when E' decreases from 30 MPa to 20 MPa. For an effective Young's modulus value of 15 MPa maximum total vertical displacement is 0.15 m (30% increase from the base case with E'=30 MPa). This is still considered to be acceptable for the material.

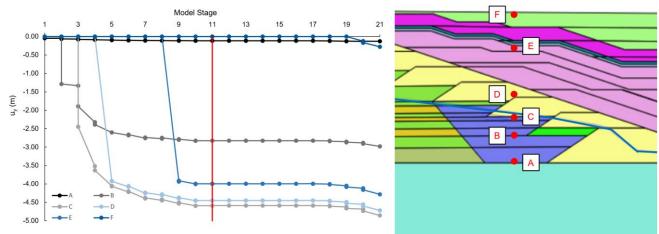


Figure 11.7 Vertical Displacement at Centre Axis of RSF1 Embankment

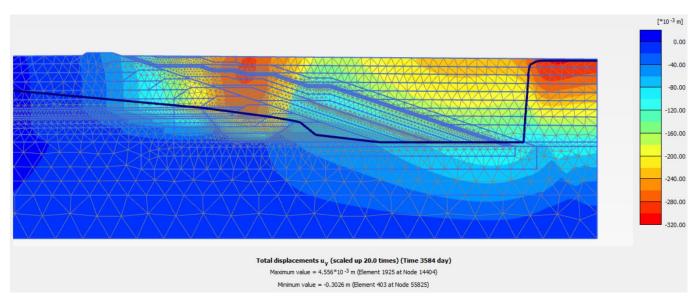


Figure 11.8 Total Vertical Displacement at Stage 21

11.5.2 Shear Strains

Shear strains are analysed in terms of total deviatoric strain. In general terms, total deviatoric strain generation during the modelling stages is quite similar to the total displacements. During RSF1 construction, the greatest values of deviatoric strain concentrate in the tailings under embankment material, again associated with the low effective Young's modulus. When tailings deposition starts at RSF2, maximum deviatoric strains are located at the

freshly deposited tailings mostly below the water table. Maximum values in these stages start at 0.3% and increase to reach maximum values of 2% at Stage 19 of the model.

At the end of tailings deposition in RSF2, maximum total deviatoric strains move back to the tailings under the embankment. Figure 11.9 shows this quantity for Stage 21, at the end of the long-term consolidation phase with a maximum value of 10.8%.

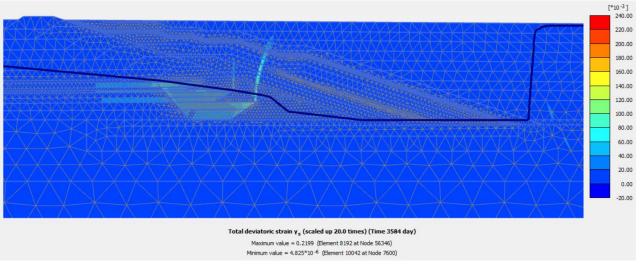


Figure 11.9 Total Deviatoric Strain, Stage 21

11.5.3 Pore Pressure

As previously commented, during RSF1 construction, tailings, embankments materials, and foundation were modelled with drained behaviour, therefore, no excess pore pressure is generated during the first 10 modelling Stages.

The timeframe between each lift in RSF2 is expected to be similar to the period between RSF1 raises; this is one year approximately. However, analysis results showed that excess pore pressure dissipation occurs in less time for the first Stages. Because of that, between Stage 12 and Stage 17 of the model, a consolidation period of 100 days was considered appropriate, reducing calculation time without affecting the modelling of the consolidation process in tailings. For the remaining deposition Stages (18 to 20 of the model) the actual timeframe between RSF2 embankment raises is adopted for consolidation, which means a significant dissipation of excess pore pressure, but not completely.

The greater values of excess pore pressure are generated in tailings under the embankment material and contiguous tailings, as presented in Figure 11.10, which shows excess pore pressure for Stage 20 of the model before consolidation, the maximum value for which is 87.9 kPa. At the end of the long-term consolidation phase (Stage 21 of the model), excess pore pressure drops to a maximum value of less than 10 kPa which is distributed uniformly within the tailings as the water migrates through the facility.

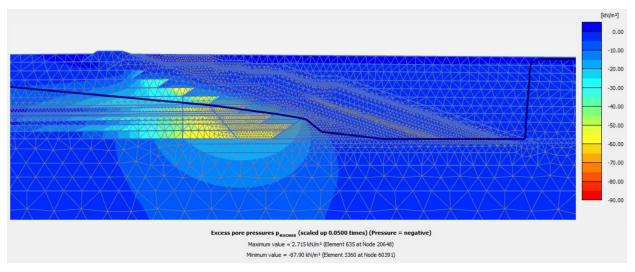


Figure 11.10 Excess Pore Pressures, Stage 20, before Consolidation Phase

11.5.4 Summary

The deformation analysis on the RSF1/RSF2 connecting embankment executed using FEM has demonstrated the final condition of the interaction between RSF1 and RSF2 is stable. The results of the analyses suggest the following.

- The maximum displacement occurs in the tailings beneath the RSF1 embankment material, due to its low stiffness. During RSF1 construction, total vertical displacements are very high in this zone of the model. This is consistent with the occurrence of settlement and cracking in RSF1 during embankment raise 2. After displacement was reset to zero at the beginning of RSF2 filling, total displacements were significantly lower, with maximum values around 0.3 m at the end of the modelling, but displacements at the axis that cross tailings under embankment are still high compared with other parts of the structure. The maximum settlement at the RSF1/RSF2 filter zone mean that the combined 2 m width of filter zones are not expected to settle enough to prevent drainage of the external RSF1 shell. It is recommended that these models be reviewed with each lift to assess actual settlement with predicted settlement to confirm the effectiveness of the connection drainage system.
- Displacements of the embankment raise and buttress are limited. Sensitivity analysis on the effective Young's modulus of the buttress shows that the greatest influence on displacement occurs at the downstream drainage system zone, however, the calculated values are still within acceptable magnitudes.
- Excavation at the RSF1 embankment toe for blanket and drainage system construction causes negligible effects on the embankment, in terms of displacement and strains, and therefore, in terms of stability.
- Excess pore pressure generation concentrates at the tailings under the RSF1 embankment material. The adopted periods used in the modelling are adequate to correctly simulate consolidation between tailings Stages. A period of 2,074 days (post-closure) is required for the final long-term consolidation phase, to dissipate excess pore pressure until a value below 10 kPa is achieved.

12. Seismicity Assessment

12.1 Overview

During the Stage 7 RSF1 embankment raise, the Seismology Research Centre (SRC), a division of Earth Sciences Pty Ltd (ESS) were engaged to undertake a site-specific seismic hazard assessment. The study can be found in SRC (2020) and is summarised in the following section.

12.2 Study Summary

Although the mine site is located in a region of low seismicity, the study reviewed the historical records of seismicity and ground motion in this zone. The review found that the largest seismic events that could occur at the site are 7.5 M_w, although unlikely, it is expected that an earthquake of such magnitude would result in extreme amounts of damage.

The earthquake recurrence models used in the study considered the local site conditions, non-linear analyses, seismotectonic models, and active and non-active faults closer to the mine site. All these factors were assessed to obtain the maximum and minimum magnitudes to be considered, resulting in a 7.5 Mw magnitude quake for the maximum credible event, and a 4.5 Mw magnitude quake for the minimum credible event.

The ground motion models used in the construction of the hazard curves considered the variability caused by earthquake magnitude, distance, wave frequency, randomity, historical attenuation functions, and epistemic uncertainties. All of these factors were used to select the best fit ground-motion models.

As part of the site-specific seismic hazard study, a probabilistic hazard assessment was undertaken, indicating the minimum earthquake magnitude that should be considered is 4 M_w and the maximum distance for an earthquake to have an effect on the mine site is 300 km.

Calculations for this study were carried out using the weighted mean of three seismotectonic models, the AUS7 model, which has been developed by SRC, together with two smoothed seismicity models, the Cuthbertson model (CUTH) and the Risk Frontiers model (RF).

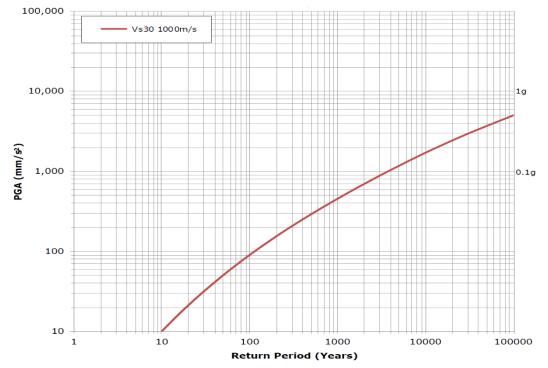
The results of the seismic hazard studies result in seismic hazard curves that present PGA and Mercalli Intensity as a function of the return period together with the sensitivity analysis of the different models used. Response spectra for different return periods are presented for informational reasons or for future works.

In addition to the seismic hazard study, a time-history analysis was carried out. Earthquake events were selected from different locations (2 Australian and 4 worldwide) with similar conditions. These event records were processed according to current methods, considering the conditions of the mine site, to generate the time-history simulations. The obtained results are consistent with those obtained from the seismic hazard curves in terms of PGA.

12.3 Study Review

ANCOLD Guidelines for Design of Dams and Appurtenant Structures for Earthquake (2019b) recommends that for a '**Significant**' consequence category dam a 1:475 AEP Operating Basis Earthquake (OBE) and 1:1,000 AEP Safety Evaluation Earthquake (SEE). However, the guidelines also note that if the structure is susceptible to liquefaction (such as the tailings or embankment foundations) or has components that will fail at ground motions only a little greater than those recommended, a check of the design for the critical ground motion. Where liquefaction of tailings is likely to affect the stability of the dam, the design may need to consider a 1:10,000 AEP earthquake. ANCOLD (2019b) also recommends 1:10,000 AEP Maximum Credible Earthquake (MCE) be used in the design of the closure arrangement. As such the SEE earthquake considered in the design is the equivalent of a 1:10,000 AEP earthquake event.

The seismicity assessment SRC (2020) shows that the actual PGA for the OBE (1:475 AEP) is slightly higher than that adopted during previous design stages, resulting in an increase from 0.015g to 0.028g. Based on the analysis, the PGA for the SEE (1:10,000 AEP) is smaller than the previously adopted value, decreasing from 0.360g to 0.180g.



A PGA of 0.360g corresponds to a return period of 40,000 years, according to the SRC study which is very unlikely to occur and exceeds the design criteria for RSF2. The complete PGA recurrence curve is shown in Figure 12.1.

Figure 12.1 PGA Recurrences for the Tomingley Gold Mine RSF.

The study results illustrate the expected earthquake magnitude for a given return period. The magnitude values adopted for this report were estimated considering the most reasonable magnitude values, the seismic hazard assessment, the time history analysis, and previous design stages adopted values. These considerations are expected to be conservative for the RSF2 Stage 1 embankment design. A summary of the earthquake loading parameters adopted for this report is shown in Table 12.1.

Table 12.1	Seismic Loading Parameters
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Event	Magnitude	PGA
OBE (1:475 AEP)	4.5 Mw	0.028 g
SEE (1:10,000 AEP)	6.5 Mw	0.180 g
MCE (1:10,000 AEP)	6.5 Mw	0.180 g

13. Liquefaction Assessment

13.1 Overview

Based on the existing tailings in RSF1, the liquefaction of tailings to be contained in RSF2 requires a strainsoftening soil response and strength loss either during seismic or monotonic loadings. It has been proved that significant strains and strength loss can occur in a broad range of saturated soils, from sand to clay. Liquefaction phenomena are likely to occur in very loose sands and very sensitive low-PI clays with a large brittleness index, i.e. IB≈1. If the phenomenon is triggered within the RSF2 facility, increased loading is expected in the future stages of the retaining structures and large strains may develop. As such, it is crucial to evaluate the post-seismic response of the facility, adopting residual or liquified shear strength parameters for materials prone to liquefaction, as later discussed.

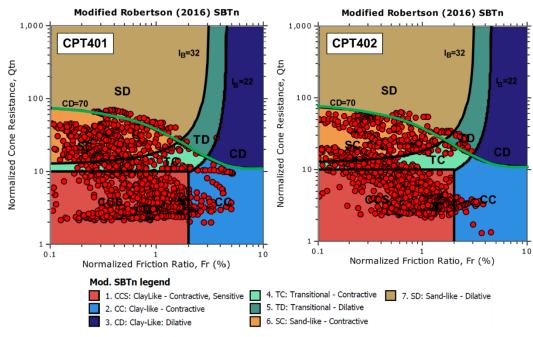
The use of the CPTu data from the existing RSF1 has been deemed to be a rational approach for the assessment of the state parameter and soil classification, and as such, is deemed adequate for the evaluation of liquefaction susceptibility of the RSF2 materials.

13.2 Liquefaction Susceptibility

13.2.1 Tailings

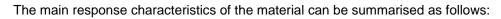
The 2020 CPTu test data on RSF1 was used to produce Soil Behaviour Type (SBT) Charts (Robertson, 2016) in terms of the normalised cone resistance vs normalised friction ratio, as shown in Figure 13.1 for CPT401 and CPT402.

The contour CD=70 in green indicates the contractive-dilative (CD) boundary, which separates soils with contractive or dilative behaviour at large shear strains. Normalized Cone Resistance (Q_{tn}) and Normalized Friction Ratio (F_r), are represented by the following expression (Robertson, 2016):



$$CD = 70 = (Q_{tn} - 11) \cdot (1 + 0.06F_r)^1$$

Figure 13.1 CPT401 and CPT402 SBT Plots



- In terms of soil behaviour type index (I_B) and SBT behaviour boundaries, the tailings show a mixed behaviour with sand, transitional, and a clay-like type of response. This is consistent among CPTs, which show an interbedded stratification as previously discussed.
- Inspection of the chart shows that tailings are generally below the dilation-contraction contour. Most of the data points show the material's tendency to contract upon shearing with a significant portion of the material lying in the SC regions.
- Similarly, a large portion of data points to plot in the CCS and CC zones. Thus, there exists a potential for cyclic softening and flow liquefaction and S_{u(rem)}/σ'_v should be estimated for the stability analysis. For this type of material large strains are required to trigger strength loss.
- Figure 13.2 shows the results of the normalised penetration resistance corrected to an equivalent clean sand value (Q_{tn,cs}). Q_{tn,cs}≈70 represents the contractive-dilative contour. It is clear that all CPTus, except for CPT408 that commences at RL 270.5 m, show contractive behaviour at all depths.
- This Q_{tn,cs}≈70 contour is similar to the dilative/contractive boundary for the state parameter ψ≈-0.05, as found by Jefferies and Been (2006) and Shuttle and Cunning (2009) for coarse grained uncemented soils. Using this approach for CPTu data, it can be confirmed that most of the soils show ψ≥-0.05, and as such, tailings are susceptible to liquefaction.

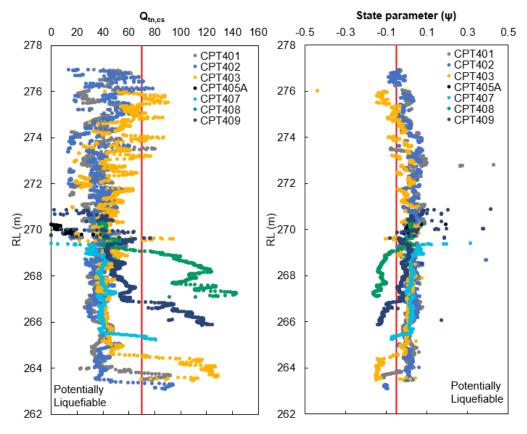
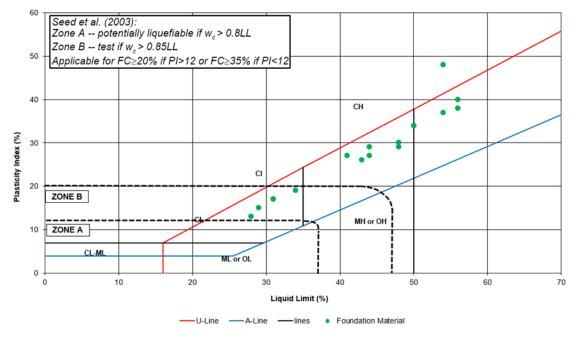


Figure 13.2 Q_{tn,cs} and State Parameter

Regarding nature, fabric, and tailings management strategies, it is expected that tailings at RSF2 behave similarly to those existing at RSF1.

13.2.2 Foundation Soil

The foundation material for RSF2 is a clay-type soil that, in limited RSF2 CU triaxial tests, has shown strainsoftening responses. Generally, for low-plasticity silts and clays, there is insufficient guidance on the engineering procedures that are most appropriate for estimating potential strains and strength loss upon undrained loading. Regarding the PI (greater than 13%) of the material, and the research of Boulanger and Idriss (2004, 2005), the soil is likely to show a clay-like behaviour with cyclic softening under cyclic loading. Different liquefaction susceptibility criteria are commonly adopted in practice, e.g. the Chinese criteria have been widely used since the early 1980s for evaluating the liquefaction susceptibility of silts. The susceptibility of foundation material to liquefy is assessed in Figure 13.3 through the liquefaction criteria proposed by Seed et al. (2003). This criterion describes three zones on the Atterberg limits chart: Zone A soils with a PI≤12 and a LL≤37 are potentially liquefiable if w≥80% of the LL; Zone B soils with a PI≤20 and a LL≤47 are considered potentially liquefiable with laboratory testing recommended if w≥85% of the LL; and Zone C soils with a PI>20 and a LL>47 are considered generally not susceptible to cyclic liquefaction. Considering these criteria, and the testing undertaken in the RSF1 and RSF2 foundation materials which found it to be a Zone B material, the foundation is considered to be potentially liquefiable if the water content is greater than 85% of the LL.





13.3 Seismic and Static Liquefaction

The liquefaction phenomena can be triggered under monotonic or seismic loading. The trigger can be a result of natural events, such as an earthquake; or due to anthropic activities, such as the rapid loading over the tailings. The liquefaction resistance and liquefaction triggering assessment of the RSF2 materials are discussed in the following sections with reference to experimental testing conducted in RSF1 and RSF2 footprint.

13.3.1 Dynamic Liquefaction

13.3.1.1 Tailings

CPTu tests have been used to undertake a liquefaction potential assessment for the tailings. The liquefaction potential of the soil can be determined by estimation of two variables:

Cyclic Stress Ratio (CSR) is a measure of the cyclic load applied to the soil by the earthquake. CSR is
estimated through expected PGA and a stress reduction factor:

$$CSR = 0.65 \cdot \left(\frac{a_{max}}{g}\right) \left(\frac{\sigma_{v0}}{\sigma'_{v0}}\right) \cdot r_d$$

 Cyclic Resistance Ratio (CRR) is the capacity of the soil to resist liquefaction. This can be determined from laboratory or in situ tests. For CPTu, CRR is found using the expression proposed by Robertson (2009) which is based on a review of case histories. If the CSR is greater than the CRR, liquefaction is likely to occur. The analysis of the data was undertaken using CLiq software, and results are shown in Figure 13.4 for the OBE and SEE events. The Factor of Safety (FoS) in the figure corresponds to the ratio between CRR and CSR for a 7.5 M_w event.

For the OBE event ($4.5M_w$ and PGA 0.028g), the CSR is smaller than the CRR, with FoS generally greater than 2 in most of the soil depths. Thus, liquefaction is unlikely to occur under this seismic scenario. For the SEE event ($6.5M_w$ and PGA 0.18g), the FoS is generally below unity for most of CPTu, as such, the material is likely to experience seismic liquefaction.

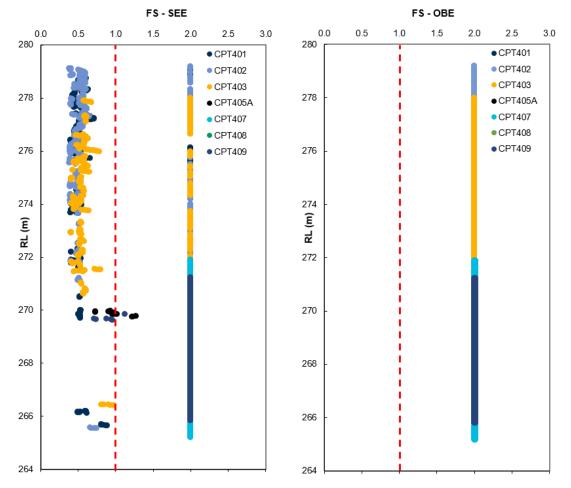


Figure 13.4 Seismic Liquefaction Potential – SEE (0.180g 1:10,000 AEP); OBE (0.028g 1:475 AEP)

13.3.1.2 Foundations

For the foundation soil, the seismic liquefaction assessment was undertaken considering the SPT results of the boreholes GBH01, GBH02, and GBH03. The CRR was obtained in accordance with Idriss et. al (2004) where CRR_{M=7.5} for clay-like soil (PI \ge 7) can be estimated as follows:

$$CRR_{M=7.5} = 0.8 \cdot \frac{S_u}{\sigma'_v} \cdot K_\alpha$$

Figure 13.5 shows a comparison of the calculated CSR and CRR for SPT data under OBE and SEE scenarios. The CRR is for all data points greater than CSR for both OBE and SEE earthquakes. As such, liquefaction is not likely to be triggered in the foundation soil.

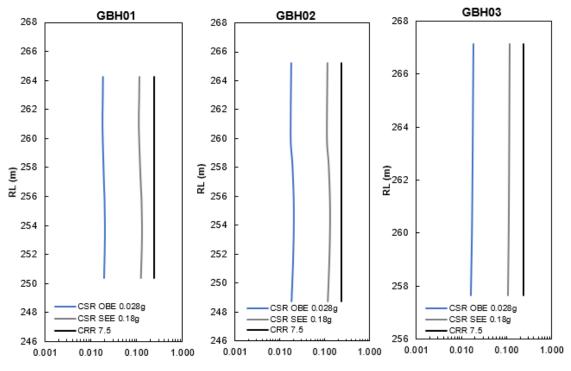


Figure 13.5 CSR vs CRR_{7.5} Plots – Foundation

13.3.2 Static Liquefaction

RSF materials such as tailings and foundation soil may be susceptible to static liquefaction if one or more of the following conditions occur:

- Sudden increase of phreatic level due to run-off or seepage.
- High rate of rise or rapid vertical loading during disposal of tailings or raises construction.
- Uncontrolled shear strains due to lateral displacements of the foundation soil or as a result of excavations near RSF toe.

This can only be controlled if proper actions are undertaken to avoid static liquefaction, such as controlling material placement rates or permanent phreatic surface monitoring.

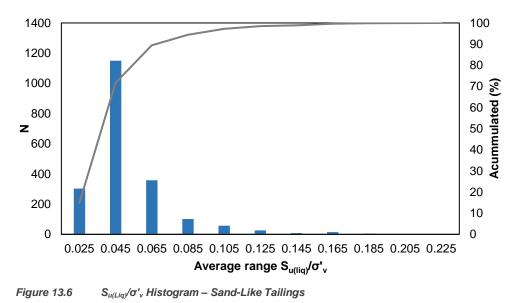
13.4 Post Liquefaction Shear Strength

13.4.1 Tailings

13.4.1.1 Post-Seismic Liquefaction

According to the results of the CPTu conducted at RSF1, a large portion of data points plot in de SC (sand-like – contractive) and TC (transitional contractive) zones, however, these tailings are located toward the top of RSF1 Stage 1. Similarly, a large portion of data points plot in the CC (Clay-like – contractive) and CCS (Clay-like – contractive – sensitive) zones. Thus, there exists a potential for cycling softening and flow liquefaction within these tailings, thus remoulded undrained shear strength ratio ($S_{u(iiq)}/\sigma'_v$) should be estimated for stability analyses.

Figure 13.6 shows a histogram for $S_{u(LIQ)}/\sigma'_v$ for the CPTu data based. The sand-like material and the equations proposed by Robertson (2019) (based on the analysis of liquefied strength ratios from flow failure case histories) were chosen as a conservative approach to estimate S_u for RSF2. Adopting the first quartile as a reasonable criterion for LE analysis, $S_{uLiq}/\sigma'_v \approx 0.04$.



13.4.1.2 Post-Static Liquefaction

During static liquefaction, residual undrained shear strength is attained. The $S_{u,res}/\sigma'_{vo}$ can be found using the Critical State Line (CSL) and a representative value for the state parameter ψ as discussed in GHD (2021b). For the analysis, a $S_{u,res}/\sigma'_{vo}\approx 0.15$ has been conservatively adopted for the tailings to be stored in RSF2.

13.5 Deformation

A seismic deformation assessment was carried out for the stockpile using the simplified empirical and probabilistic methods of Pells & Fell (2002), Swaisgood (2013), and Bray et al. (2007).

13.5.1 Pells and Fell (2002)

This empirical method uses PGA and earthquake magnitude to classify a dam according to expected damage (i.e. settlement and longitudinal crack width), Table 13.1.

Damage Class		Maximum Longitudinal	Maximum Relative Crest
Number	Description	Crack Width (mm)	Settlement
0	No or Slight	<10	< 0.03%
1	Minor	10 – 30	0.03% - 0.2%
2	Moderate	30 - 80	0.2% - 0.5%
3	Major	80 – 150	0.5% - 1.5%
4	Severe	150 – 500	1.5% - 5%
5	Collapse	> 500	> 5%

 Table 13.1
 Damage Classification System, Pells & Fell (2002)

Utilising the earthquake magnitude and PGA's developed as part of the Seismicity Assessment, the OBE and SEE events result in a Number 0 Damage Class and a Number 1 Damage Class respectively. The 0 Damage Class results in a maximum crest settlement of 0.03% of the Stage 1 Embankment, whereas the 1 Damage Class results in a maximum crest settlement of 0.2% of the Stage 9 Embankment. The results of this analysis are seen in Table 13.2.

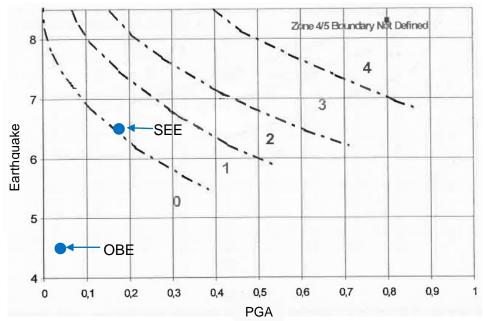


Figure 13.7 Contours of Damage Class Versus Earthquake Magnitude and PGA

13.5.2 Swaisgood (2013)

Alternatively, this method considers an extensive review of case histories of embankment dam behaviour during an earthquake. The original data considers hydraulic fill, earthfill, and other types of rockfills dams. Figure 13.8 shows historic crest settlements of embankment dams that have been subjected to major earthquakes as a function of the reported PGA (USBR. 2015).

Using the regression analysis proposed by Swaisgood, the amount of settlement expected for the embankment crest is 0.01% and 0.05% for OBE and SEE, respectively. This corresponds to a "Minor" relative degree of damage. For the maximum RSF2 Stage 9 embankment height, the settlements are summarized in Table 13.2.

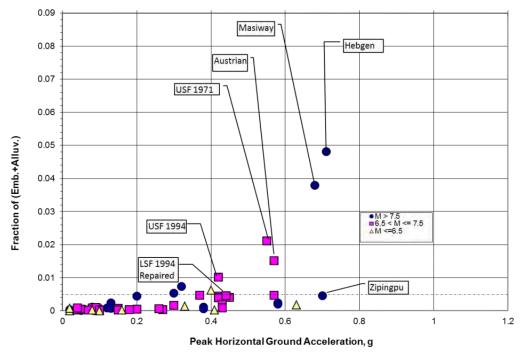


Figure 13.8 Earthquake Induced Settlements of Embankment Dams, USBR (2015)

13.5.3 Bray et al. (2007)

Bray et al. proposed a simplified procedure for estimating earthquake-induced deviatoric slope displacements. The authors, using a comprehensive database of 688 recorded ground motions, propose the following equation to compute seismic displacements.

$$\ln(D) = -1.10 - 2.83 \ln(k_y) - 0.33 (\ln(k_y))^2 + 0.57 \ln(k_y) \cdot \ln(S_a) + 3.04 \ln(S_a) - 0.24 (\ln(S_a))^2 + 1.5T_s + 0.28(M - 7) \pm \varepsilon$$

where k_y is yield coefficient; T_s is the initial fundamental period of the sliding mass in seconds; S_a is the 5% damped elastic spectral acceleration of the site's design ground motion at a period of $1.5T_s$ in the unit of g; *M* is the earthquake's moment magnitude, and ε normally distributed random variable with zero mean and standard deviation (σ) of 0.66. This method is discussed in the ANCOLD Guidelines for Design of Dams and Appurtenant Structures for Earthquake (ANCOLD 2019).

For the RSF2 Stage 9 embankment, the yield coefficient is estimated in LEM analysis and found as equal to 0.180; T_s is calculated using expression $T_s = 2.6H/V_s$. Shear wave velocity in the tailings is expected to be greater than 350 (m/s), and thus, T_s smaller than 0.238 s. S_a is obtained from design spectra for Sites class A and B. The maximum seismic displacement as obtained from this method is summarised in Table 13.2.

13.5.4 Summary of Result

Table 13.2	Maximum Settlement Results for Different Methods
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			Maximum settlement by Bray (mm)
OBE	7	2	3
SEE	44	10	6

The results show that seismic crest settlements for the Stage 9 embankment are negligible with maximum values of less than 44 mm under SEE loading. This is expected due to relatively low seismic features in the area and the geotechnical characteristics of the RSF2.

14. RSF 2 Closure and Rehabilitation

14.1 General Requirements

Closure for the facility will conform to industry standards to develop a final landform that is:

- Physically safe to humans and animals.
- Geotechnically stable.
- Geochemically non-polluting/ non-contaminating.
- Capable of sustaining an agreed post-mining land use.
- Decommissioned and rehabilitated in an ecologically sustainable manner.

Based on these the aims of the closure design are as follows:

- Ancillary infrastructure associated with the RSF2 and stockpiles shall be removed and the RSF2 shall be made safe.
- The closure landform shall be free draining, stable, and non-polluting.
- The cover and landform design shall require minimal ongoing surveillance and maintenance post closure.
- Vegetation used in the closure of the facility shall be self-sustaining.

14.2 RSF 2 Closure Arrangement

Given the location of RSF2 in relation to RSF1 a single closure landform is being developed to assist in the assimilation of the landform into the surrounding topography and optimize water management elements of the design.

14.2.1 External Geometry

The closure arrangement of the RSF 2 involves a range of elements to assist in the successful encapsulation and isolation of tailings from the surrounding environment to assist in the successful rehabilitation of the area to achieve a safe and stable landform. The following design elements are included in the closure of the RSF.

The closure landform for the RSF2 facility will consist of a water-shedding cover as detailed below in Section 14.2.2, this cover will be graded at 1% from the centre of the northern embankment toward the southern, eastern, and western embankment to allow for effective and safe removal of rainfall. The cover then extends down to the downstream face of the existing embankment which will be constructed with a downstream face of 3H:1V. It is recommended a detailed closure plan for the RSF1 and RSF2 storages be prepared following the initiation of construction of the RSF1 Stage 8 embankment raise to detail the closure arrangement, planning and operation requirements for the two facilities.

Further detail as to the combination of RSF1 and RSF2 closure facilities to form a single landform will be assessed during the development of the aforementioned RSF Closure plan.

14.2.2 Capping Arrangement

The capping design for the tailings and embankment will consist of the following items:

- Following the completion of the tailings placement and removal of the decant water, the tailings shall be
 prepared (allowed to dry and or be compacted by track rolling) such that they have a suitable bearing capacity
 for the placement of the fill to achieve the final landform and low permeability capping layer.
- Following the conditioning of the tailings, general fill / Zone 3 material will be placed across the tailings to form the closure landform profile and a workable platform for the efficient placement of the clay capping.
- Following reprofiling, a low permeability layer shall be placed over the tailings to reduce the potential for infiltration. The conceptual arrangement of the capping allows for a minimum 0.5 m clay layer across the TSF surface for a low permeability cover, however, alternatives including reduced thickness liners, geosynthetic liner systems or geo-composite liners may be considered during further design stages.

 A 1.0 m waste rock layer will be layered from finer to coarser material directly over the low permeability to form a non-erodible growth medium with topsoil layer placed over the waste rock to form a rock mulch layer on the tailings and embankment and followed by revegetation using local occurring grassland species.

Identification of potential material sources and planning for investigations into the volume and suitability of the materials will be detailed in the aforementioned RSF Closure plan.

14.2.3 Civil Infrastructure Decommissioning

Following the final placement of tailings within the RSF, the civil infrastructure associated with RSF2 including the tailings deposition pipeline can begin to be decommissioned. The decommissioning of the civil infrastructure shall begin with the tailings deposition pipeline while the dewatering infrastructure is used to reduce the size of the pond in each of the cells. Following the removal of the decant ponds the water management infrastructure will be decommissioned including the removal of pipework and infilling and capping of the concrete decant tower.

14.3 Constructability Assessment

14.3.1 Constructability Assessment

The following key construction activities are discussed in this section:

- Dewatering of the Residue Storage Facility.
- Tailings movement and compaction.
- TSF capping construction.

14.3.2 Dewatering of RSF

Dewatering of the existing decant pond within the RSF 2 to remove as much water as reasonably practical will be required prior to construction. All dewatering works are to be undertaken in accordance with existing environmental release requirements.

The dewatering is designed to assist in reducing the saturation of the tailings, improve the workability of the material and reduce the risk of operating machinery bogging on the tailings. As such, an assessment of the bearing capacity of the material is required prior to accessing the area to ensure it is capable of safely accommodating construction equipment.

14.3.3 RSF Capping Construction

Once the tailings / fill material has been successfully dewatered enough as deemed suitable by the design engineer, the construction of the capping over the tailings can commence.

It is proposed that the capping will be constructed using material from the WRE, comprising of material won from mining of the satellite pit.

15. Instrumentation Design

This section outlines the proposed instrumentation to be installed as part of the RSF2 Stage 1 embankment to assist with the safety and performance monitoring of the RSF. The instrumentation for Stage 1 consists of Global Navigation Satellite Systems (GNSS) surface movement monuments distributed across the embankment crest, and AprilTags on the downstream face of the embankment for movement monitoring, pond level monitoring and vibrating wire piezometers across the foundation of the RSF for pressure monitoring.

15.1 Pond Level Indicator

The decant water level is currently monitored in RSF1 visually daily using markers to ensure a minimum beach, the decant level monitoring for RSF2 is proposed to consist of gauge boards installed at the decant finger monitoring the decant level.

15.2 Settlement Monitoring

A GNSS Survey arrangement and AprilTags are proposed to be installed along the crest and downstream face of the embankment as shown in the design drawings in Appendix A. A total of 11 GNSS units are proposed for installation along the crest of the dam to monitor the near real-time movement in three dimensions. A total of 120 AprilTags are proposed to be installed across each of the embankments to be surveyed on a monthly basis using photogrammetry to monitor large-scale movement across the embankment.

15.3 Vibrating Wire Piezometer

An arrangement of Vibrating Wire Piezometers (VWPs) is proposed to be installed at the upstream toe of each of the embankments as shown in the design drawings in Appendix A. A total of 12 VWPs are proposed for installation along the upstream to monitor the pore water through the tailings and the efficiency of the underdrainage system located around the upstream toe.

15.4 Monitoring Frequency

To ensure correct operation, adequate performance, and early identification of failure in the RSF, regular inspection of the facility is required during its operational life. Recommended Inspection Frequencies in accordance with a "**Significant**" Category Dam as outlined in ANCOLD (2019 – Guidelines on Tailings Dams) are shown in Table 15.1 and Monitoring Types and Frequencies as outlined in ANCOLD 2003 in Table 15.2.

Inspection Type	ANCOLD Recommended Frequency	GHD Recommended Frequency
Comprehensive	On First Filling, Every 2 Years	On First Filling, Every 2 Years
Intermediate	Annually to 2-Yearly	Annually
Routine Visual	Twice Weekly to Weekly	Twice Weekly
Special	As Required	As Required

 Table 15.1
 Inspection Types and Frequencies – Significant Consequence Category Dam

Note 1: Dam owners may undertake a review to determine if a reduced or increased frequency of inspection is acceptable. The review should be carried out by a dams engineer and take into account such matters as Regulator requirements, dam hazard, and risk, type and size of the dam, dam failure modes, and monitoring arrangements.

Table 15.2 Monitoring Types and Frequencies

Monitoring Type	ANCOLD Recommended Frequency	GHD Recommended Frequency
Rainfall	Twice Weekly to Weekly	Daily
Storage Level	Twice Weekly to Weekly	Telemetered - Near-Real-Time
Seepage	Twice Weekly to Weekly	Twice Weekly
Chemical Analysis of Seepage	Consider	Consider*
Pore Pressure	3-Monthly to 6-Monthly	Monthly
Surface Movement Control	5-Yearly to 10-Yearly	Telemetered - Near-Real-Time
Surface Movement Normal	2-Yearly	Telemetered - Near-Real-Time
Internal Movement / Stresses	2-Yearly	N/A
Seismology	Consider	Consider [#]
Post-Tensioning	5-Yearly to 10-Yearly	N/A

Note 1: The frequencies quoted assume manual reading of the instrumentation. Where automated readings are available more frequent reading would be appropriate. (TR- Telemetry recommended) (TC-Telemetry to be considered)

Note 2: Recommended annually for concrete dams, tailings dams, and embankments constructed from, or on, potentially dispersive materials were specified by the designer or safety reviewer

Note 3: The frequency of reading and location of the monitoring instruments need to be at the discretion of the dams engineer. Seismological instruments, where installed, are recommended to be incorporated into state-wide seismic networks.

Note 4: A control survey uses monuments that are remote from the dam site to check the location of the survey monuments at the dam site.

* Chemical analysis of the seepage should be considered on a case by case basis.

TGO should consider subscribing to an earthquake notification service given the risk posed by liquefaction of the tailings.

16. Safety in Design

16.1 Overview

Safety in design is a strategy aimed at preventing injuries by considering hazards as early as possible in the planning and design process, enhancing safety through choices in the design process. A safety in design approach considers the safety of those who construct, operate, maintain, clean, repair, and demolish an asset (includes building, structure, plant, or equipment). Parties involved in the planning and design stage of a project are in a position to reduce the risks that arise during the life cycle of the asset and have a legal requirement to do so.

At each design stage "designers" can make a significant contribution by identifying and eliminating hazards, and reducing likely risks from hazards where elimination is not possible. Often the most cost-effective and practical approach is to avoid introducing a hazard to the workplace in the first place, by eliminating hazards at the design stage.

The definition of "designer" not only affects the actual designer but also those who are connected with the design (e.g. during construction), including parties where the end product is to be used, or could reasonably be expected to be used, as, or at a workplace (e.g. during end use, inspection, operation, cleaning, maintenance, and closure/demolition). Furthermore, the "designer" must ensure, so far as is reasonably practicable, that the plant, substance, or structure is designed to minimize risks to the health and safety of workers where the design is for the purposes of a workplace.

It is, therefore, reasonable to consider the wider practical definition of "designer" to include:

- Design professionals.
- Head contractors, project managers, clients, end-users, and workers.
- Quantity surveyors, insurers, quality assurance staff, work safety professionals, and ergonomics practitioners.
- Suppliers including manufacturers, importers, those who hire plant, constructors, installers, and trades and maintenance, people.

GHD has been engaged to provide the design services described in this report. As such GHD has undertaken a component of the designer's role in this project. In this role, GHD has identified and mitigated a number of potential risks within the limitations of our scope, in consultation with other members of the design team.

During the detailed design phase, a Safety in Design review was carried out by GHD. A number of design improvements were identified to eliminate hazards and improve overall safety. GHD has prepared a summary of the risks identified and mitigation measures adopted or recommended. The risk register is included in Appendix F.

Some residual risks, both safety, and design related have been identified in the risk register. Proposed mitigation measures and responsible parties for implementing these are also identified.

GHD formally hands the responsibility of the residual risks to TGO for further mitigation and trusts that you will complete the safety in design review process for the phases of this project within your responsibility. The safety in design document should be provided to each of the parties who may be identified as being able to influence design. The safety in design risk assessment should be continually updated to reflect the current risks associated with all current activities associated with the asset.

16.2 Stakeholders

The key stakeholders involved in the design, construction and operation of the RSF 2 Stage 1 embankment project are:

- GHD Designer.
- TGO Owner.
- TBC Contractor.

17. Recommendations

The following recommendations are summarised from the report relevant to the RSF2 Stage 1 design to provide guidance for construction, operational phase and closure planning:

Table 17.1	Summary of Recommendations
------------	----------------------------

No.	Recommendations	Section Reference
1	ANCOLD (2019b) also recommends 1:10,000 AEP Maximum Credible Earthquake (MCE) be used in the design of the closure arrangement.	Section 3.6.1
2	It is recommended that shallow monitoring bores be installed around the outside of the embankment of RSF 2 to detect seepage through the foundation. Monitoring bores should be installed to depths of 10 m below ground level (bgl) and at an interval of approximately 250 m around the perimeter of the RSF. Bores should be screened between 2 m depth and the base of the bore.	Section 9.4
3	It is recommended that these models be reviewed with each lift to assess actual settlement with predicted settlement to confirm the effectiveness of the connection drainage system.	Section 11.5.4
4	It is recommended a detailed closure plan for the RSF1 and RSF2 storages be prepared following the initiation of construction of the RSF1 Stage 8 embankment raise to detail the closure arrangement, planning and operation requirements for the two facilities.	Section 14.2
5	TGO should consider subscribing to an earthquake notification service given the risk posed by liquefaction of the tailings.	Section 15.4

18. References

ANCOLD (2012), Guidelines on the Consequence Categories for Dams, (ANCOLD), 2012.

ANCOLD (2019a) Guidelines on Tailings Dams, (ANCOLD), 2019.

ANCOLD (2019b) Guidelines for Design of Dams and Appurtenant Structures for Earthquake (ANCOLD), 2019.

Ball J, Babister M, Nathan R, Weeks W, Weinmann E, Retallick M, Testoni I, (Editors) Australian Rainfall and Runoff: A Guide to Flood Estimation, © Commonwealth of Australia (Geoscience Australia), 2019.

Coffey Geotechnics (2007) Preliminary Groundwater Investigation, Alkane Exploration Limited, Tomingley NSW

DE Cooper and Associates Pty Ltd (2011) Tomingley Gold Project, Residue Management, Design Report, Rev 1, August 2011

GHD (2020) TGO RSF 2 Concept Design Report, GHD, November 2020

GHD (2021) Tomingley Gold Operations TGO RSF 2 Geotechnical investigation, Geotechnical Factual Report, August 2021

Impax Group (2011) Tomingley Gold Operations Groundwater Assessment

Jacobs (2021) Tomingley Gold Extension Project Groundwater Assessment, Jacobs Australia Pty Limited, September 2021.

Appendices

Appendix A RSF2 Detailed Design Drawings

TOMINGLEY GOLD MINE RSF 2 STAGE 1 DETAILED DESIGN 12545239

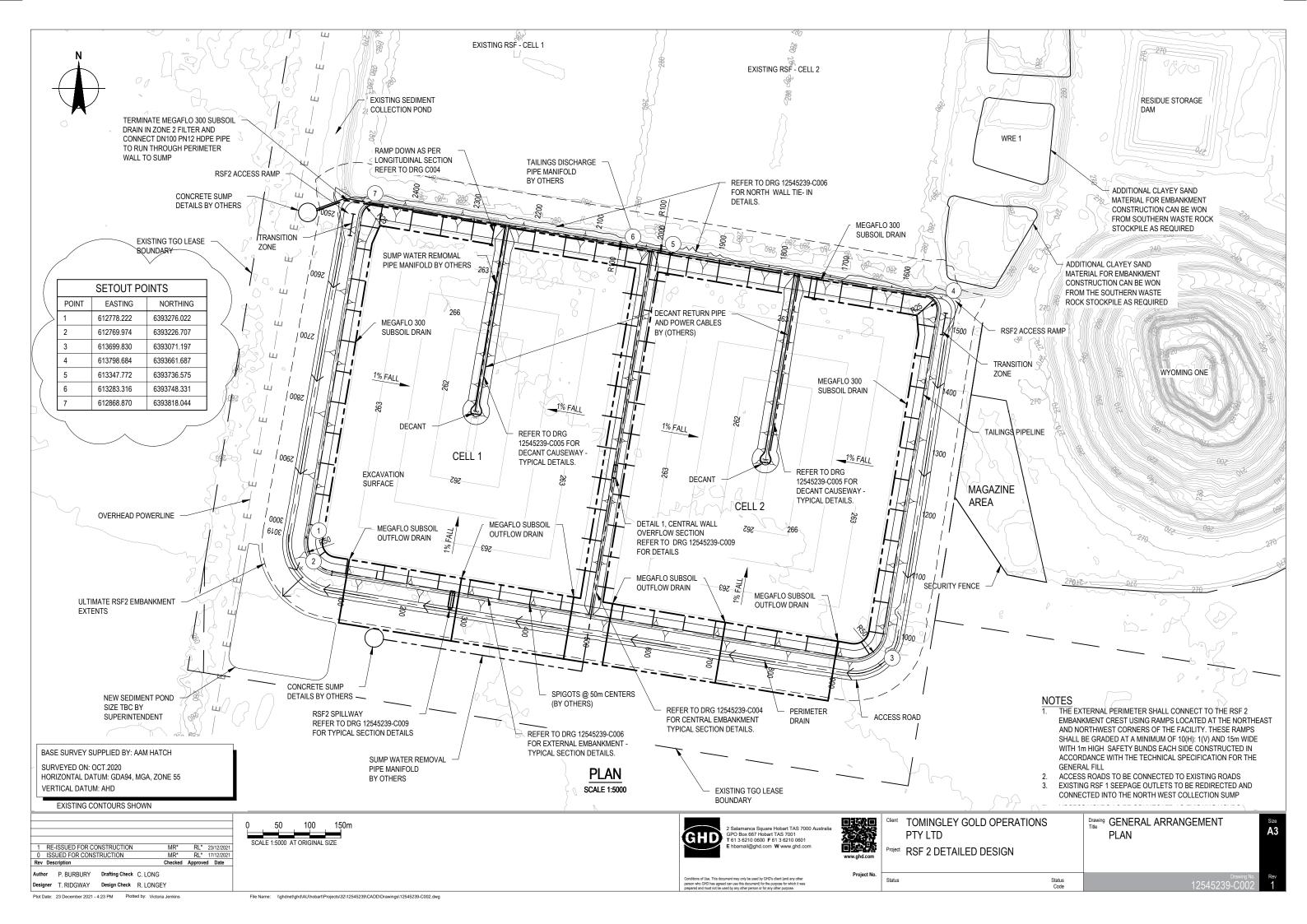


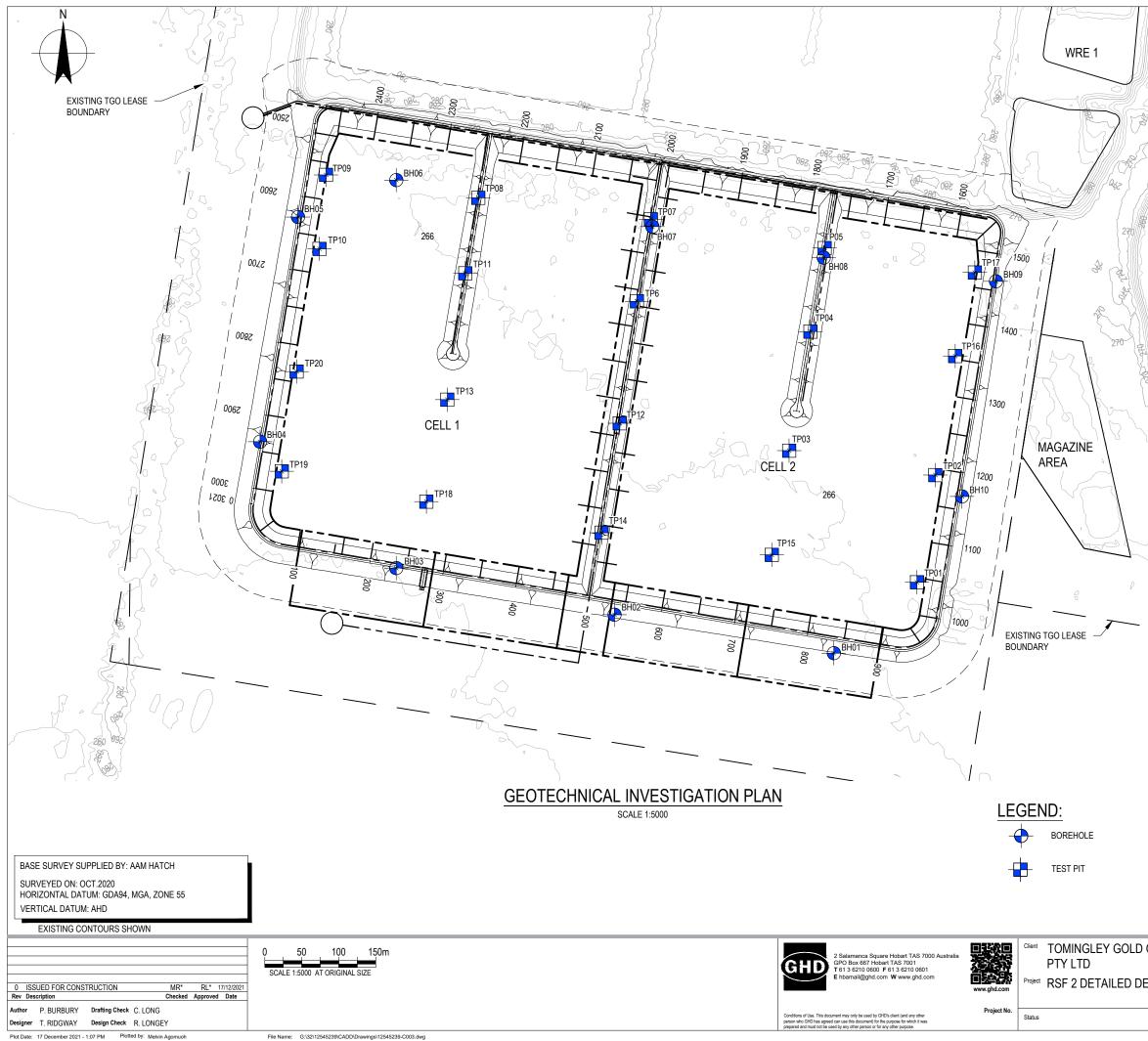


DRAWING TITLE

- COVER SHEET AND DRAWING LIST
- GENERAL ARRANGEMENT PLAN
- GEOTECHNICAL INVESTIGATION PLAN
- TYPICAL EMBANKMENT SECTIONS AND DETAILS SHEET 1 OF 4
- TYPICAL EMBANKMENT SECTIONS AND DETAILS SHEET 2 OF 4
- TYPICAL SECTIONS SHEET 3 OF 4
- TYPICAL SECTIONS SHEET 4 OF 4
- MATERIALS SPECIFICATION
- TYPICAL SPILLWAY SECTION AND DETAIL
- INSTRUMENTATION PLAN
- GPS INSTRUMENTATION TYPICAL SECTIONS
- GNSS GPS UNIT DETAILS
- SURVEY REFERENCE APRIL TAG MOUNT
- STAND ALONE APRIL TAG MOUNT DETAILS
- LONGITUDINAL SECTIONS SHEET 1 OF 5
- LONGITUDINAL SECTIONS SHEET 2 OF 5
- LONGITUDINAL SECTIONS SHEET 3 OF 5
- LONGITUDINAL SECTIONS SHEET 4 OF 5
- LONGITUDINAL SECTIONS SHEET 5 OF 5
- ULTIMATE TYPICAL SECTIONS
- CONCEPTUAL CLOSURE PLAN
- CONCEPTUAL CLOSURE SECTION AND DETAILS

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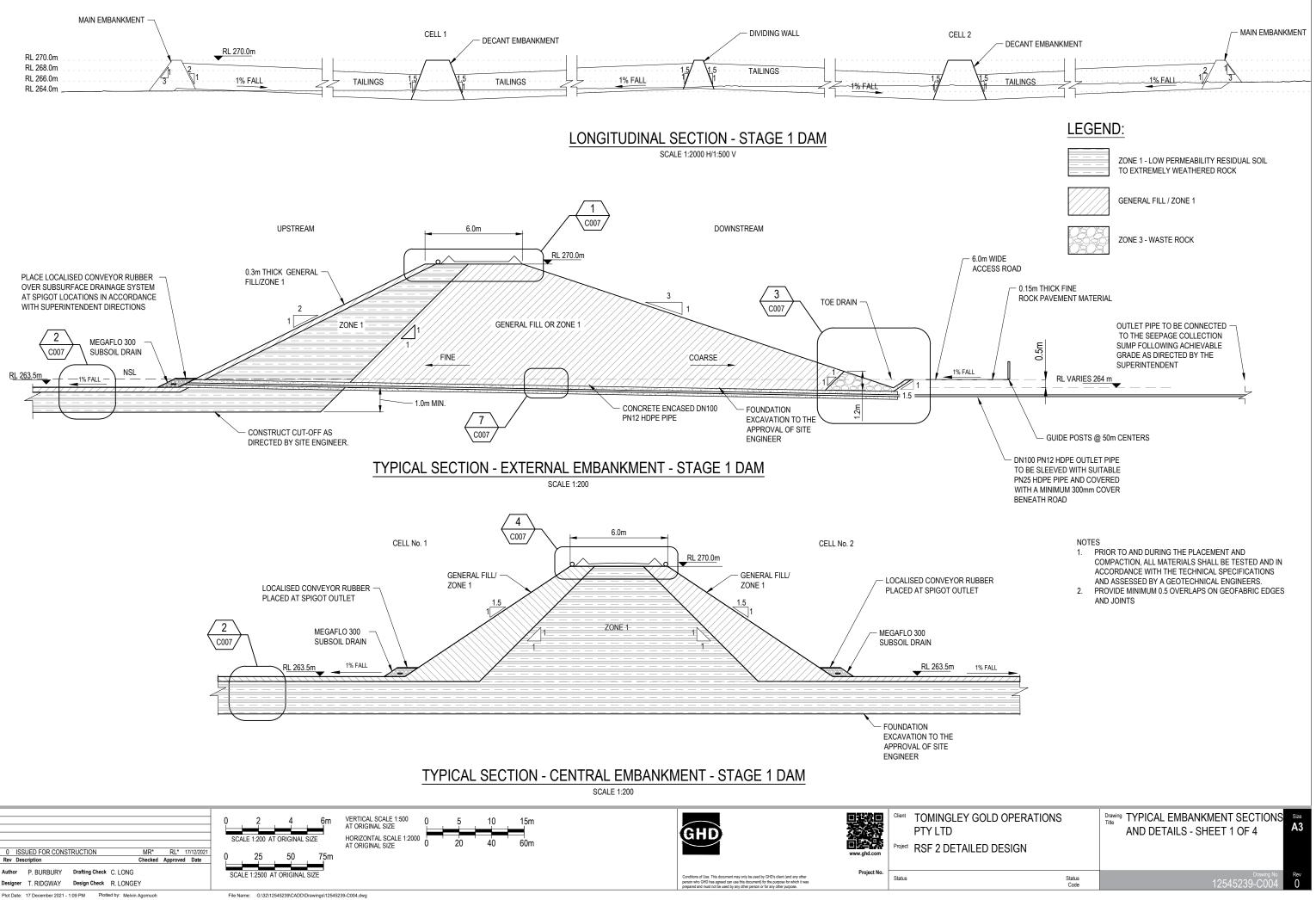


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TP03	613505.900	6393348.000	
TP04	613534.700	6393509.000	
TP05	613554.000	6393622.000	
TP6	613299.900	6393550.000	
TP07	613318.800	6393661.000	
TP08	613085.400	6393690.000	
TP09	612879.500	6393721.000	
TP10	612870.500	6393621.000	
TP11	613067.700	6393588.000	
TP12	613276.800	6393385.000	
TP13	613043.500	6393417.000	
TP14	613252.200	6393237.000	
TP15	613482.500	6393207.000	
TP16	613730.600	6393476.000	
TP17	613757.100	6393589.000	
TP18	613015.100	6393279.000	
TP19	612819.500	6393320.000	
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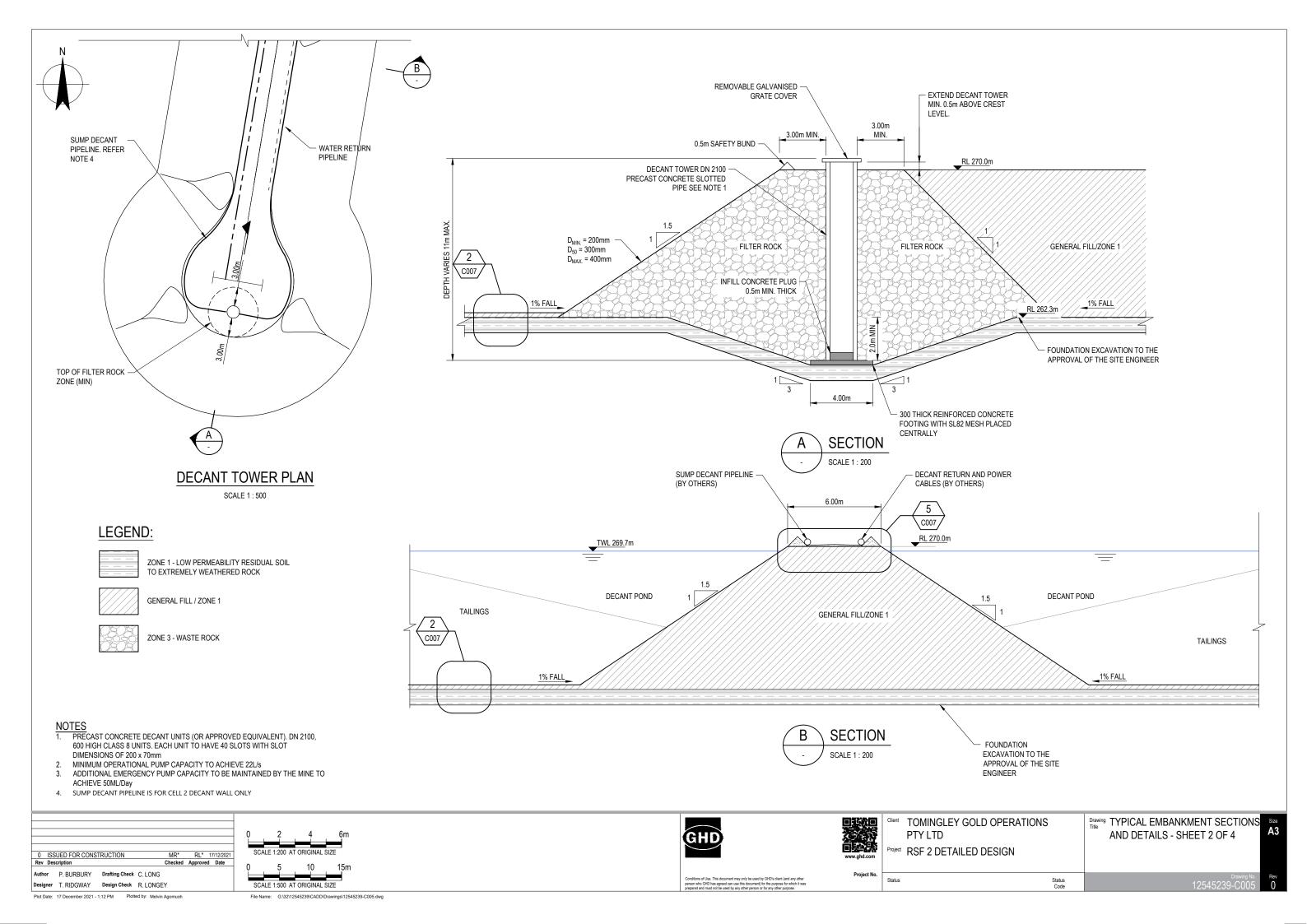
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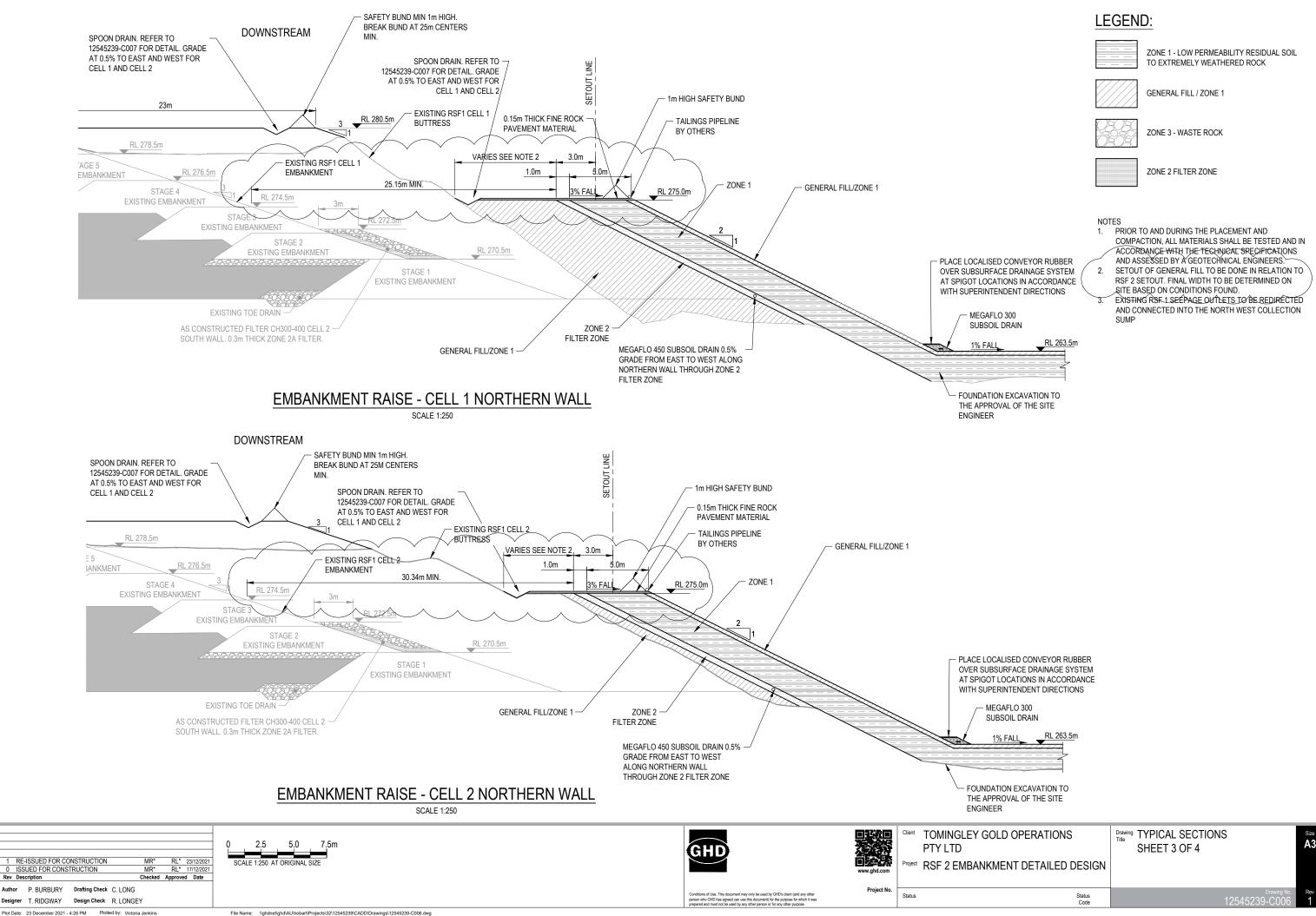
BOREHOLE SETOUT POINTS		
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BH04	612790.100	6393360.000
BH05	612841.400	6393664.000
BH06	612974.400	6393714.000
BH07	613320.900	6393651.000
BH08	613552.700	6393609.000
BH09	613785.600	6393577.000
BH10	613739.800	6393286.000

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DESIGN		
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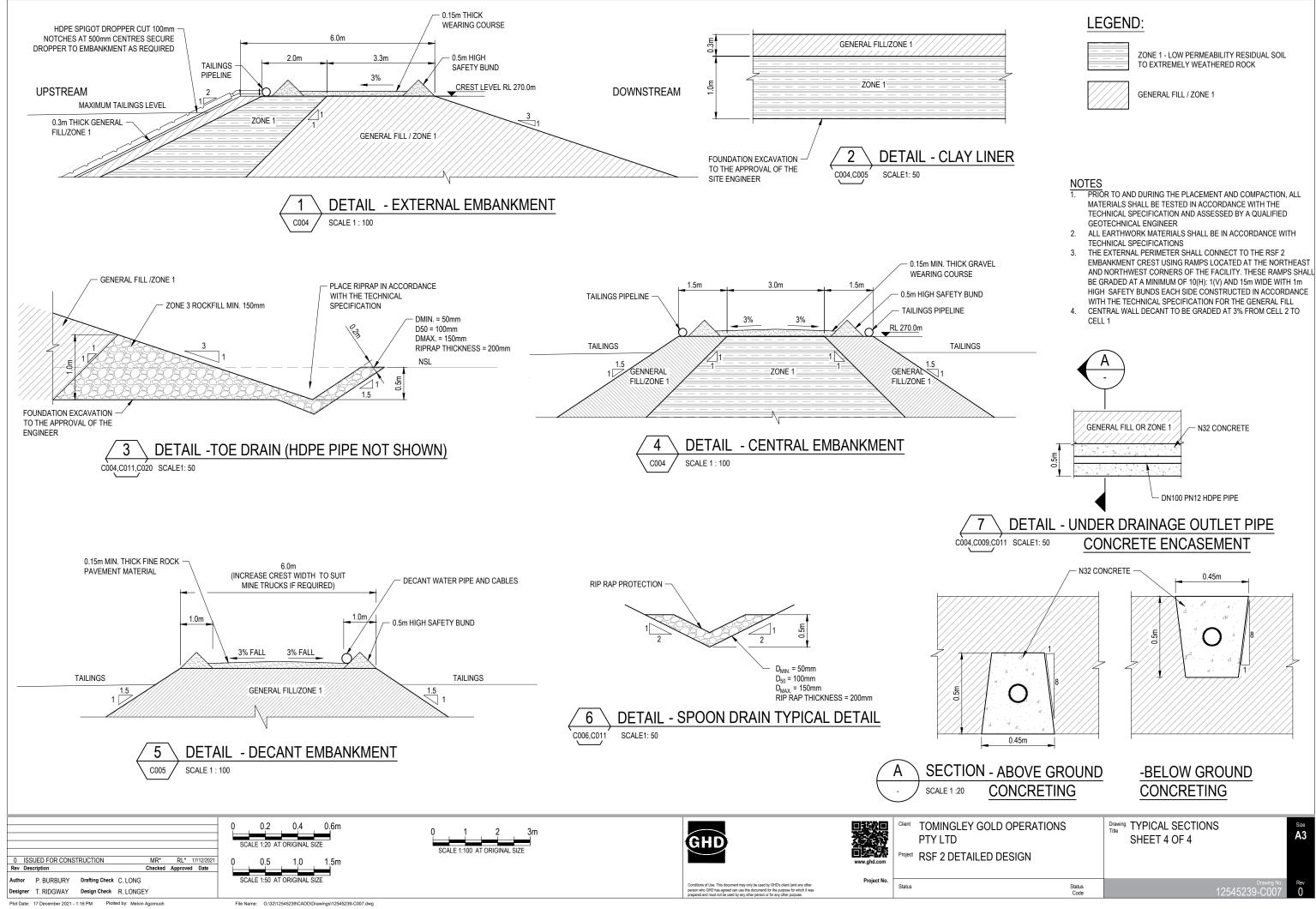








OPERATIONS	Drawing TYPICAL SECTIONS SHEET 3 OF 4	Size A3
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	Drawing No.	Rev







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TABLE 1 - MATERIAL CONSTRUCTION REQUIREMENTS

MATERIAL	DESCRIPTION	CONSTRUCTION REQUIREMENTS				
ZONE 1	EXTREMELY WEATHERED SANDY CLAY OR CLAYEY SAND MATERIAL WON FROM THE FOUNDATION EXCAVATIONS, MEETING SPECIFICATIONS SETOUT IN TABLE 2.	 SPREAD IN LAYERS 150mm THICK AFTER COMPACTION. COMPACT WITH MINIMUM 6 PASSES OF A 10 TONNE VIBRATING PAD FOOT ROLLER TO NO LESS THAN 98% MAX. DRY DENSITY (MDD). SANDY CLAY TO BE PLACED BETWEEN 1% DRY & 3% WET OF OPTIMUM MOISTURE CONTENT (OMC) 				
ZONE 1/GENERAL FILL	EXTREMELY WEATHERED SANDY CLAY OR CLAYEY SAND TO GRAVELLY CLAY MATERIAL WON FROM THE FOUNDATION EXCAVATIONS, MEETING SPECIFICATIONS SET OUT IN TABLE 2.	 SPREAD IN LAYERS 300mm THICK AFTER COMPACTION COMPACT WITH MINIMUM 6 PASSES OF A 10 TONNE VIBRATING PAD FOOT ROLLER NO LESS THAN 95% MAX. DRY DENSITY (MDD). SANDY CLAY TO BE PLACED BETWEEN 1% DRY AND 3% WET OF OPTIMUM MOISTURE CONTENT (OMC) 				
ZONE 2	SAND MATERIAL SOURCED FROM CRUSHED ROCK IN ACCORDANCE WITH THE MATERIAL SPECIFICATIONS SET OUT IN SECTION 6.3 OF THE TECHNICAL SPECIFICATION.	FILTER SAND TO BE PLACED IN ACCORDANCE WITH THE SPECIFICATION SET OUT IN SECTION 6.3 OF THE TECHNICAL SPECIFICATION.				

TABLE 2 - GHD SPECIFICATION

TABLE 3 - FOUNDATION PREPARATION

MATERIAL

STORAGE FOUNDATION

TABLE 4 - MATERIAL TESTING REQUIREMENTS FOR SANDY CLAY/ CLAYEY SAND

REQUIRED PROPERTIES OF ZONE 1 FILL AFTER COMPACTION MINIMUM % BY WEIGHT PASSING 0.075mm AS 1152 SIEVE 30 MINIMUM % BY WEIGHT PASSING 0.3mm AS 1152 SIEVE 70 MINIMUM % BY WEIGHT PASSING 1.18mm AS 1152 SIEVE 80 MINIMUM PLASTICITY INDEX (%) (AS 1289.3.3.1) 15 LAYER THICKNESS 150mm PERMEABILITY K < 1 X 10⁽⁻⁹⁾ m's (SATURATED) REQUIRED PROPERTIES OF GENERAL FILL AFTER COMPACTION 20 (REFER NOTE 2) MINIMUM % BY WEIGHT PASSING 0.075mm AS 1152 SIEVE MINIMUM % BY WEIGHT PASSING 0.3mm AS 1152 SIEVE 40 MINIMUM % BY WEIGHT PASSING 1.18mm AS 1152 SIEVE 60 LAYER THICKNESS 300mm PERMEABILITY K < 1 X 10⁽⁻⁷⁾ m/s (SATURATED). (REFER NOTE 2).

TEST DETAILS FIELD DENSITY & HILF COMPACTION OR STANDARD COMPACTION ATTERBERG LIMITS PARTICLE SIZE DISTRIBUTION EMERSON CLASS TESTING - PROCESS WATER EMERSON CLASS TESTING - FRESH WATER TRIAXIAL PERMEABILITY EMBANKMENT FOUNDATIO

PUSH TUBE SAMPLING AND TRIAXIAL TESTING

NOTE 2: IF MATERIALS MEET THE REQUIRED PERMEABILITY SPECIFICATION THEN THIS CAN SUPERCEDE THE REQUIRED GRADING SPECIFICATION.

	0 ISSUED FOR CONSTRUCTION MR* RL* 17/12/2021 Rev Description Checked Approved Date	2 Salamanca Square Hobart TAS 7000 Australia GPO Box 667 Hobart TAS 7001 T 61 3 6210 0600 F 61 3 6210 0601 E hbamail@ghd.com W www.ghd.com	WWW.ghd.com	Client TOMINGLEY GOLD OPERATIONS PTY LTD Project RSF 2 DETAILED DESIGN	Drawing MATERIALS SPECIFICATION
Designer T. RIDGWAY Design Check R. LONGEY Status Status Status Code 12545239-C	Author P. BURBURY Drafting Check C. LONG Designer T. RIDGWAY Design Check R. LONGEY		Project No.	Status Status Code	Drawing N 12545239-C00

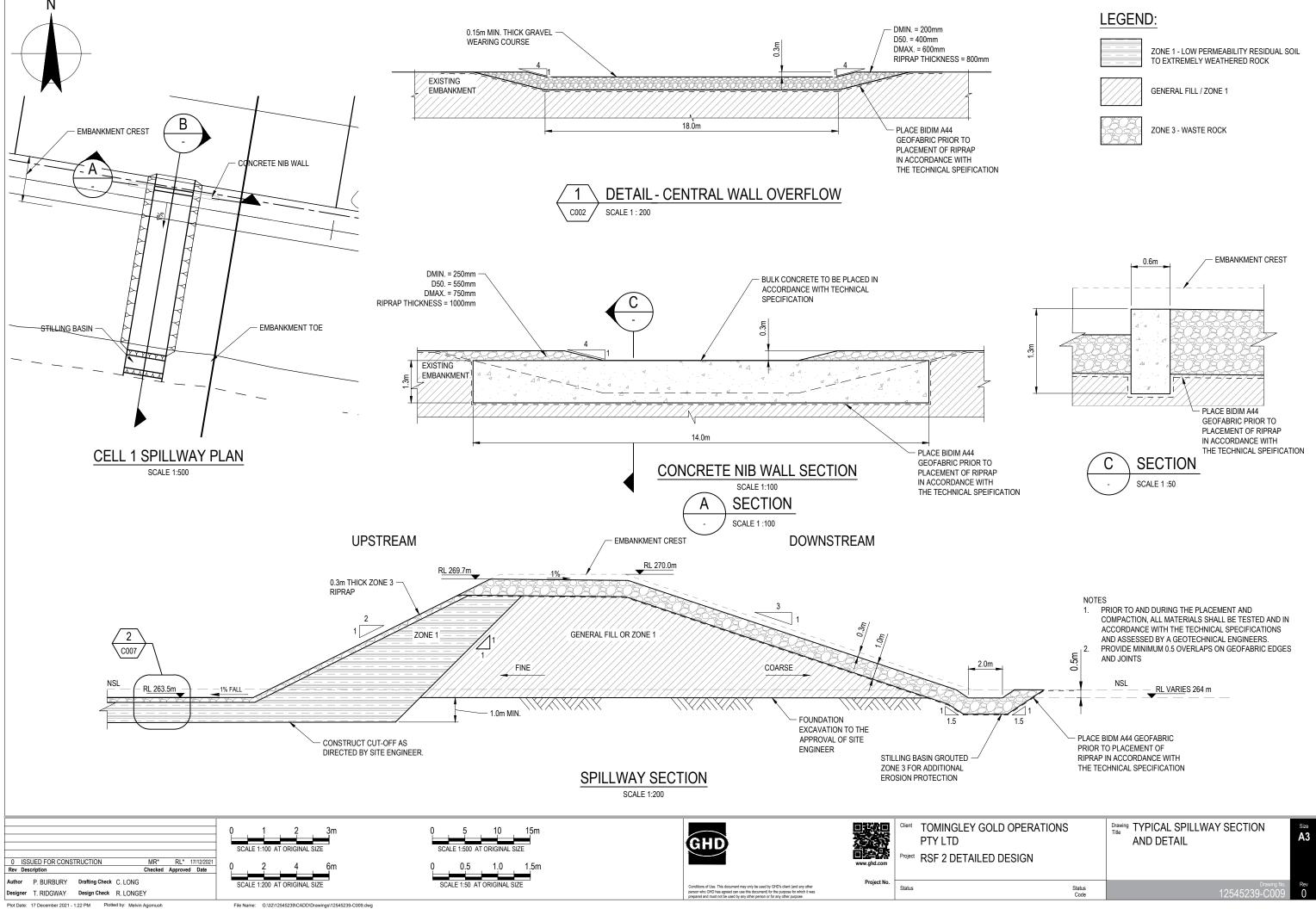
REQUIREMENTS

STORAGE FOUNDATION TO BE PREPARED IN ACCORDANCE WITH THE TECHNICAL SPECIFICATION

MINIMUM TEST FREQUENCY/ VOLUME PLACE (m³)

	EVERY 300mm AND 1,000m ³ /5,000m ³)
	1,000m ³ /5,000m ³
	1,000m ³ /5,000m ³
	5,000m ³ /10,000m ³
	5,000m ³ /10,000m ³
	10,000m ³ /20,000m ³
	5,000m ³ /10,000m ³
ON	TESTING REQUIREMENTS

1 EVERY 500m

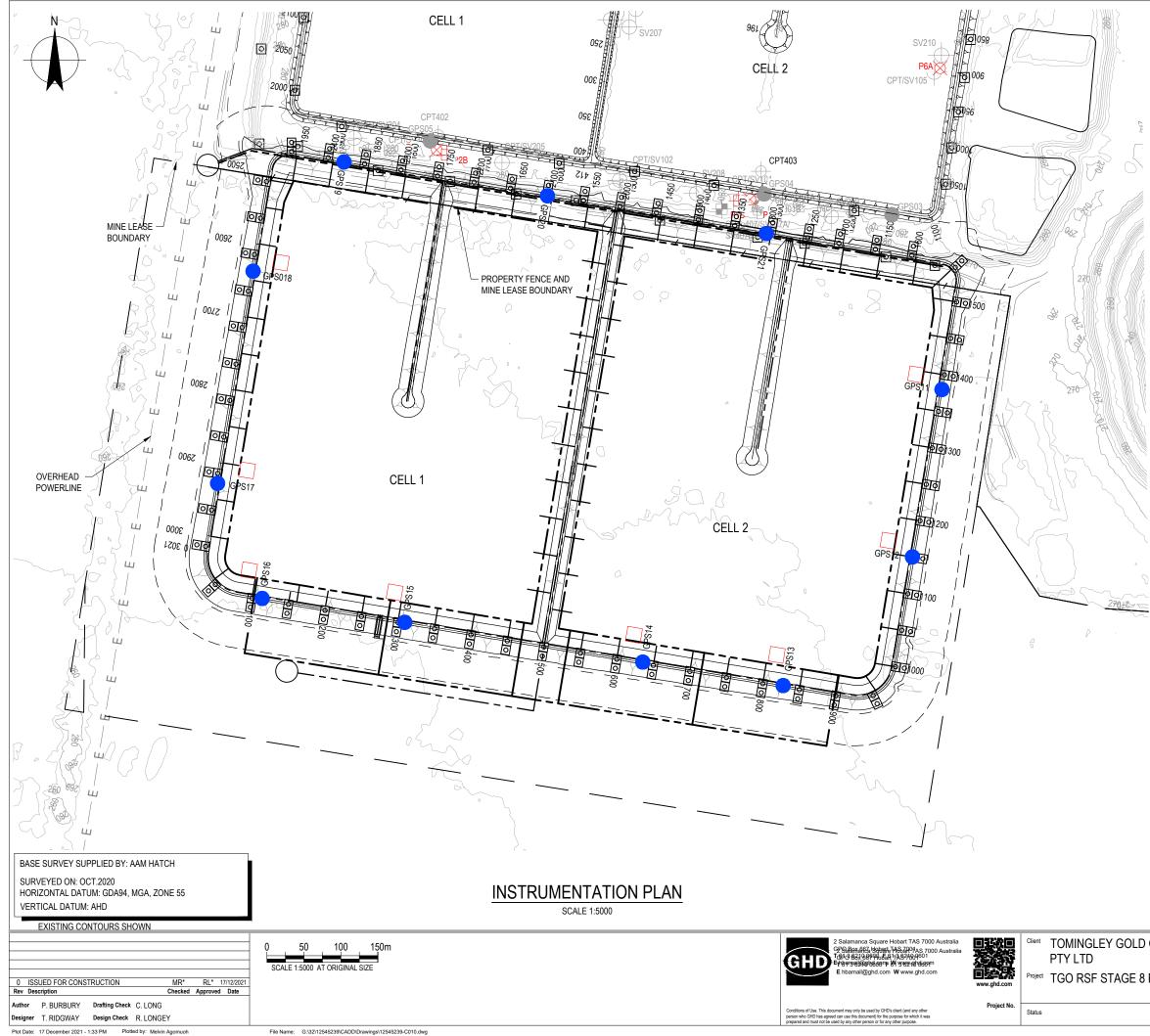


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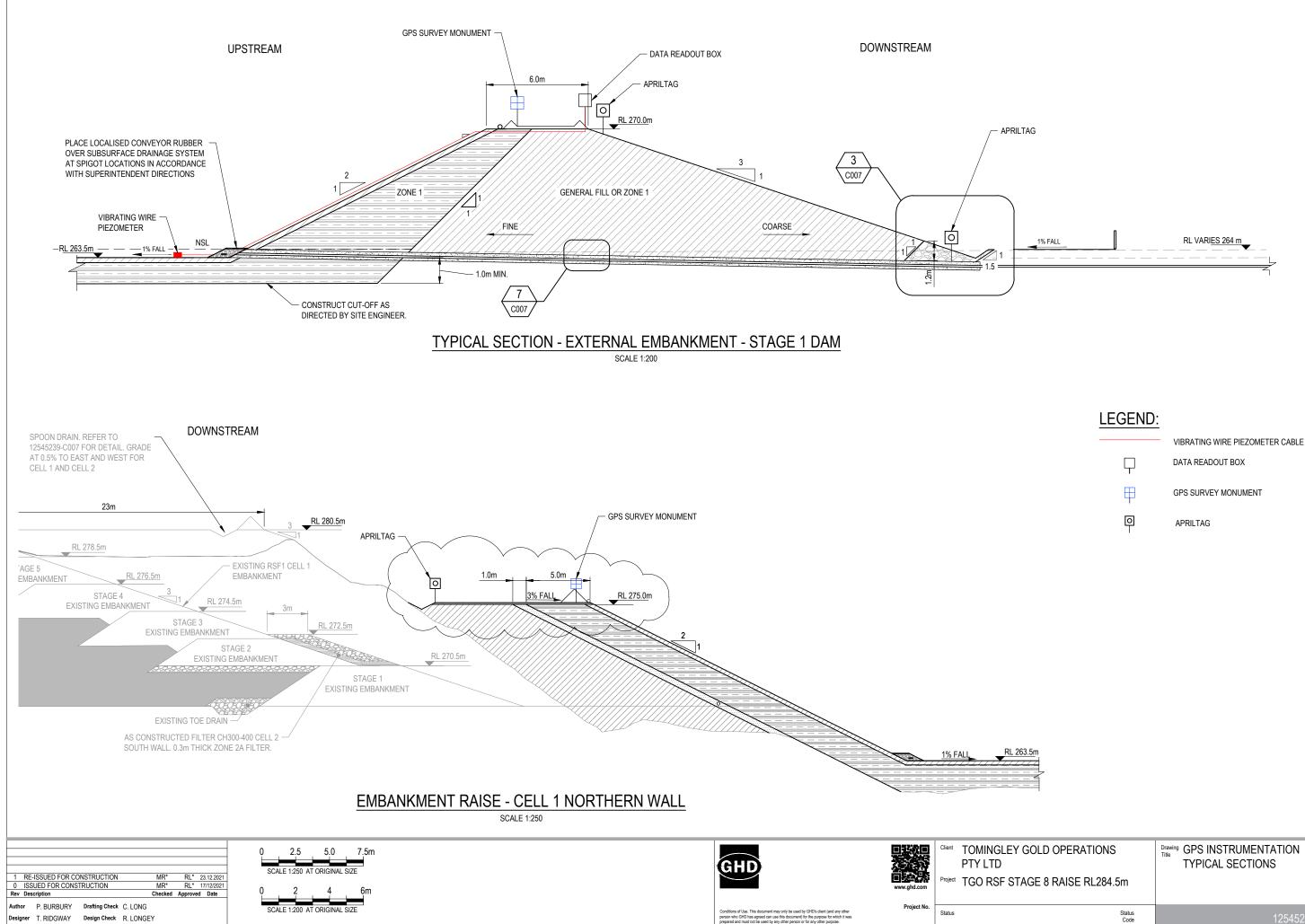
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GPS15	613046.595	6393180.445				
GPS16	612853.942	6393212.664				
GPS17	612793.650	6393368.276				
GPS018	612841.610	6393655.048				
GPS19	612964.450	6393802.685				
GPS20	613239.498	6393756.948				
GPS21	613535.455	6393705.913				

LEGEND:

	VIBRATING WIRE PIEZOMETER CABLE
\boxtimes	SET A PIEZOMETERS - PUSH IN FROM STAGE 4 CREST
+	SET B PIEZOMETERS - PUSH IN FROM BUTTRESS
0	APRILTAG LOCATION
GPSXX	GPS SURVEY MONUMENT - NEW
GPSXX	GPS SURVEY MONUMENT - EXISTING
	VIBRATING WIRE PIEZOMETER

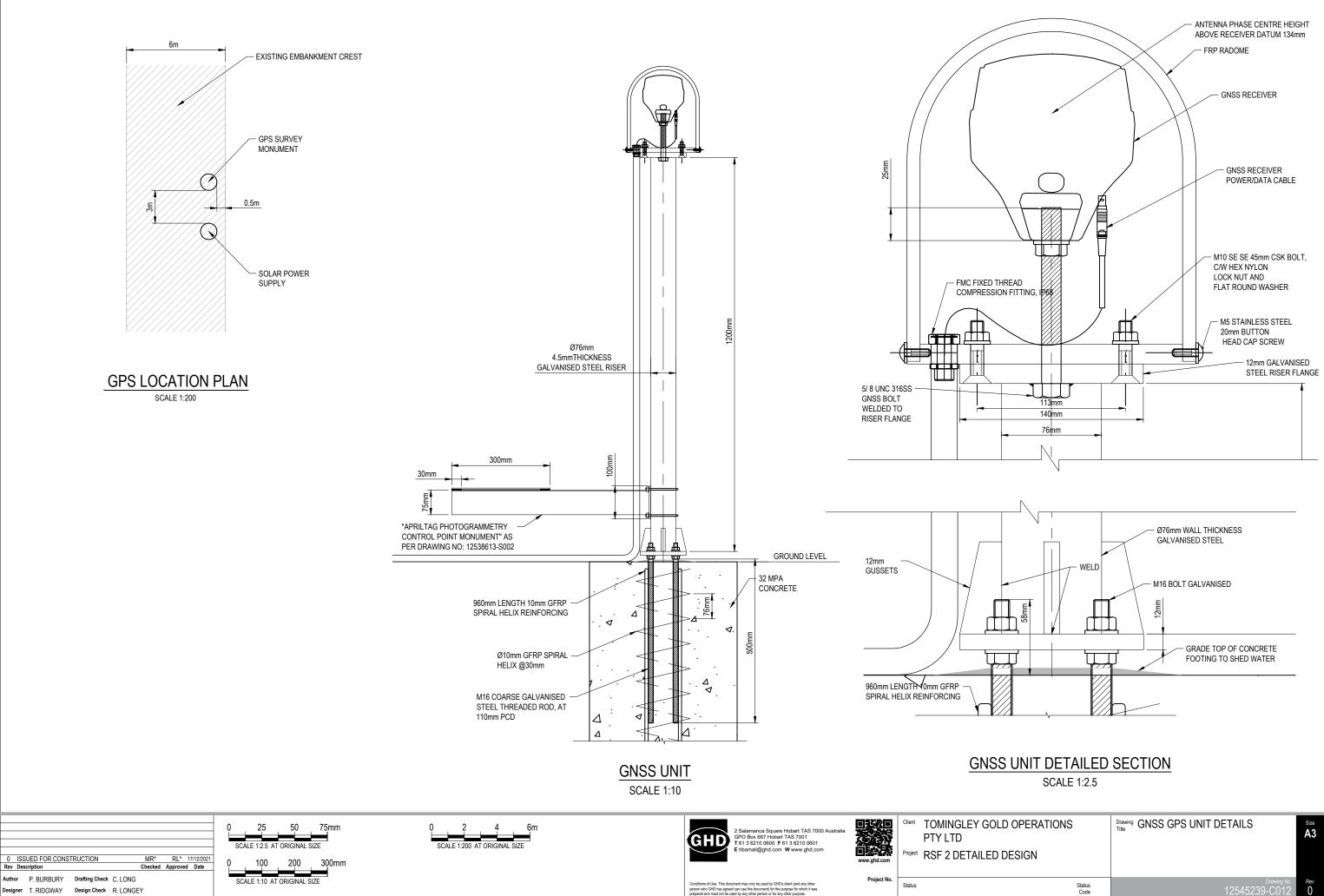
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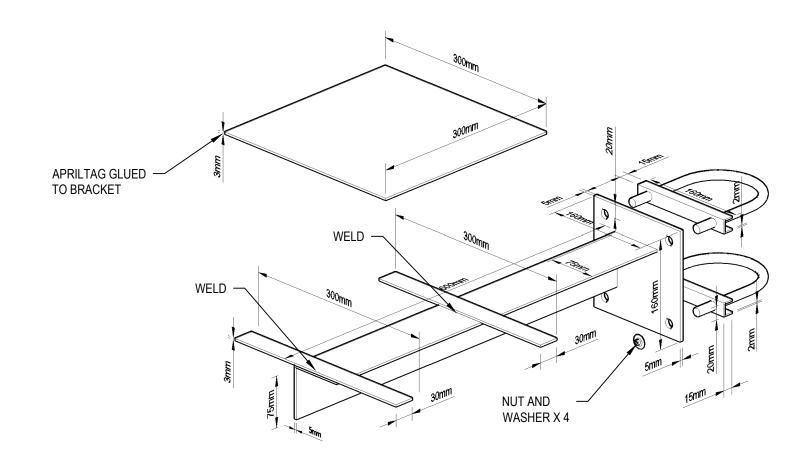
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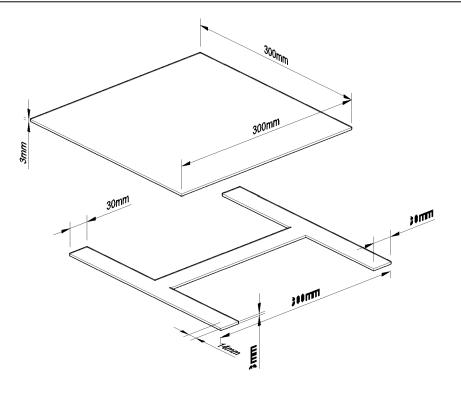


SURVEY REFERENCE APRIL TAG MOUNT

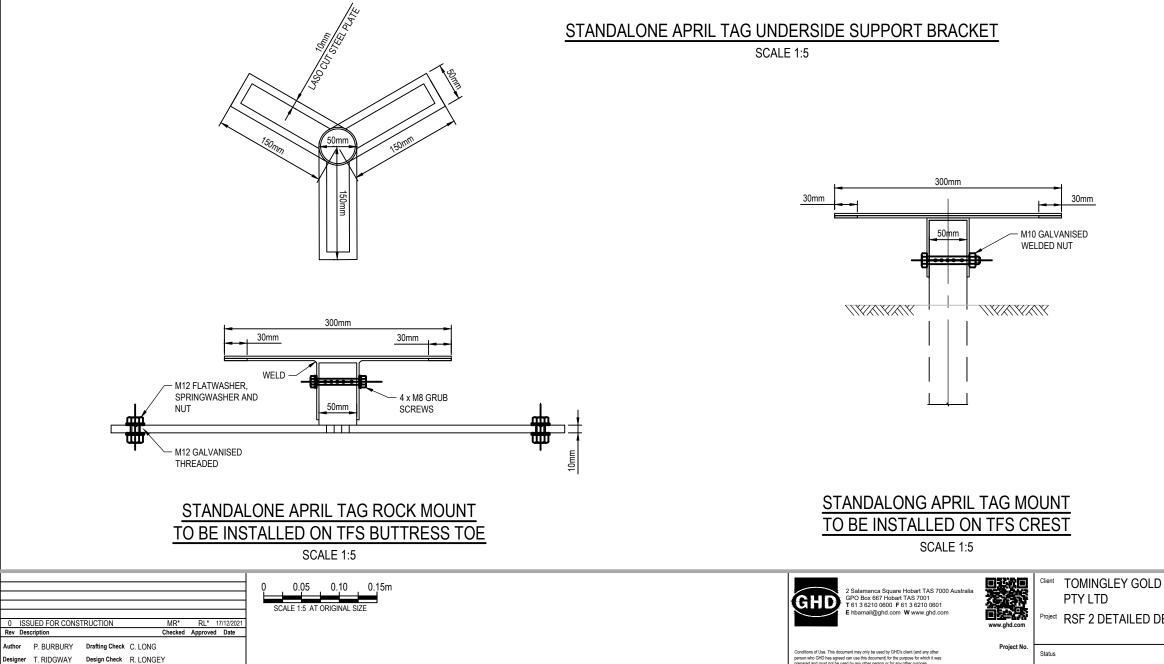
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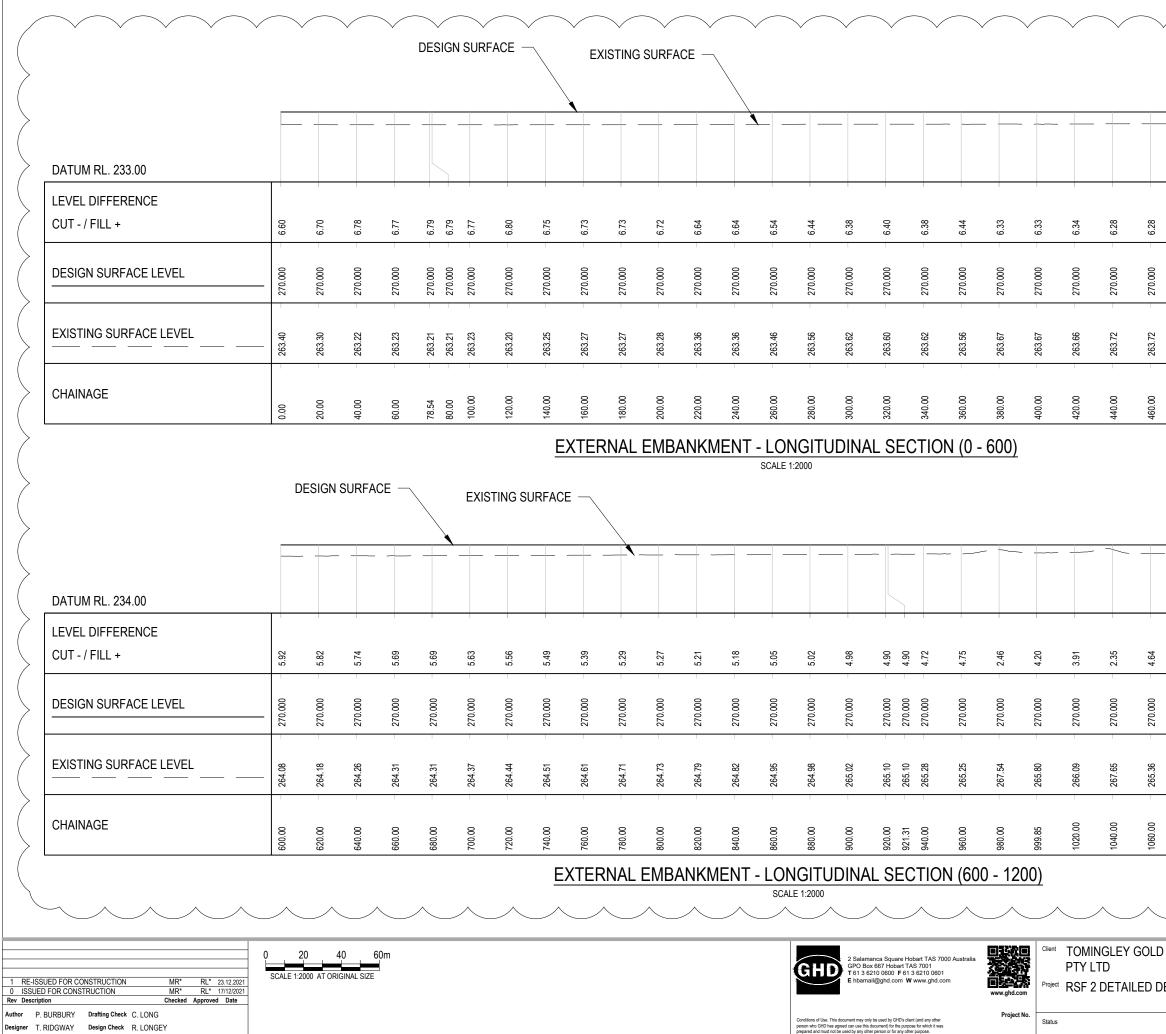




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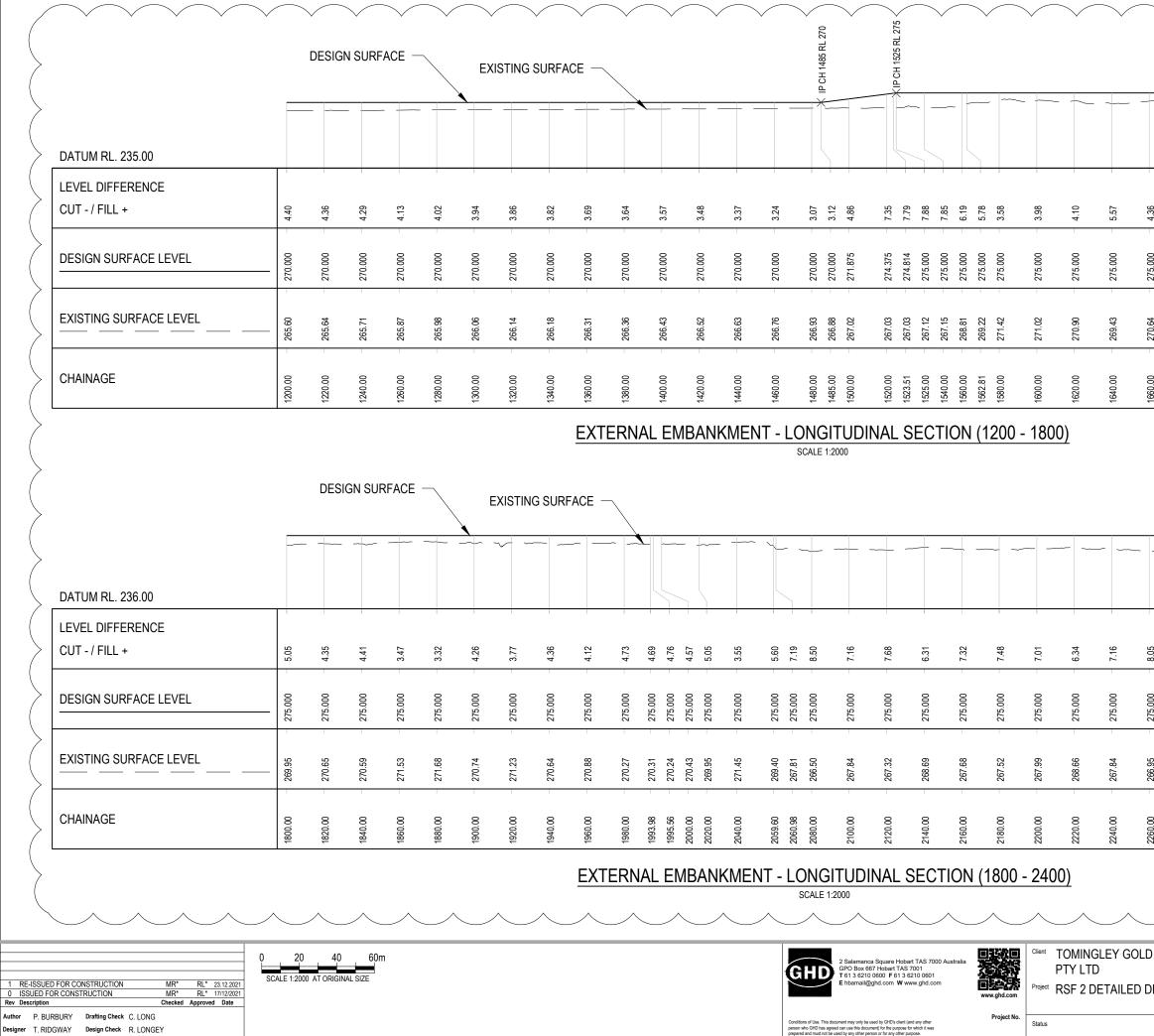
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c c	0.23	6.19	5.85	6.04	6.00	5.98	5.92	$\left\{ \right\}$		
	2/0.000	270.000	270.000	270.000	270.000	270.000	270.000	$\left\langle \right\rangle$		
		263.81	264.15	263.96	264.00	264.02	264.08			
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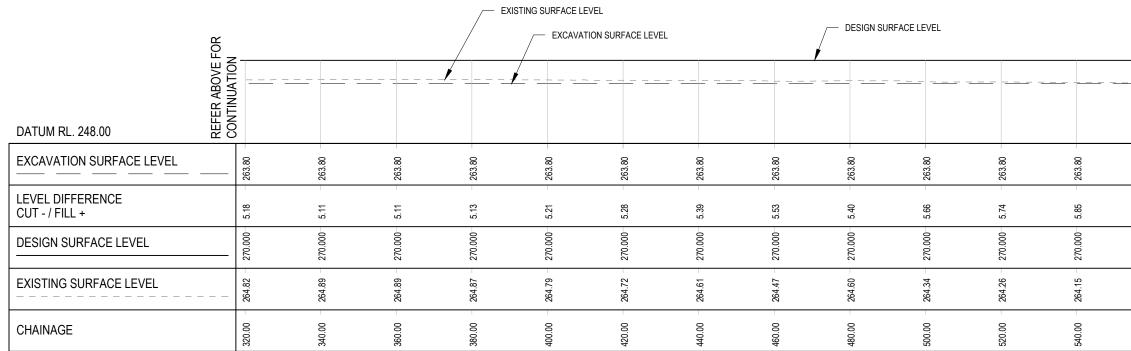
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DESIGN SURFACE LEVEL	275.000	272.500	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000
EXISTING SURFACE LEVEL	268.40	265.30	265.63	265.60	265.57	265.45	265.35	265.22	265.16	265.06	265.05	264.99
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DIVIDING WALL EMBANKMENT - LONGITUDINAL SECTION

SCALE 1:1000



DIVIDING WALL EMBANKMENT - LONGITUDINAL SECTION (CONTINUED)

SCALE 1:1000

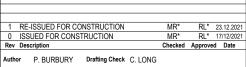
GHD

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Project No.

2 Salamanca Square Hobart TAS 7000 Australia GPO Box 667 Hobart TAS 7001 T 61 3 6210 0600 F 61 3 6210 0601 E hbamail@ghd.com W www.ghd.com

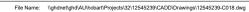
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Designer T. RIDGWAY Design Check R. LONGEY

Plot Date: 23 December 2021 - 4:40 PM Plotted by: Victoria Jenkins



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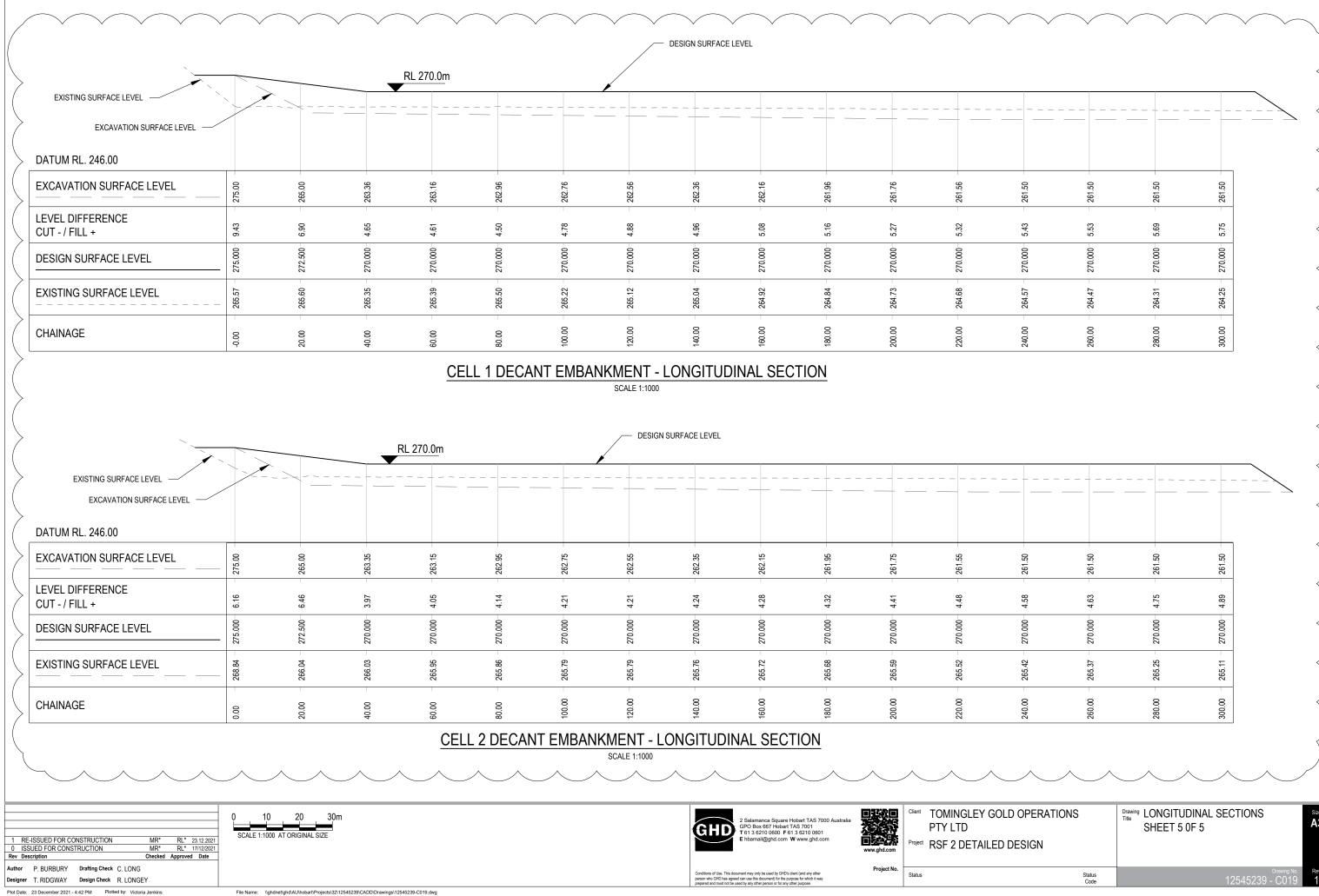
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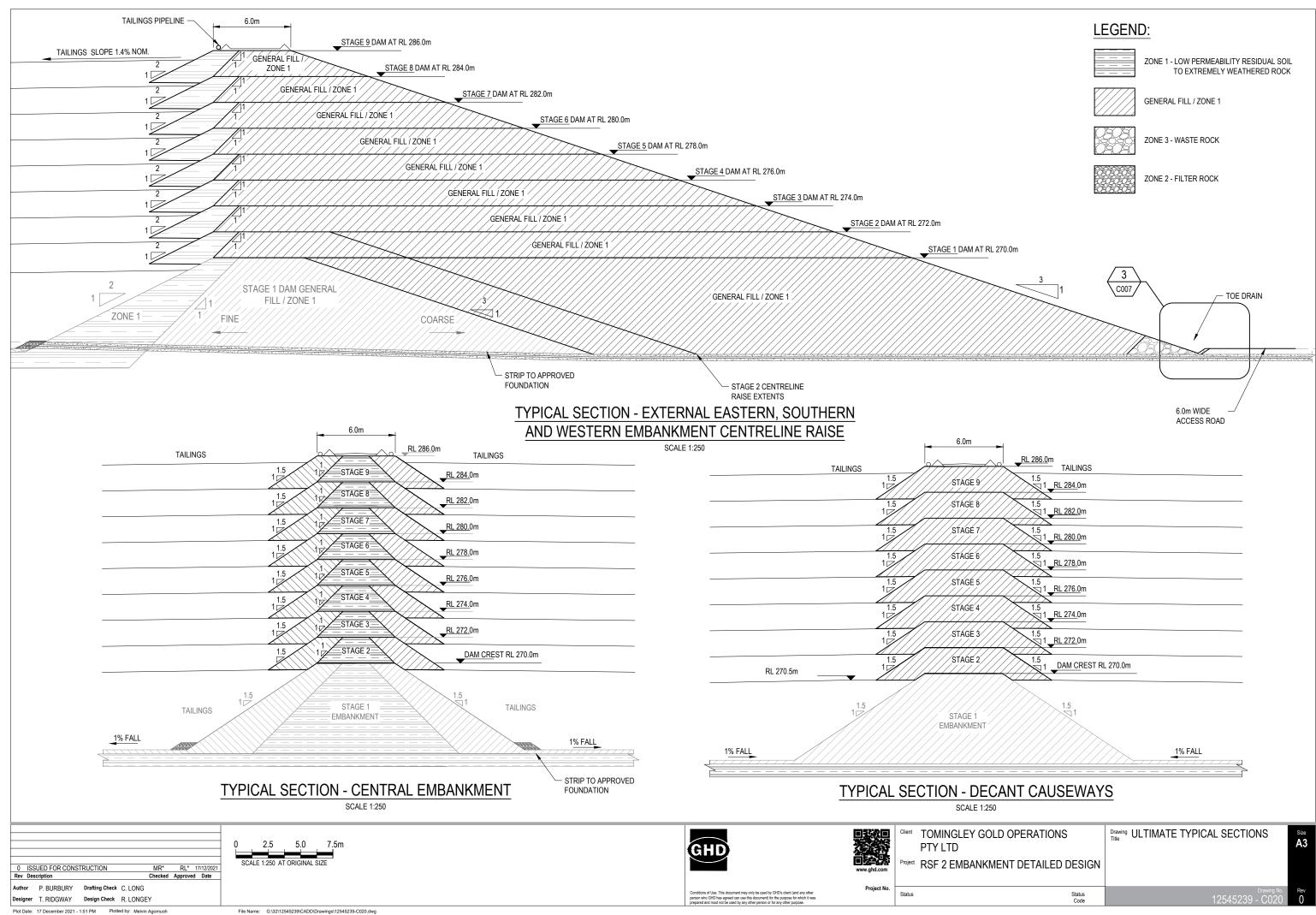


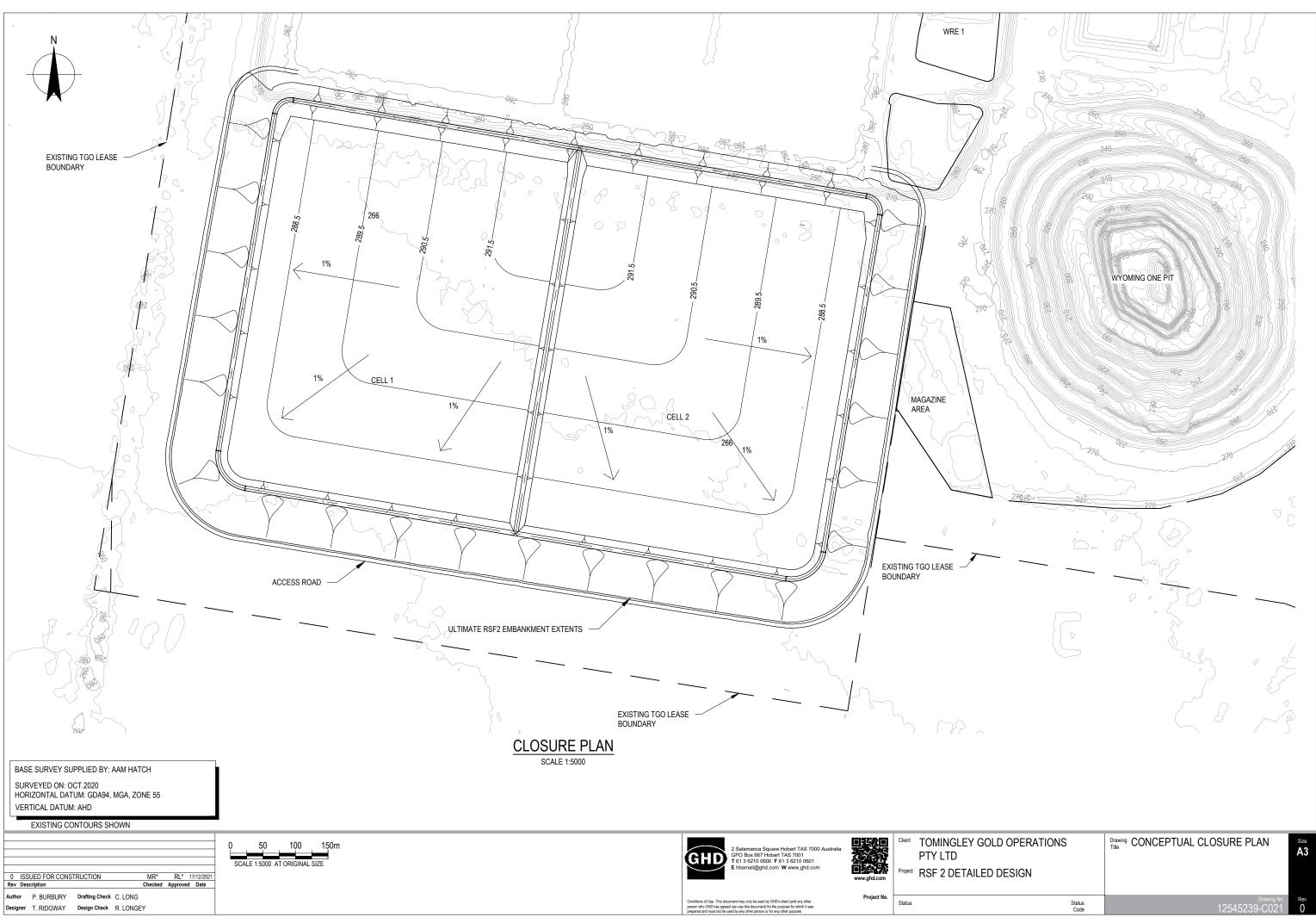
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240.00	260.00	280.00	300.00



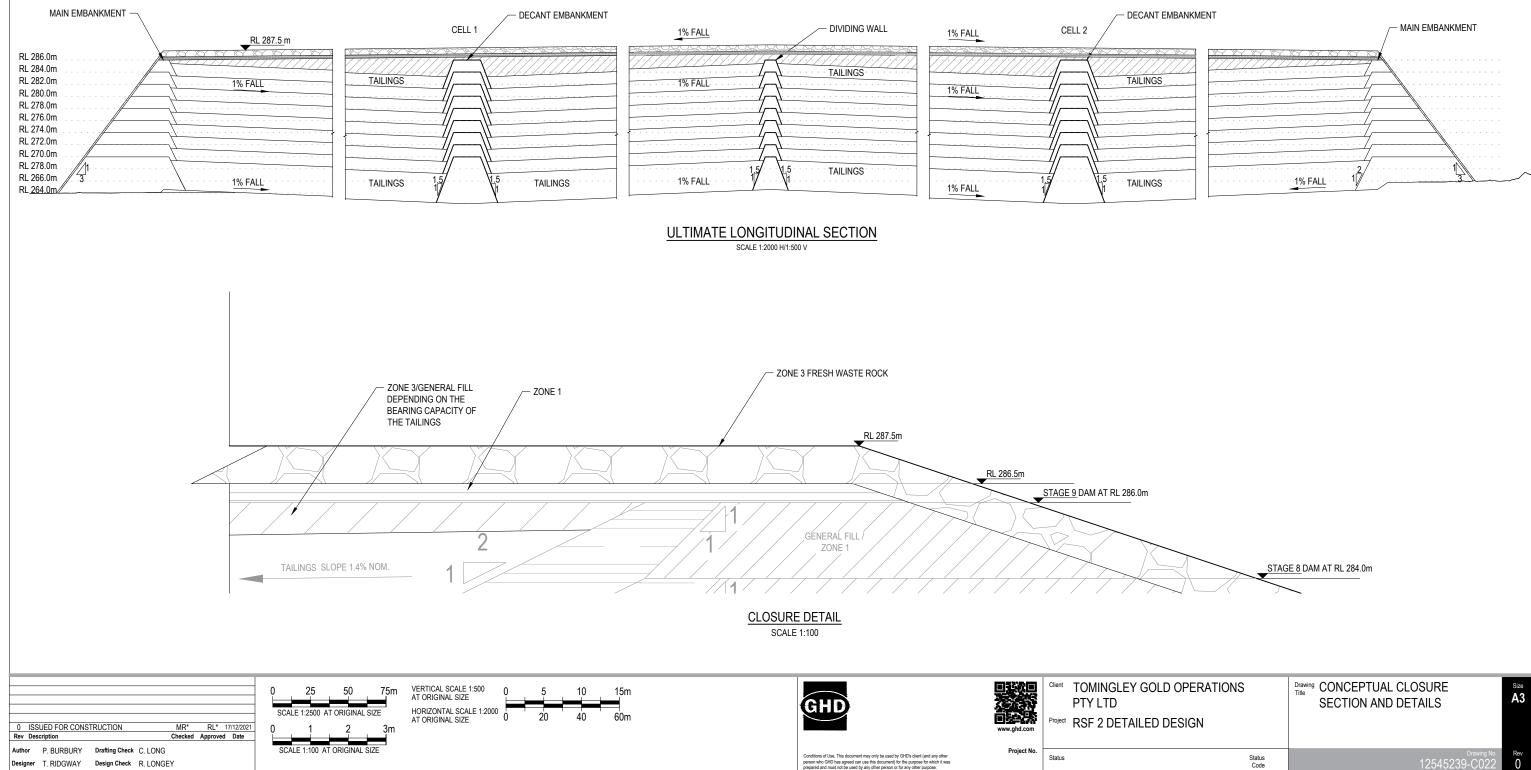


Plot Date: 17 December 2021 - 1:53 PM Plotted by: Melvin Agomuoh

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NOTES 1. DECANT WATER TO BE REMOVED FROM RSF USING EXISTING DECANT INFRASTRUCTURE

- PRIOR TO THE PLACEMENT OF THE CLOSURE MATERIAL, THE TAILINGS ARE TO BE 2. TRACK ROLLED FOR CONSOLIDATION
- WASTE ROCK AND GENERAL FILL IS TO BE USED TO INFILL THE DECANT POND AREA 3. PRIOR TO CAPPING THE RSF LANDFORM ALL CONTAMINATED TAILINGS INFRASTRUCTURE SUCH AS THE TAILINGS DEPOSITION
- 4. PIPELINE IS TO BE DISPOSED IN AN APPROVED LOCATION THE DECANT TOWER IS TO BE INFILLED WITH BULK CONCRETE FOLLOWING THE DECANT POND WATER REMOVAL AND PRIOR TO THE CAPPING OF THE FACILITY 5.
- ADDITIONAL WASTE ROCK IS T BE PLACED ON THE DOWNSTREAM FACE OF THE 6.
- PERIMETER EMBANKMENT FOR EROSION PROTECTION IF REQUIRED THE CAPPING SURFACE AND DOWNSTREAM FACE OF THE EMBANKMENT IS TO BE 7.
- REHABILITATED USING LOCALLY AVAILABLE NATIVE GRASSES



Plot Date: 17 December 2021 - 1:55 PM Plotted by: Melvin Agomuch

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LEGEND:



ZONE 1 - LOW PERMEABILITY RESIDUAL SOIL TO EXTREMELY WEATHERED ROCK

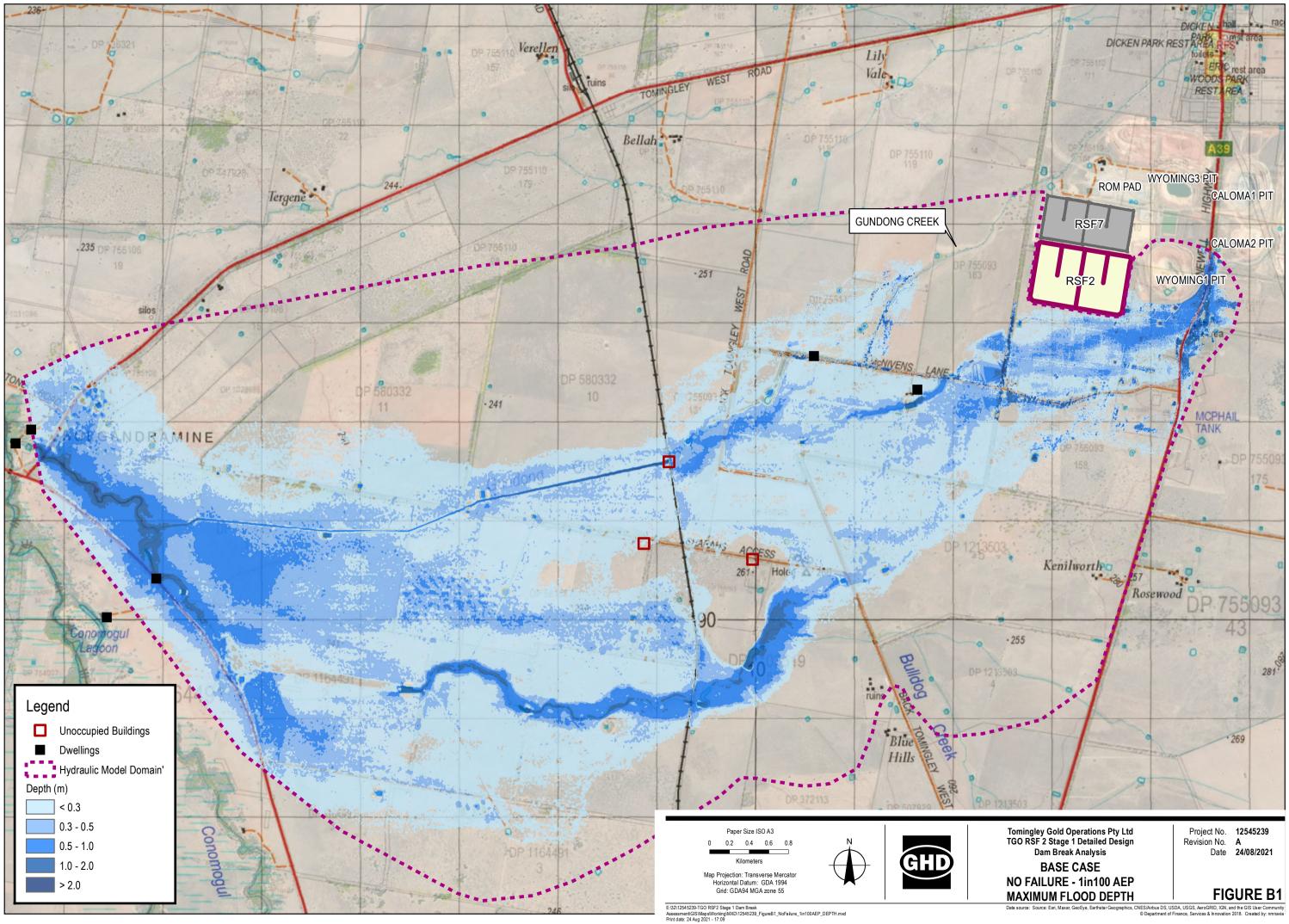


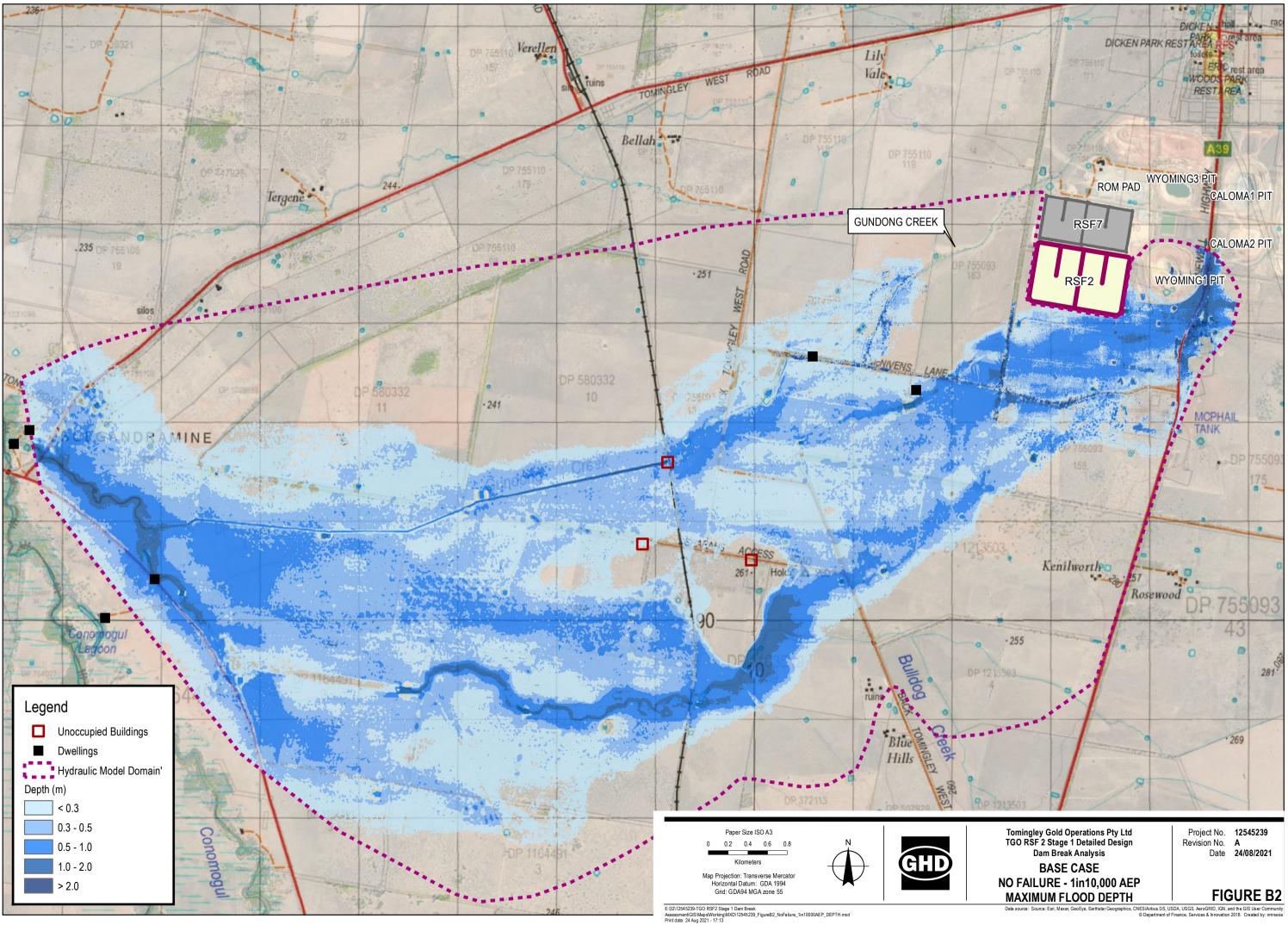
GENERAL FILL / ZONE 1

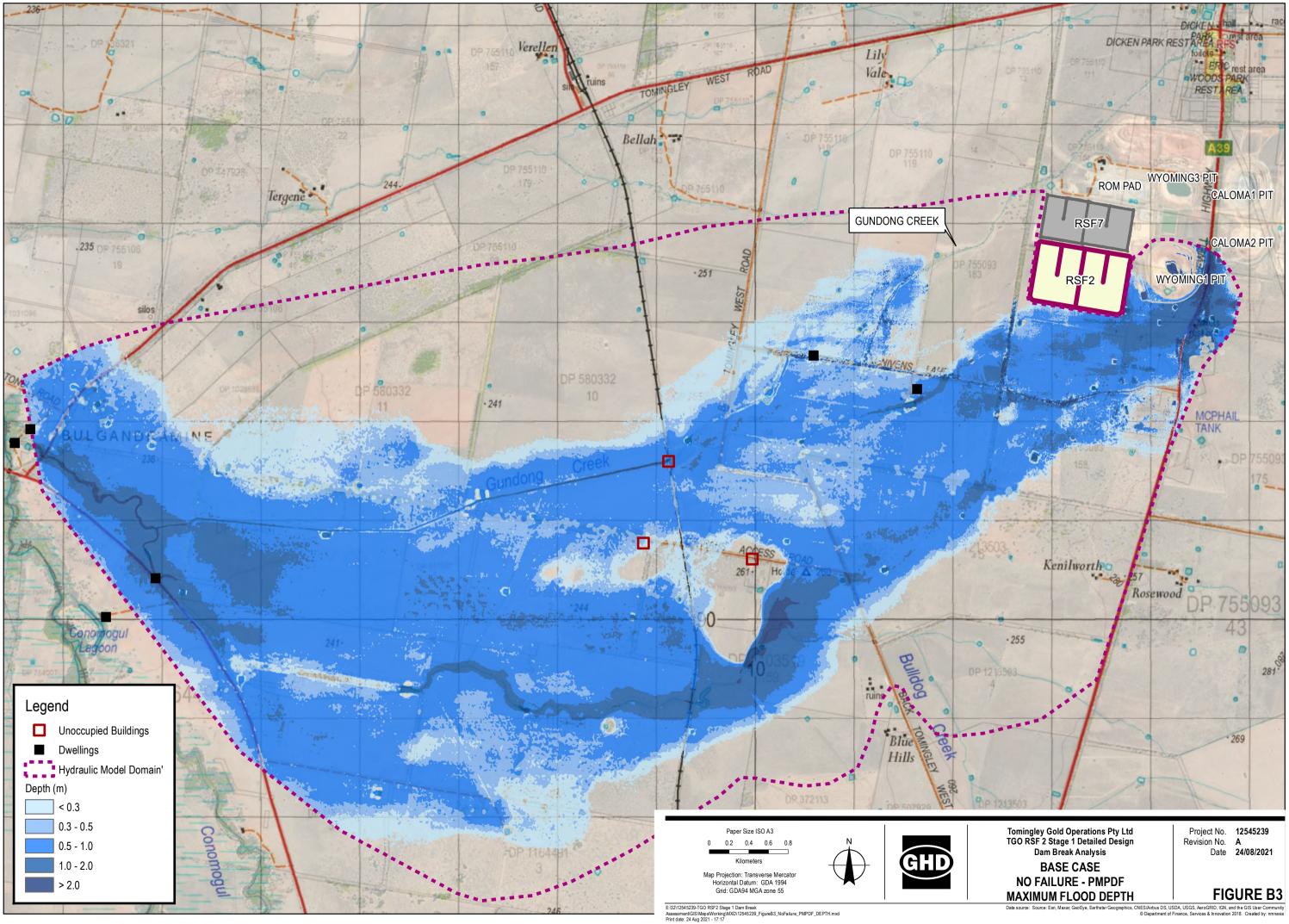


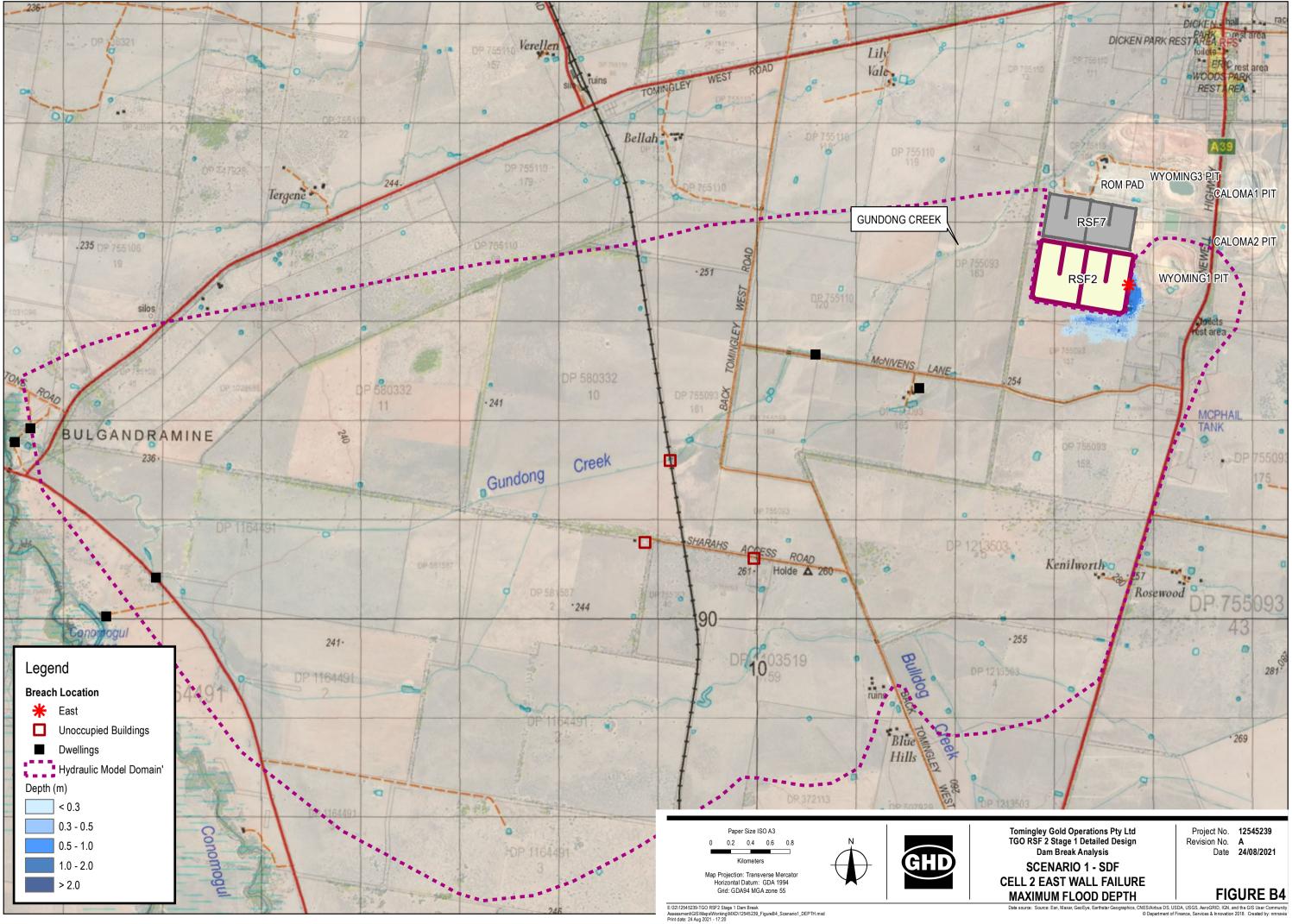
ZONE 3 - WASTE ROCK

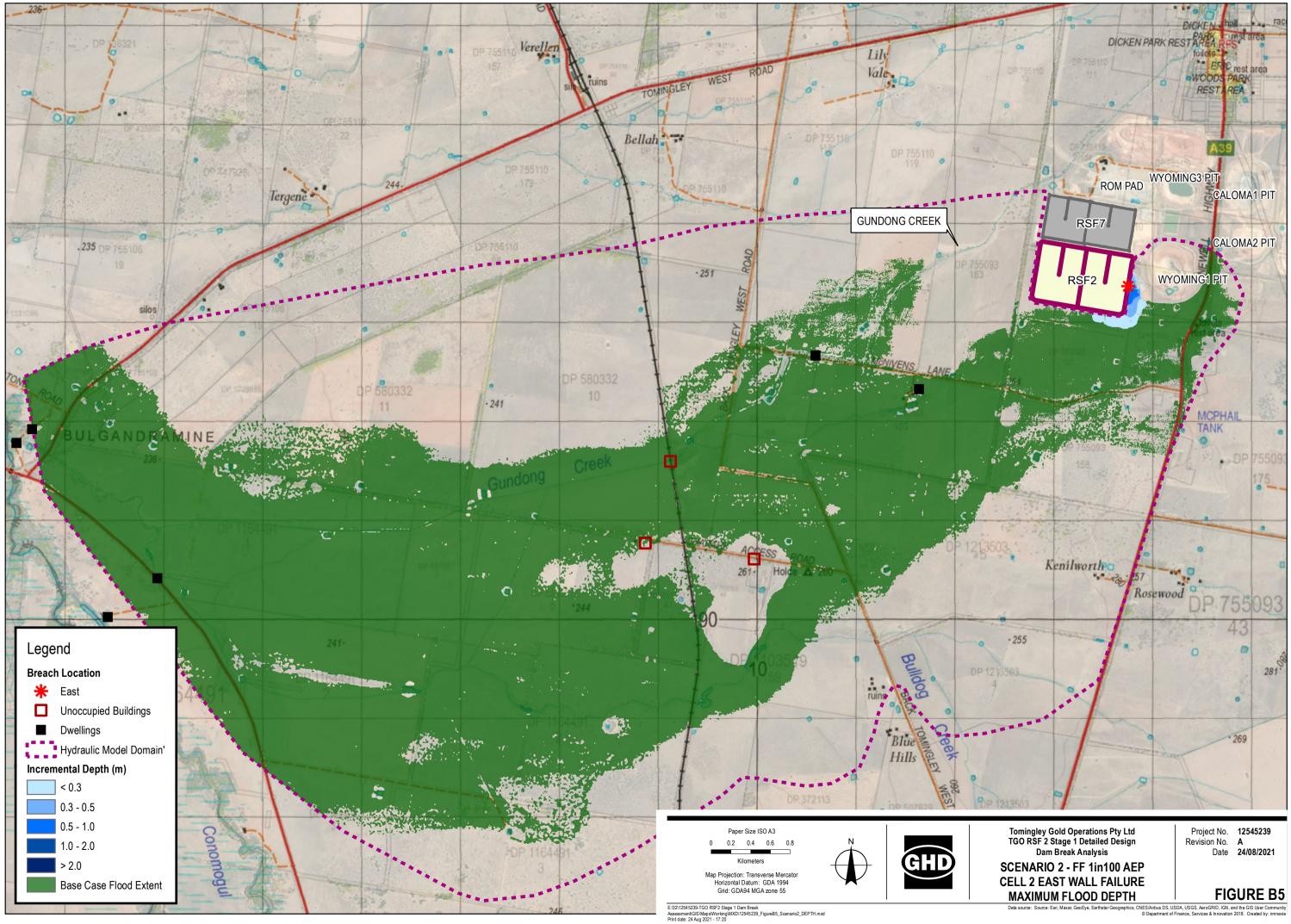
Appendix B RSF2 Dam Break Modelling Results

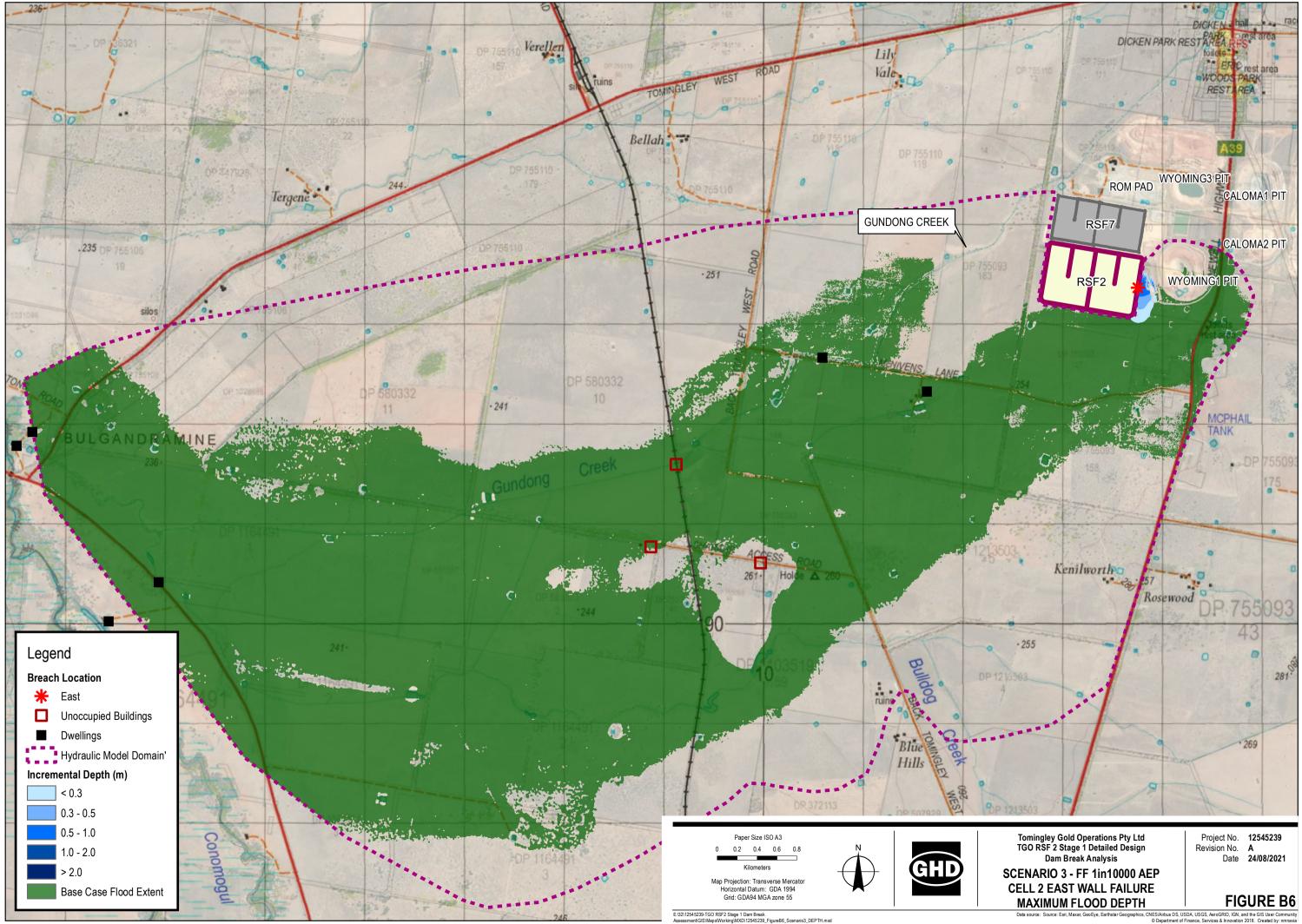




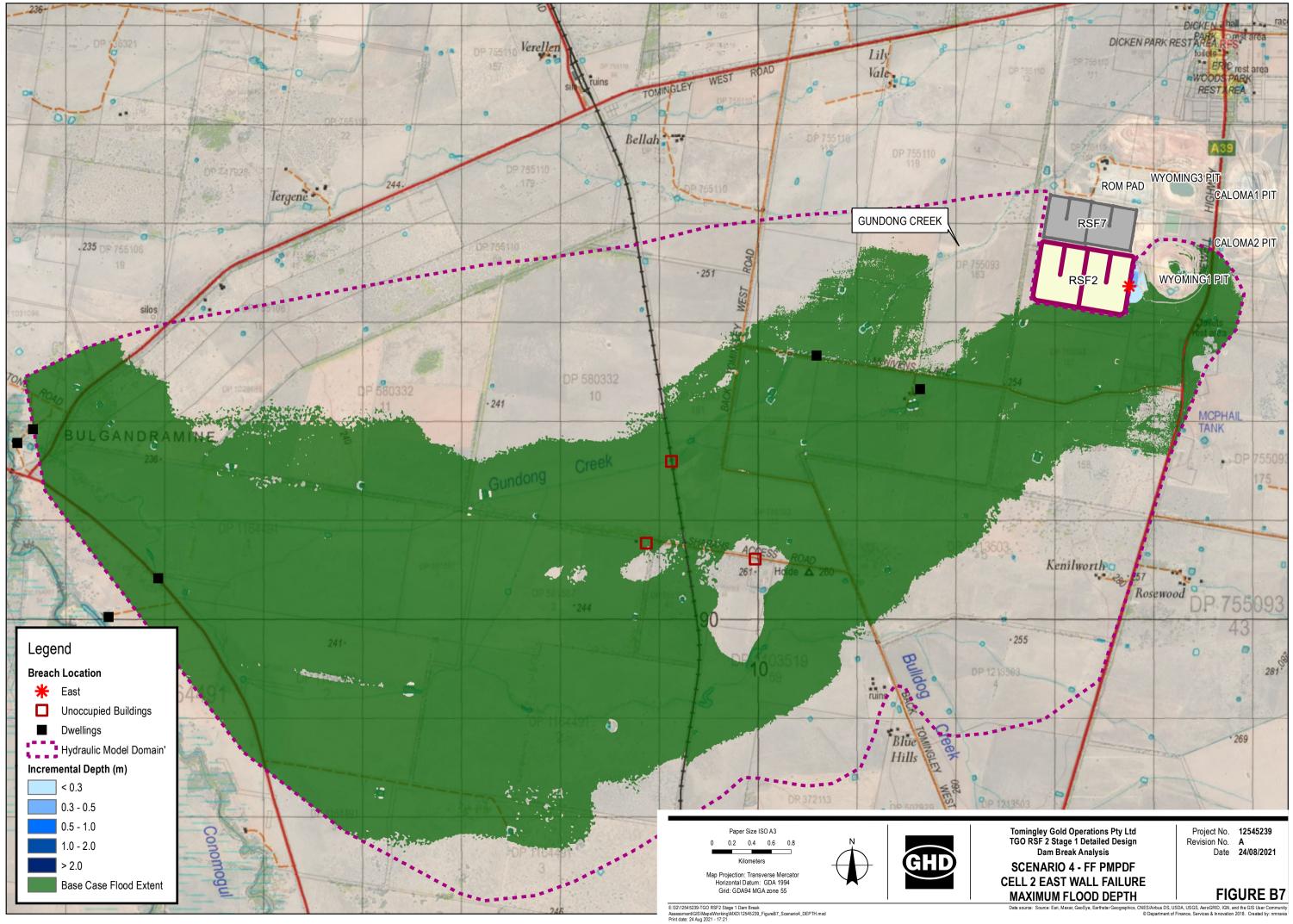


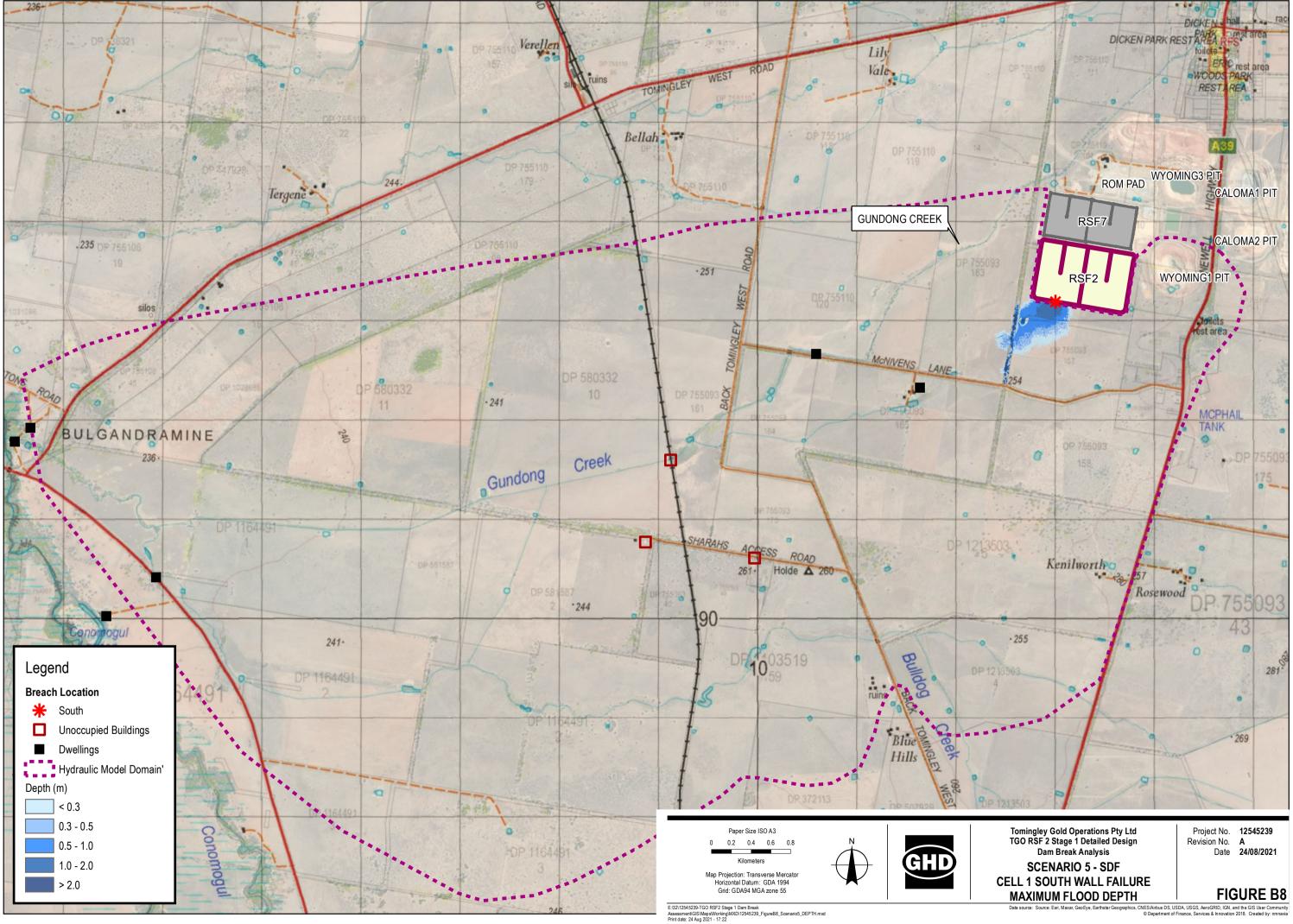


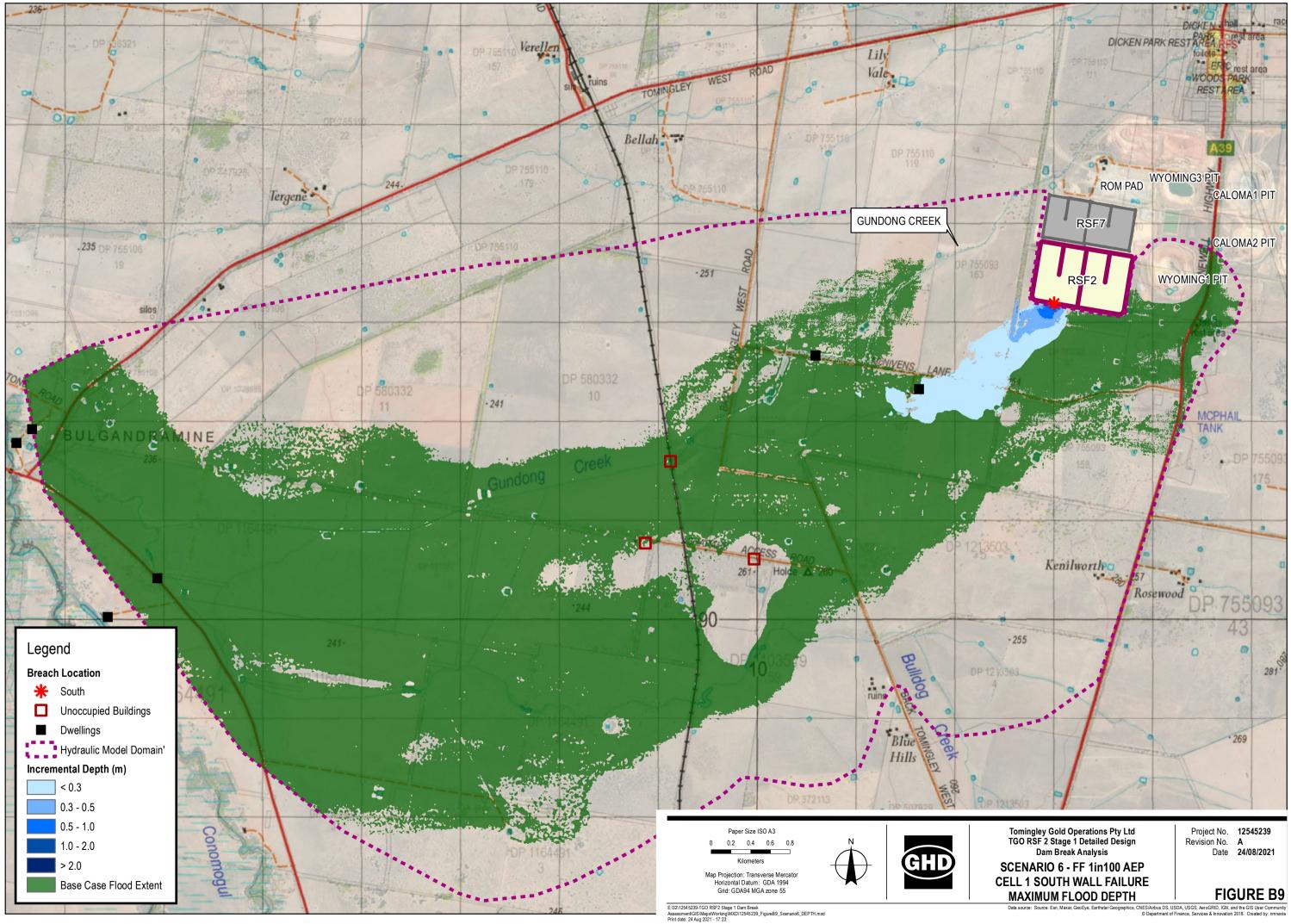


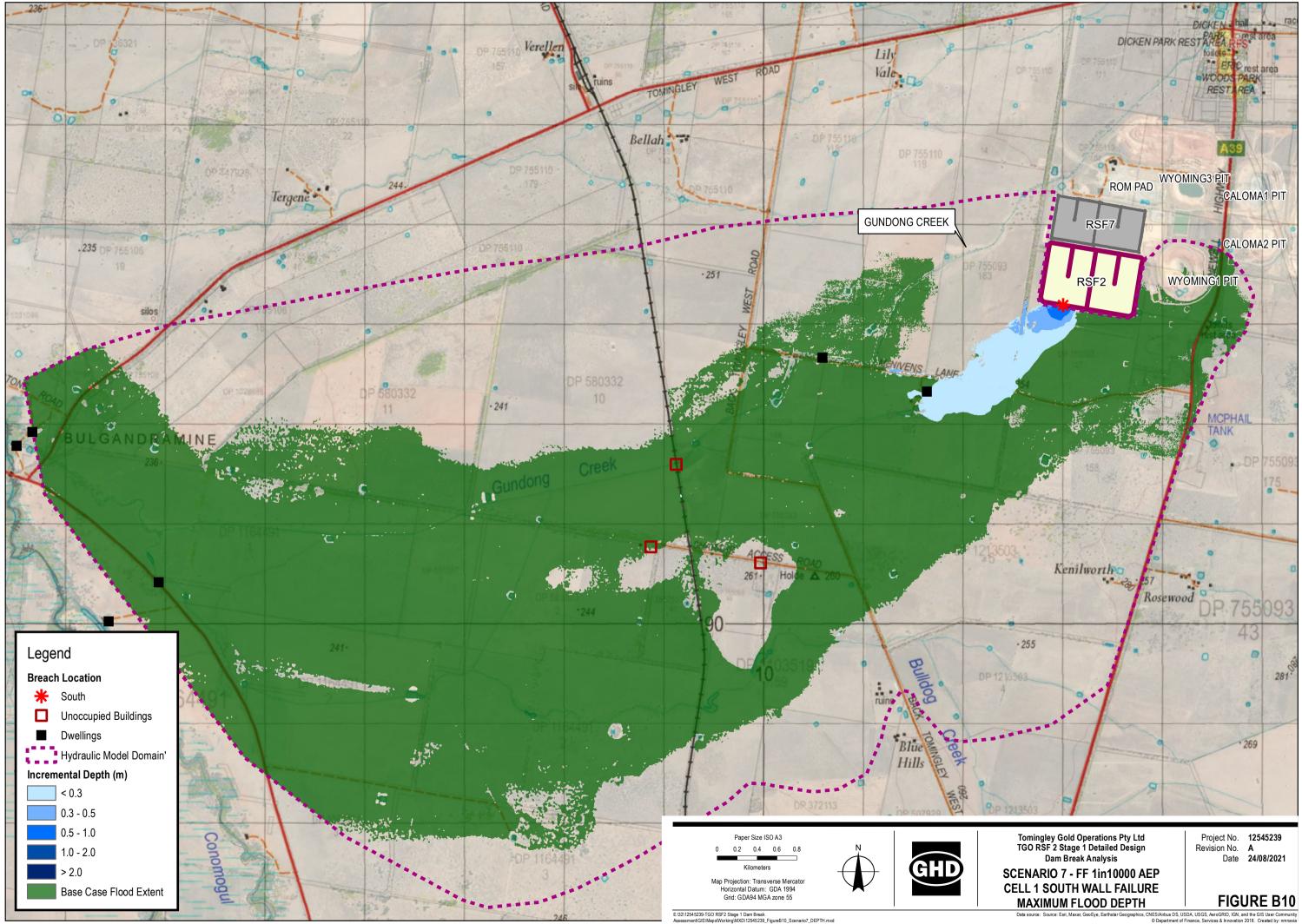


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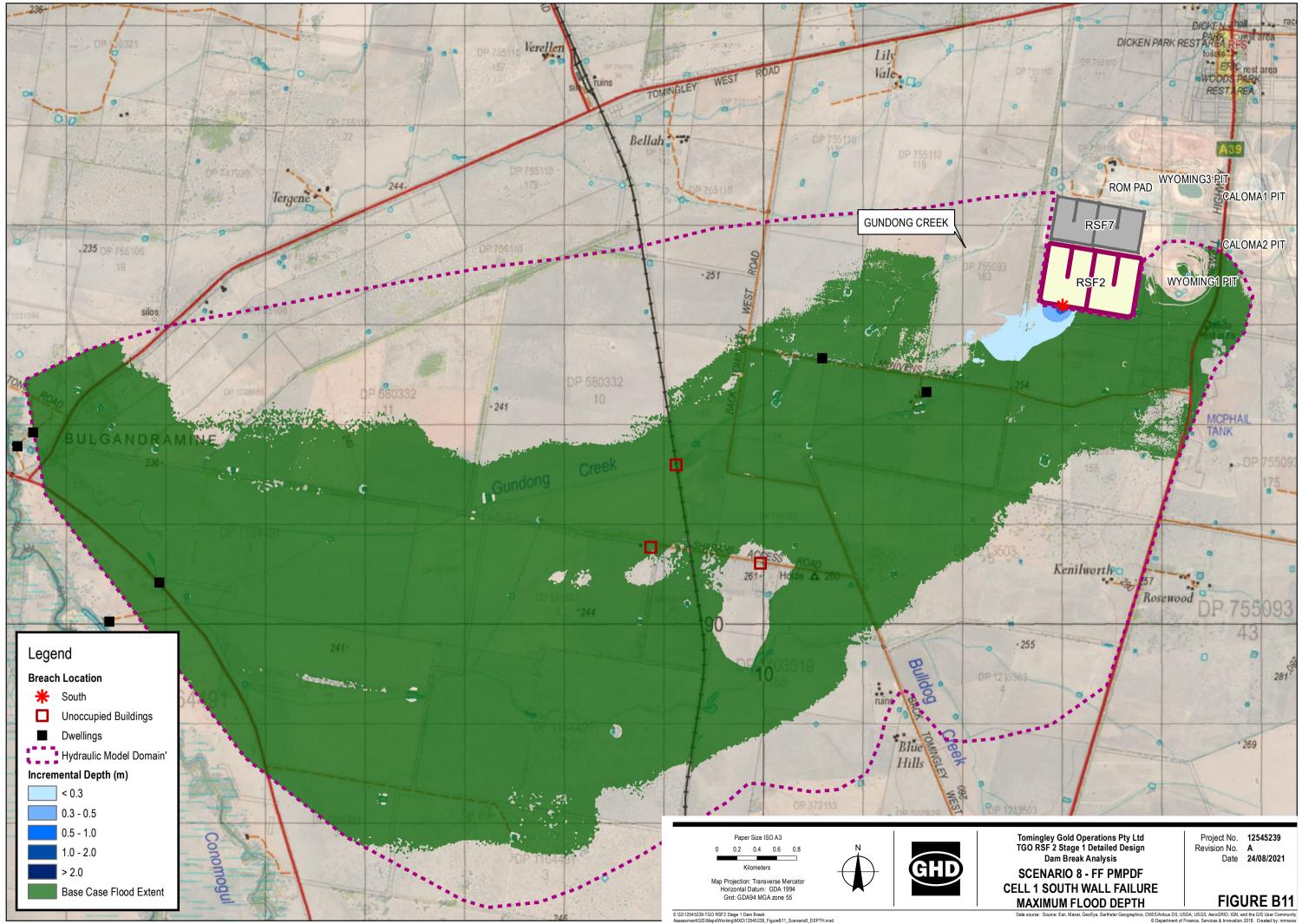




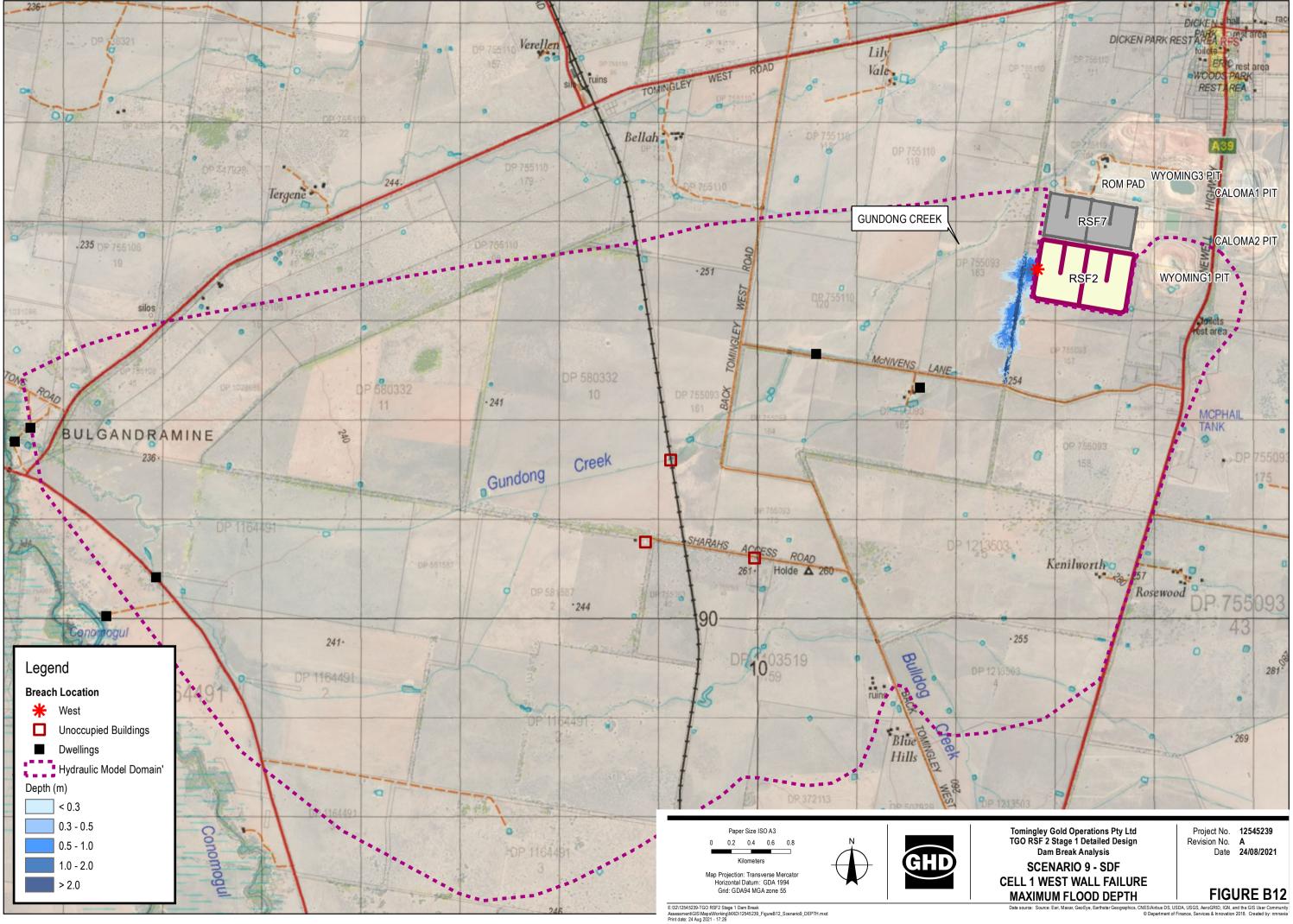


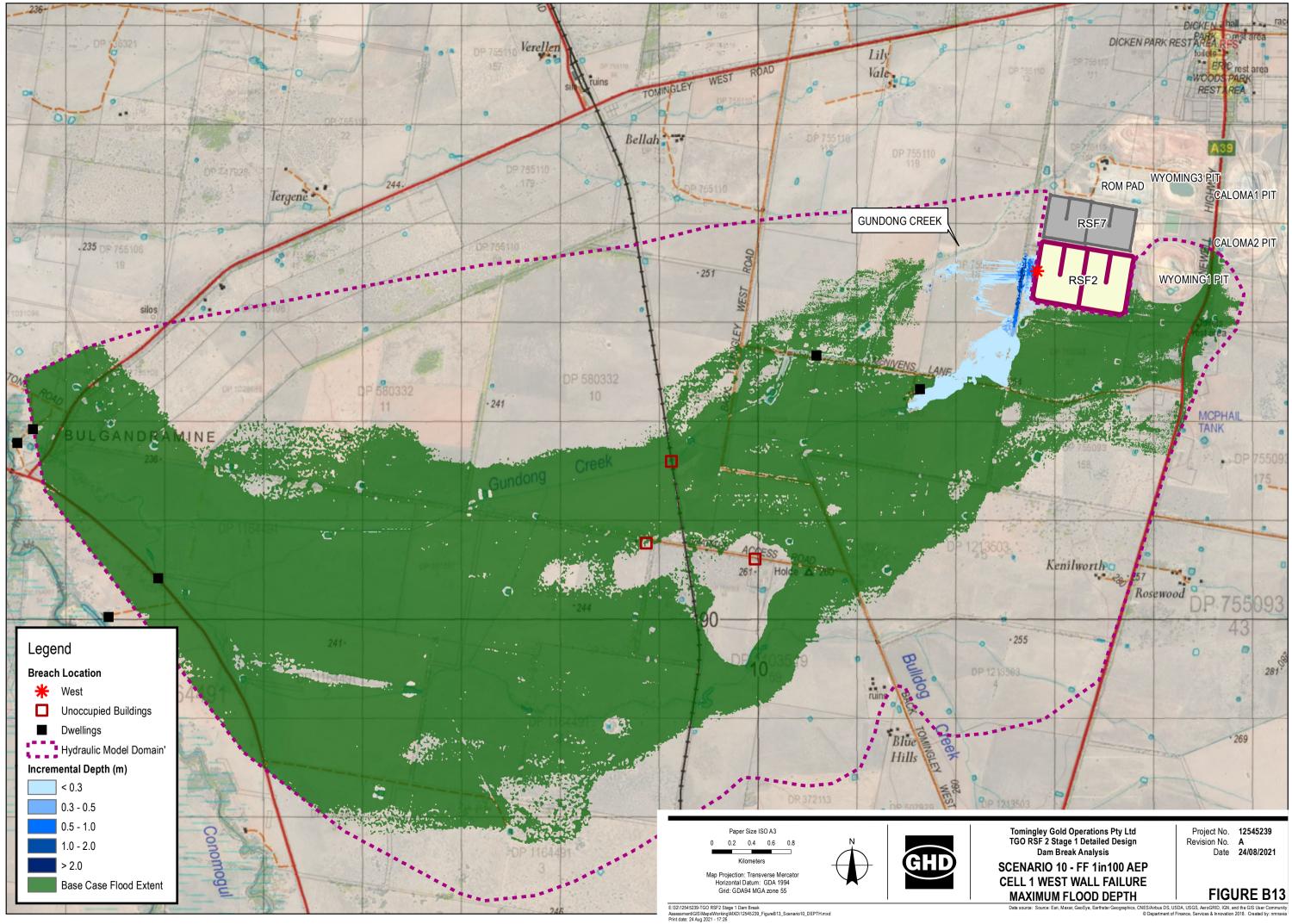


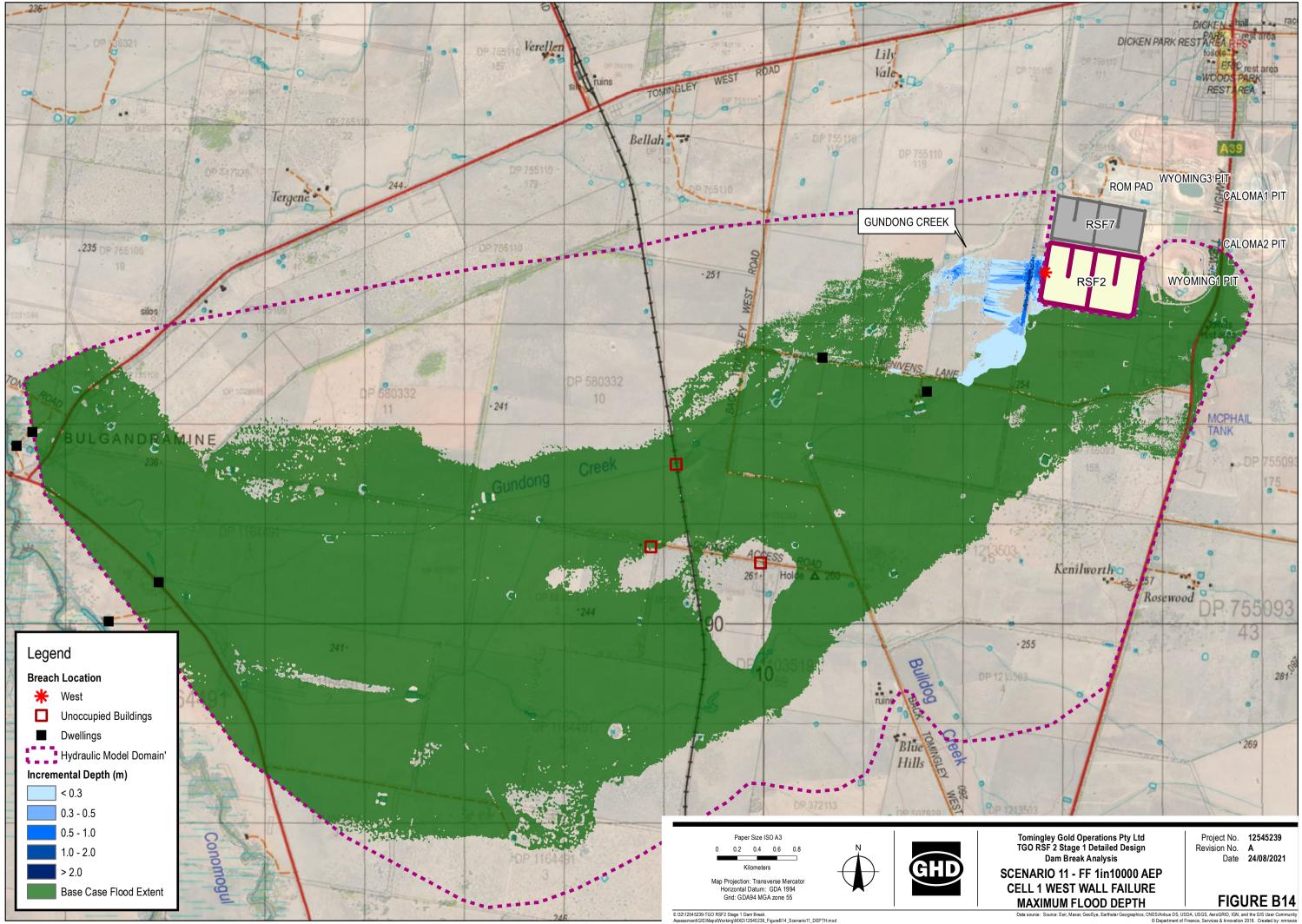
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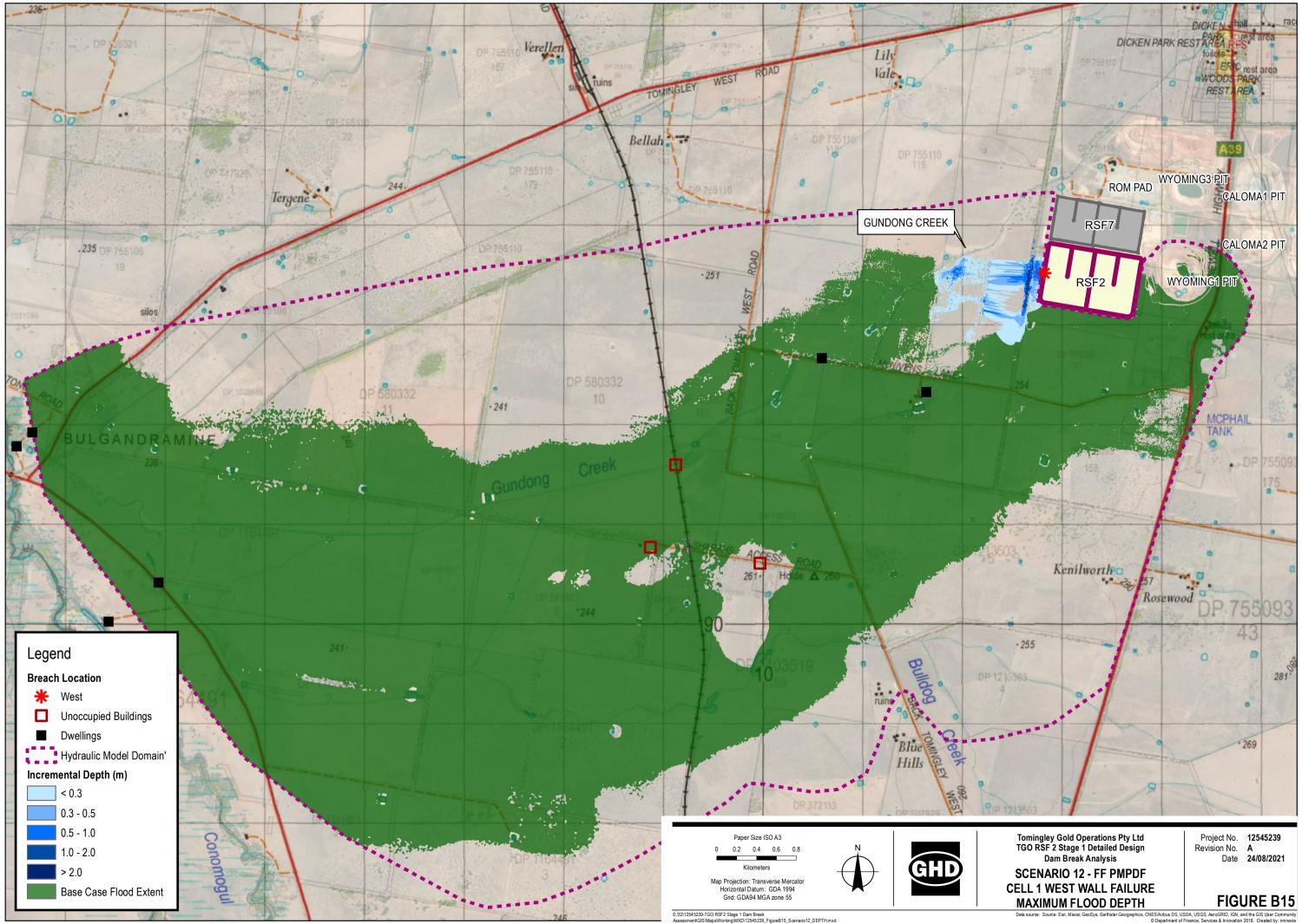
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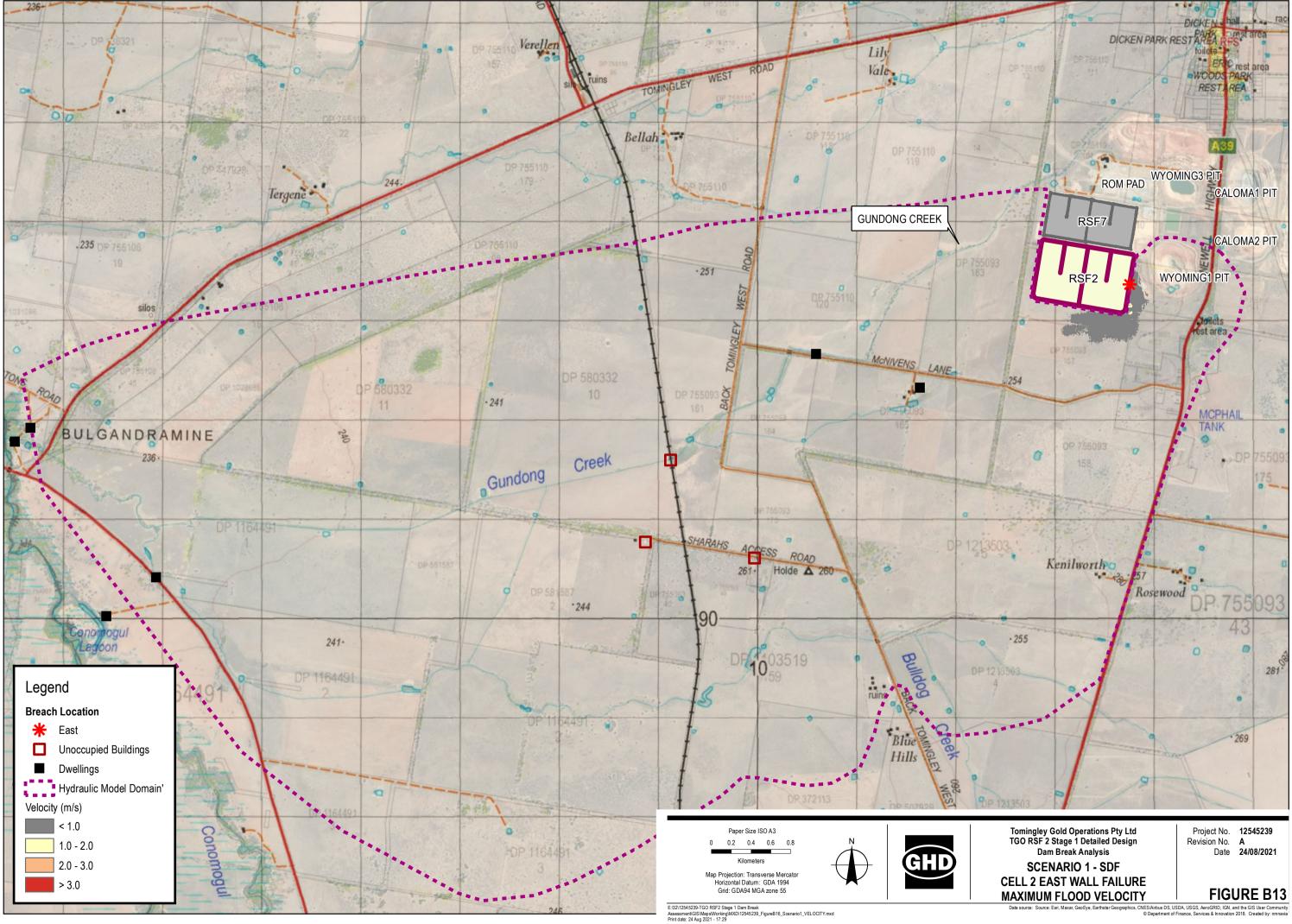


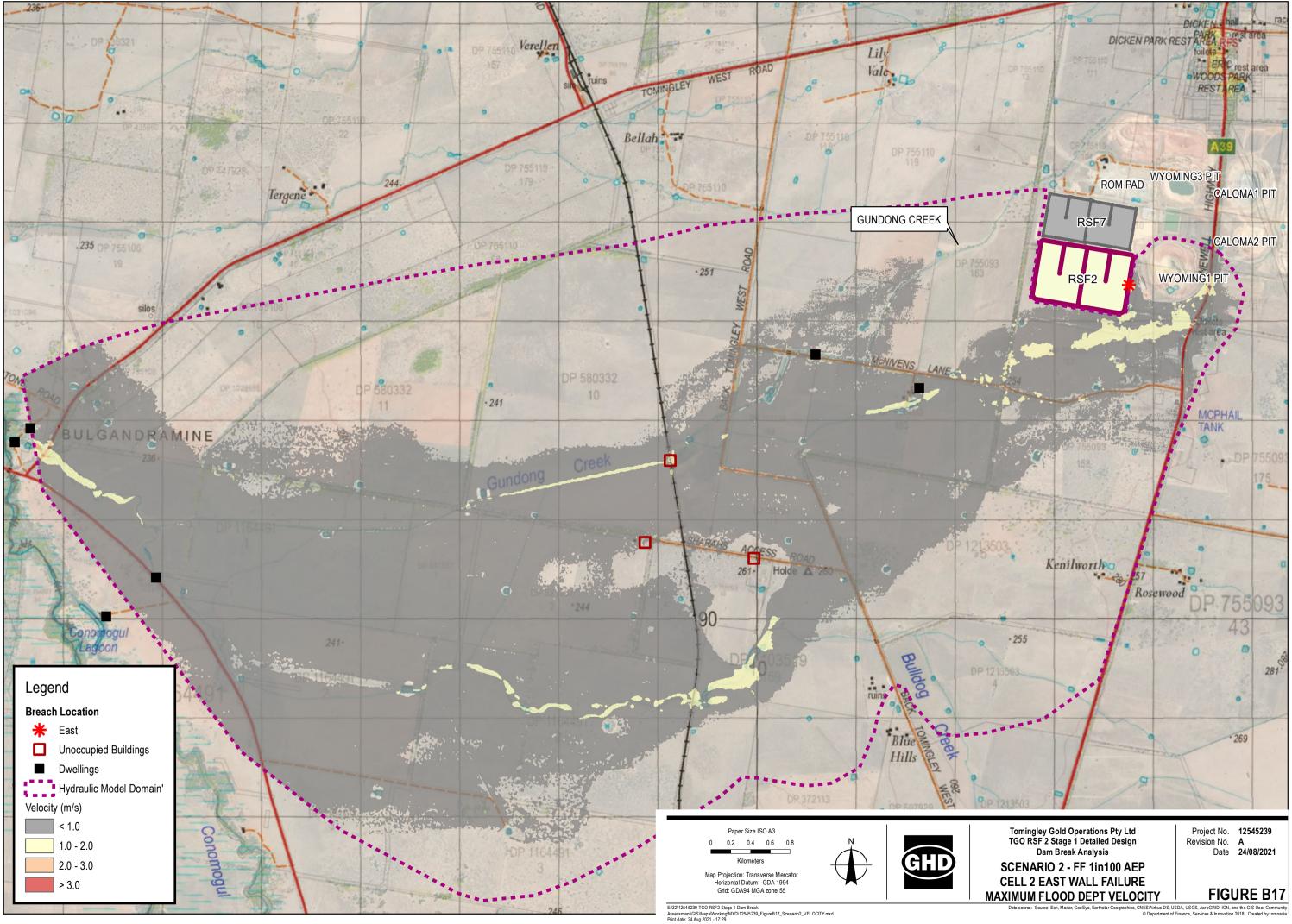


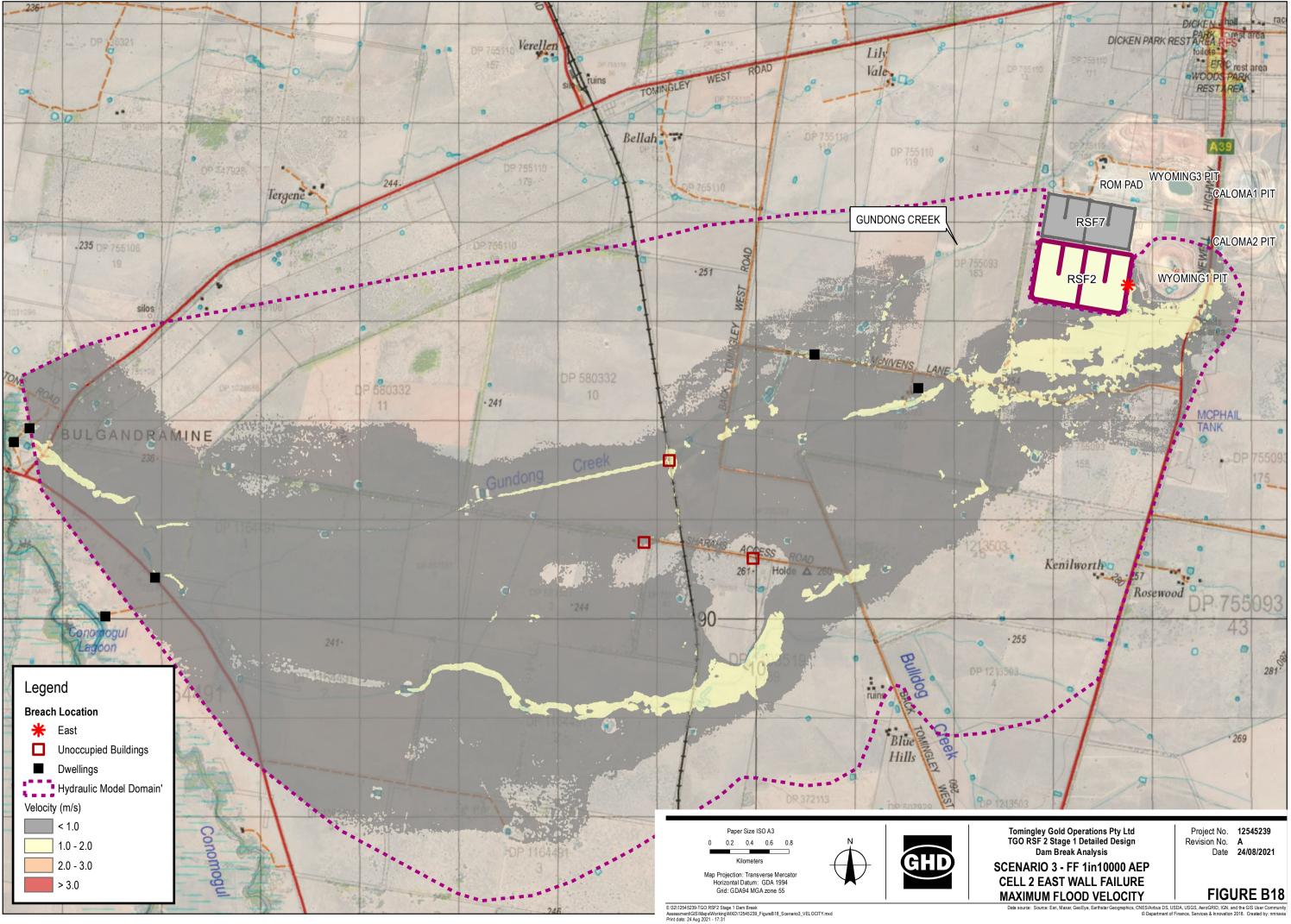
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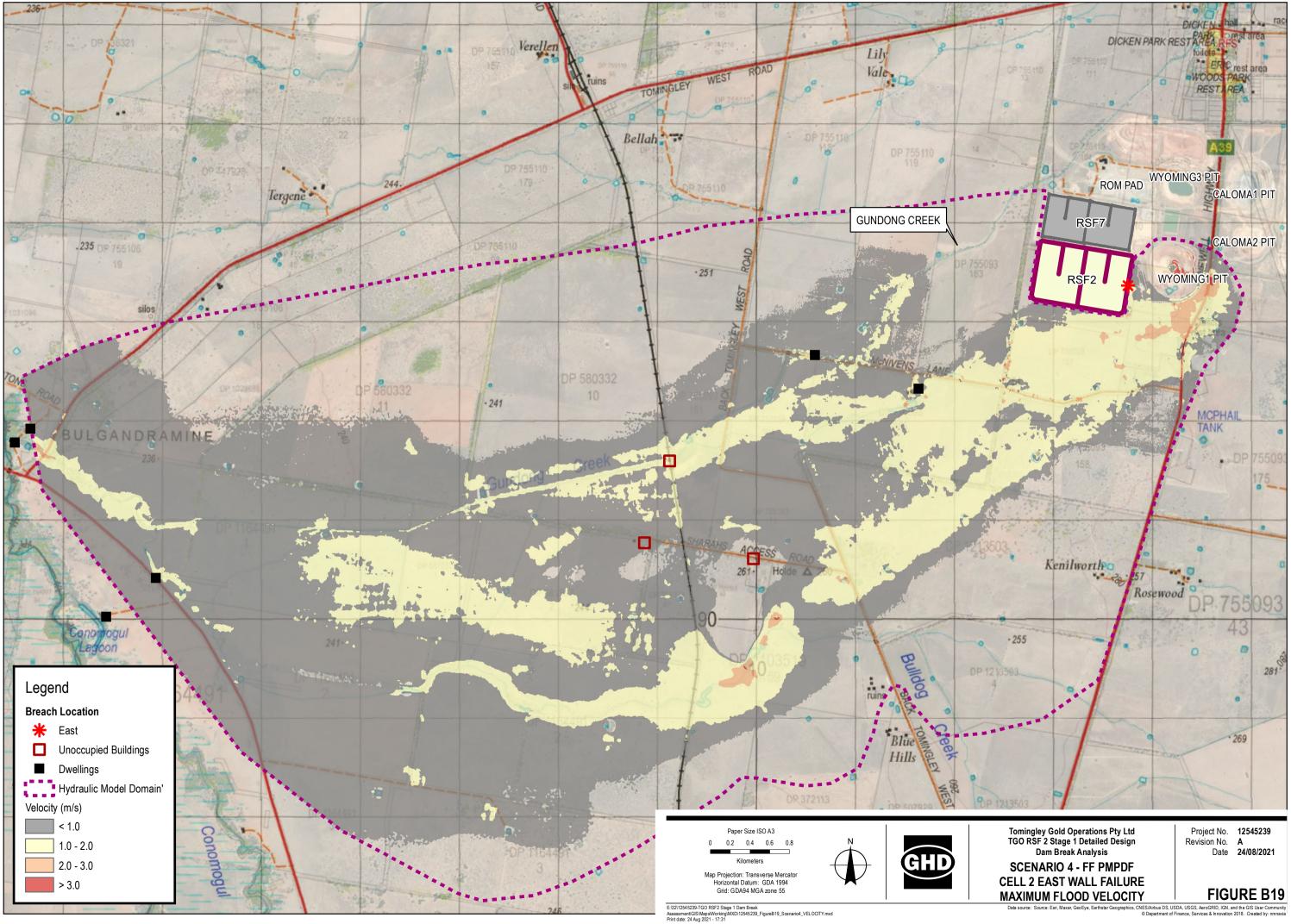


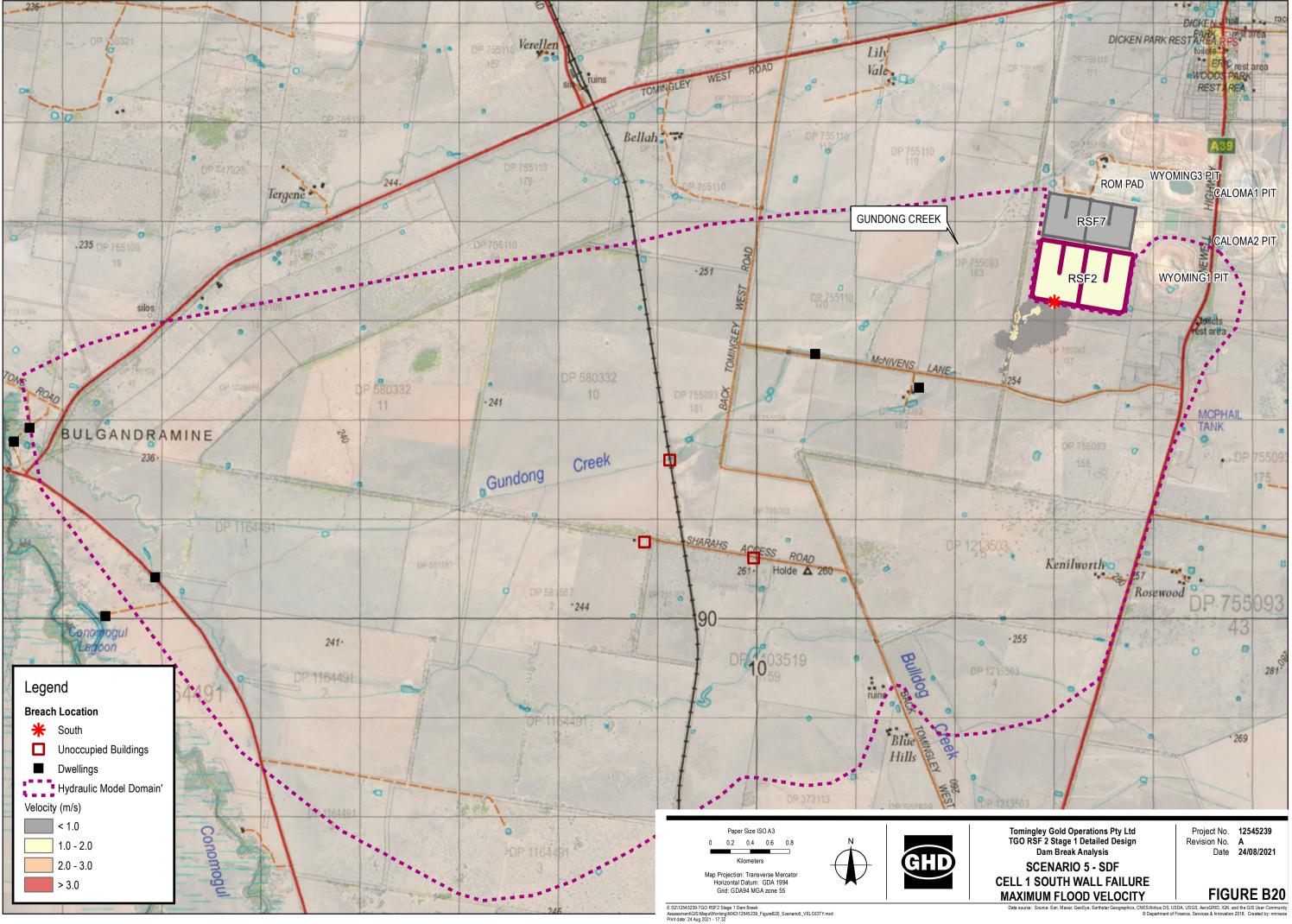
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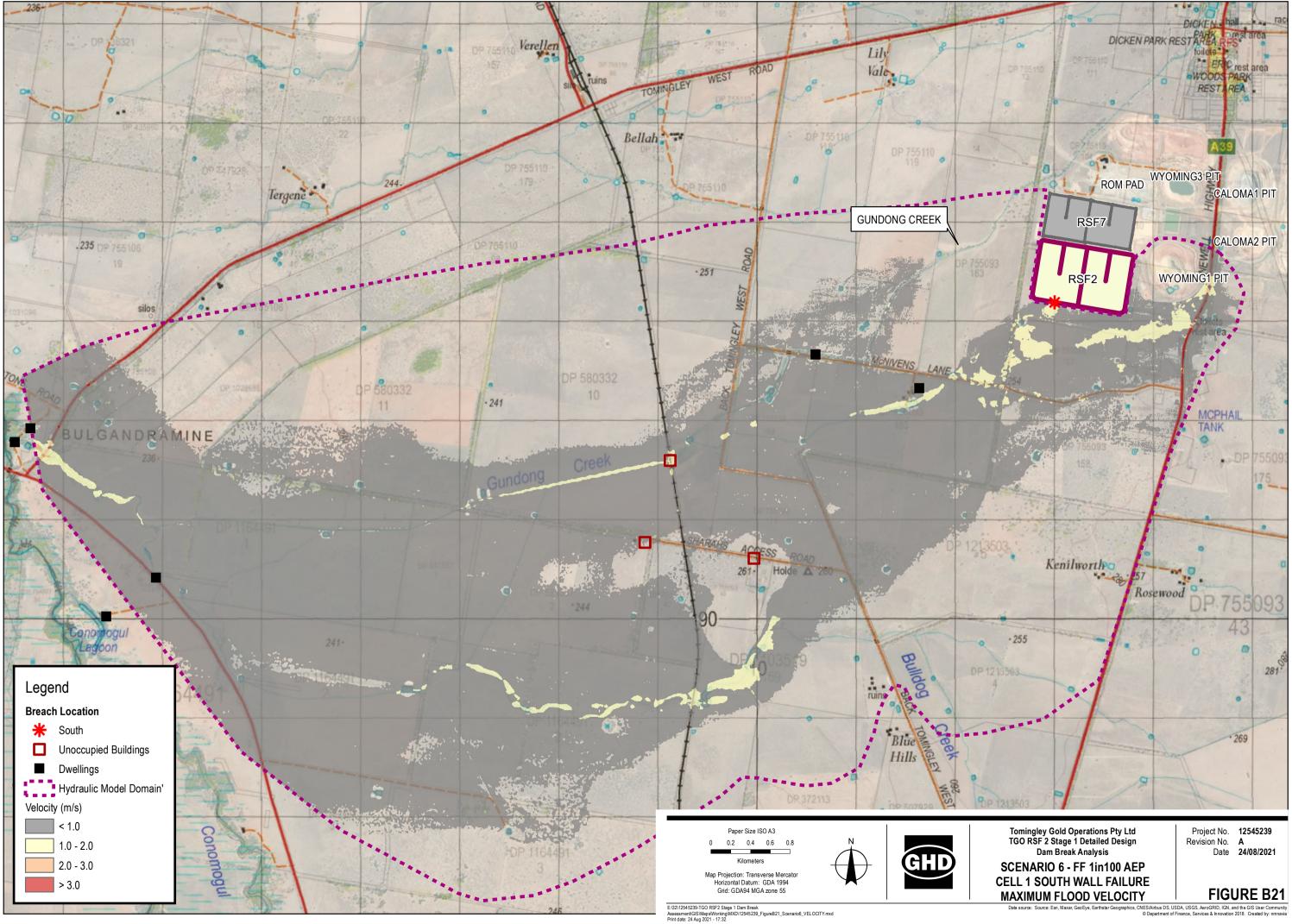


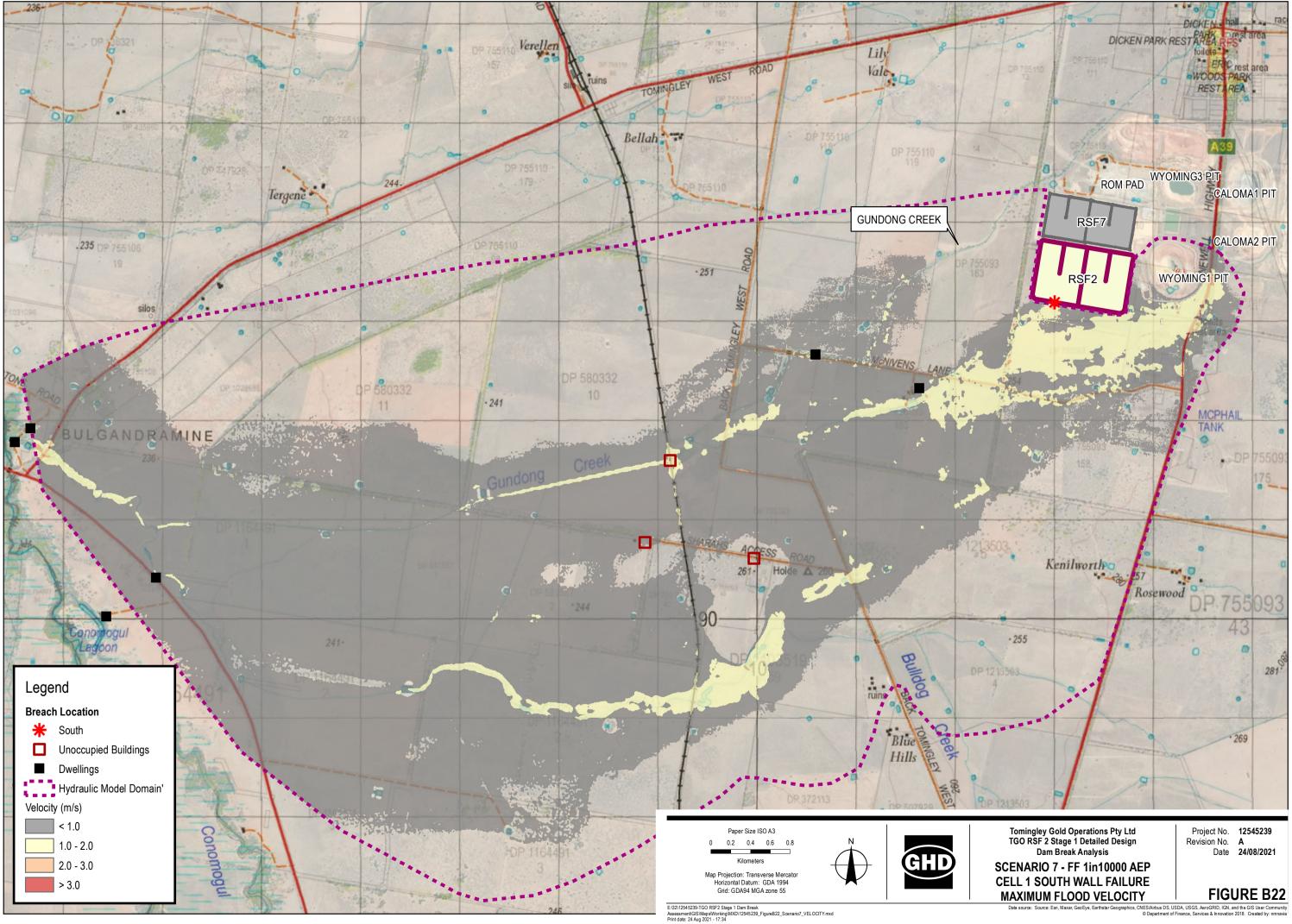


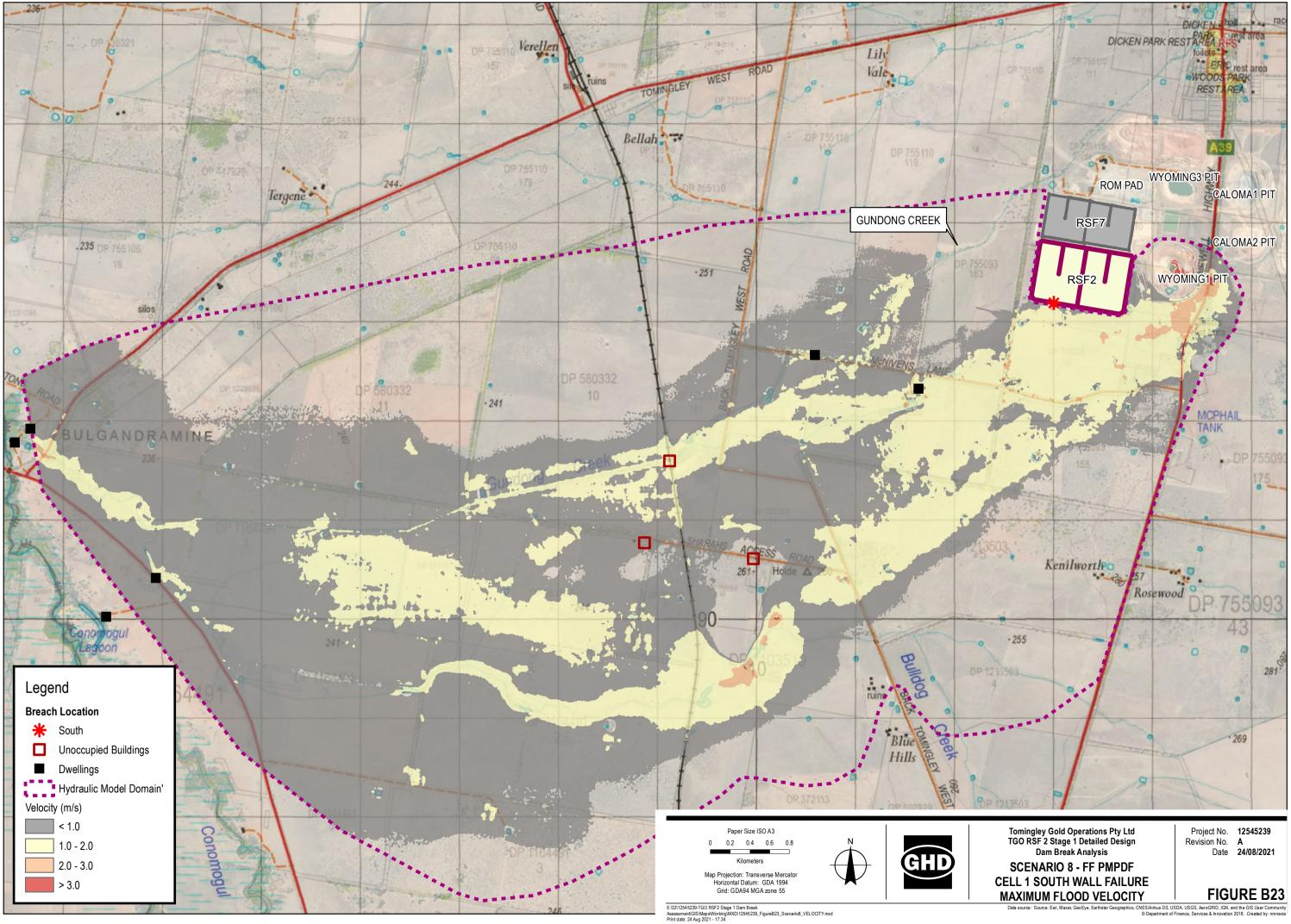


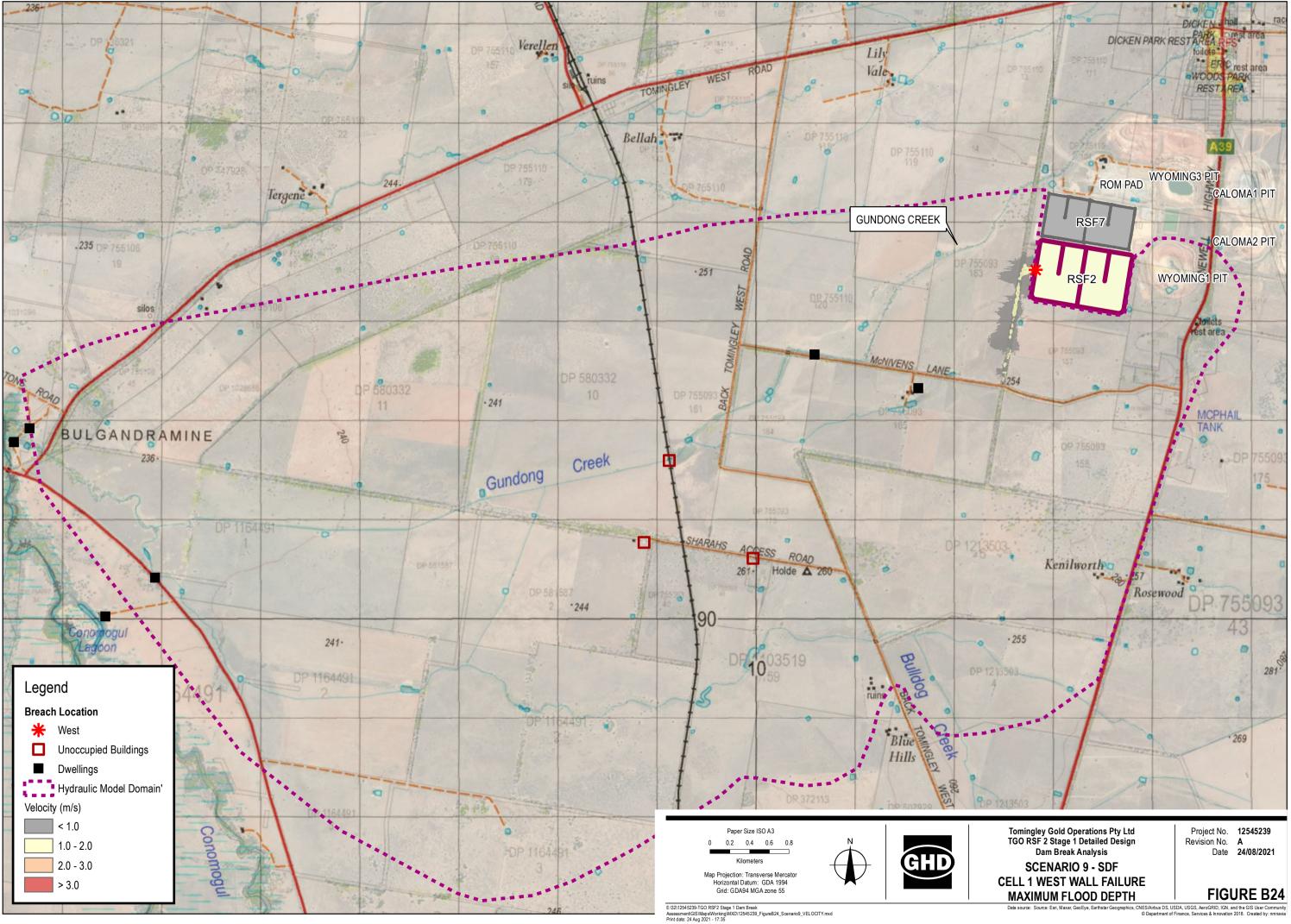


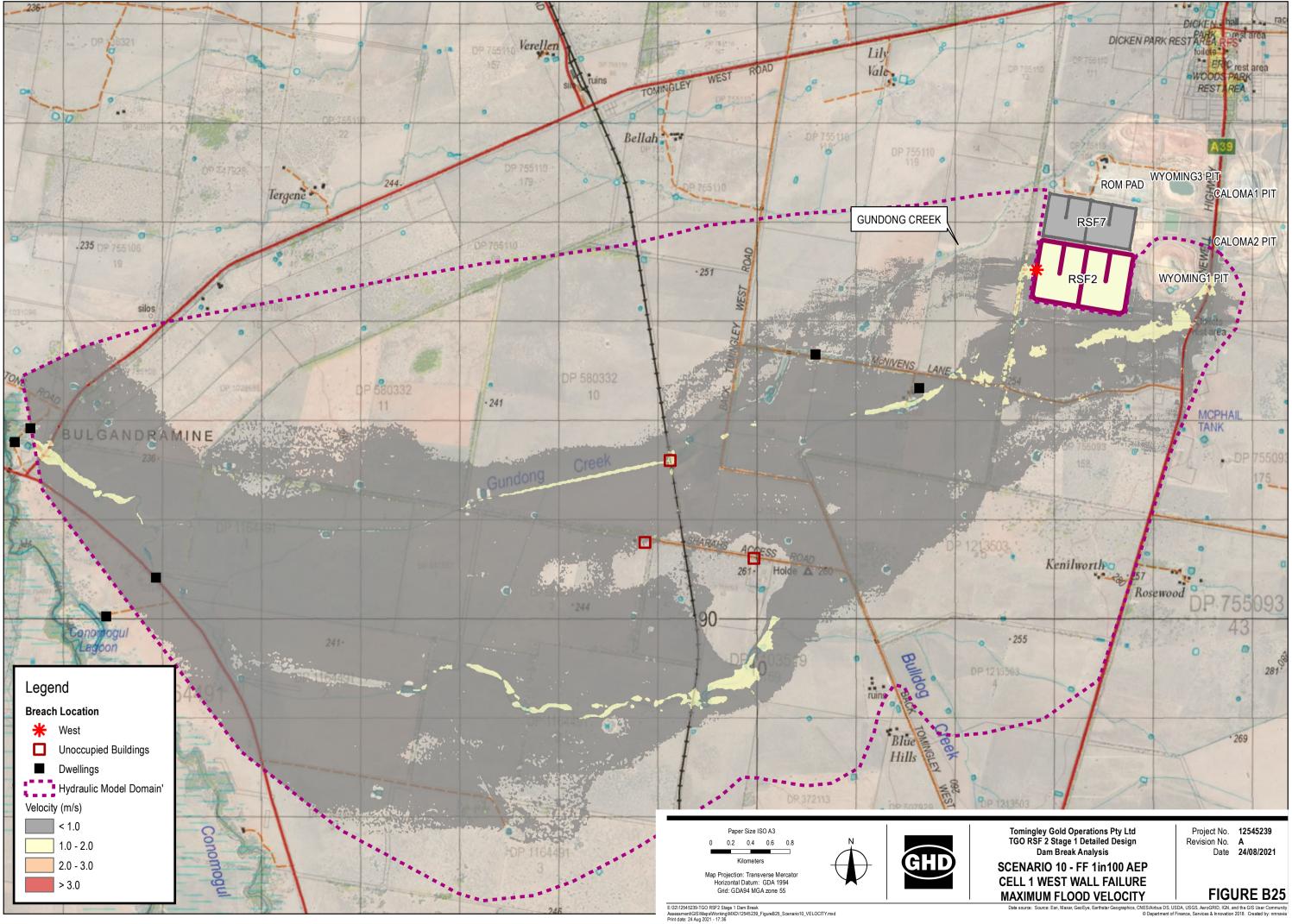


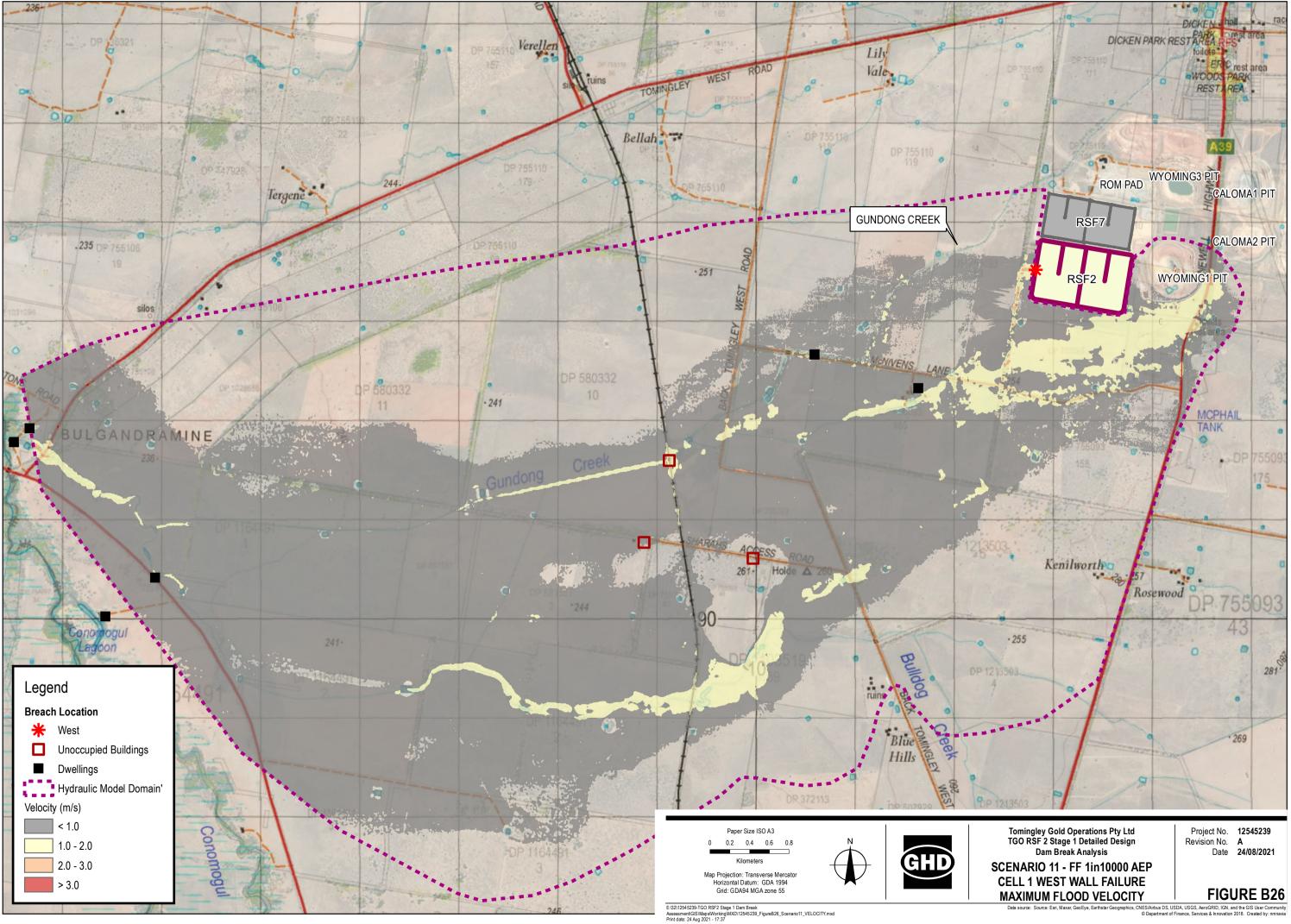


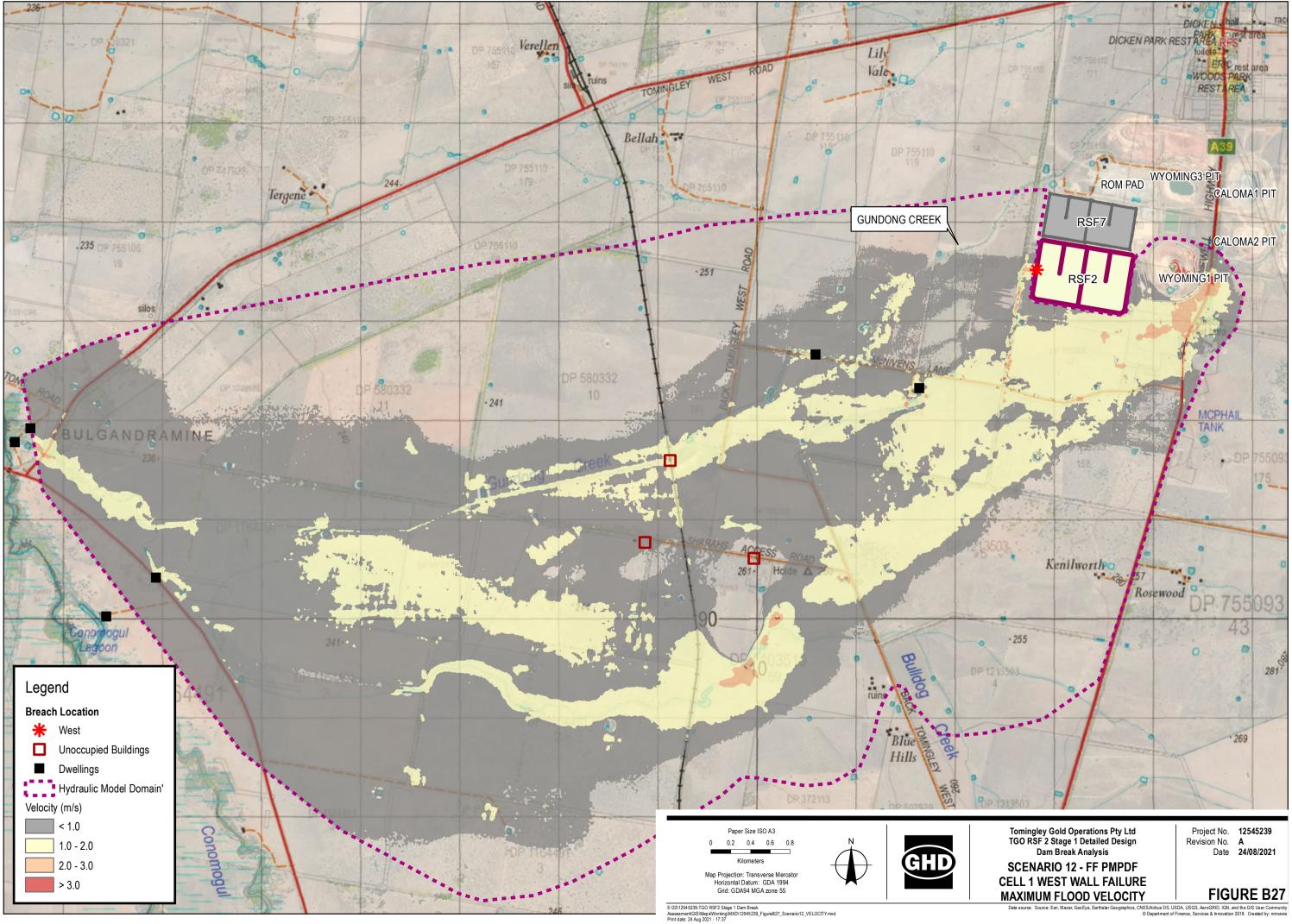












CONSEQUENCE CATEGORY ASSESSMENT



Sunny Day Failure Scenario

Sunny Day Failure Scenario					
Client Name	TGO				
Dam Name	RSF2				
Dam ID. No. (If existing dam)					
Stream Name		-			
Dam Height (Metres)	6	Crest	RL	270.	.0 m
Estimated Capacity at FSL (Megalitres)					
Location	Tomingley				
			•.		
		S	everit	y Lev	
					Catastrophic
Damage and Loss	Estimate		ε		trop
		or	diu	Major	ast
		Minor	Medium	Ma	Cat
TOTAL INFRASTRUCTURE COSTS (costs are indicative of		-			
Residential	<10M	YES	•	•	
Commercial	\$10M - \$100M		YES		•
Community Infrastructure	<10M	YES		•	•
Dam replacement or repair cost	\$10M - \$100M	•	YES MEC		
TOTAL INFRASTRUCTURE COSTS severity level IMPACT ON DAM OWNER'S BUSINESS			WIEL		
Importance to the business	Restrictions needed during peak days and peak hour		YES		
Effect on services provided by the owner			-	•	
	Reduced services are possible with reasonable restrictions	•	YES	·	•
Effect on continuing credibility	Extreme discontent			YES	
Community reaction and political implications	Extreme discontent			YES	
Impact on financial viability	Able to absorb in 1 financial year	YES			
Value of water in storage (assessed by the owner in relation	Can be absored in one financial year	YES			
IMPACT ON DAM OWNER'S BUSINESS damage and loss HEALTH and SOCIAL IMPACTS	severity level		MA	IUR	
Public health	<100 people affected	YES			
Loss of service to the community	<100 people affected	YES			
Cost of emergency management	<1,000 person days	YES			
Dislocation of people	<100 person months	YES			-
Dislocation of businesses	<20 business months	YES			
Employment affected	<100 jobs lost	YES			
Loss of heritage	Local facility	YES	•	·	
Loss of recreational facility	Local facility	YES			
HEALTH and SOCIAL IMPACTS damage and loss severit	y level		MIN	OR	
NATURAL ENVIRONMENT Area of Impact	<1km2	YES			
Duration of Impact	<1 (wet) year	YES	-	-	·
Stock and Fauna		120	•	•	•
	Discharge from dambreak would contaminate water supplies		YES		
	used by stock and fauna. Health impacts not expected.				
Ecosystems	Discharge from dambreak would have short term impacts on				
	ecosystems with natural recovery expected after 1 wet season.	· ·	YES		•
	Remediation possible.				
Rare and endangered fauna and flora	Species exist but minimal damage expected. Recovery within	YES			
NATURAL ENVIRONMENT damage and loss severity leve	one year.		MED		
HIGHEST DAMAGE AND LOSS SEVERITY LEV	EL		MA	IOR	
Population at Risk (PAR)	<1				
PAR includes all those persons who would be directly exposed to flood waters within the dam break affected zone if		S	IGNIF	ICAN	Т

Completed and Reviewed By	Tom Ridgway
completed and Keviewed by	Tom Ragway
Date	14/09/2021
Dale	1-1/03/2021

CONSEQUENCE CATEGORY ASSESSMENT



Environmental Release Scenario

Environmental Release Scenario					
Client Name	TGO				
Dam Name	RSF2				
Dam ID. No. (If existing dam)					
Stream Name					
Dam Height (Metres)	6	Crest	RL	270	.0 m
Estimated Capacity at FSL (Megalitres)					
Location	Tomingley				
	5,		-	_	
		Se	everit	y Lev	
					Catastrophic
Damage and Loss	Estimate		٦		rop
		ē	ni	o	ast
		Minor	Medium	Major	Cat
TOTAL INFRASTRUCTURE COSTS (costs are indicative	only)			_	
Residential	<10M	YES			
Commercial	\$10M - \$100M		YES		
Community Infrastructure	<10M	YES	•		
Dam replacement or repair cost	\$10M - \$100M	<u> </u>	YES	•	
TOTAL INFRASTRUCTURE COSTS severity level			MED	NUM	
IMPACT ON DAM OWNER'S BUSINESS			VEO		
Importance to the business	Restrictions needed during peak days and peak hour	•	YES	•	•
Effect on services provided by the owner	Reduced services are possible with reasonable restrictions		YES		
Effect on continuing credibility	Severe widespread reaction		YES		
Community reaction and political implications	Severe widespread reaction	•	YES	•	•
Impact on financial viability	Able to absorb in 1 financial year	YES	120	•	•
Value of water in storage (assessed by the owner in relation			•	•	
to the business)	Can be absored in one financial year	YES	•	•	•
IMPACT ON DAM OWNER'S BUSINESS damage and los	s severity level		MED	NUI	
HEALTH and SOCIAL IMPACTS					
Public health	<100 people affected	YES	•		
Loss of service to the community	<100 people affected	YES			
Cost of emergency management	<1,000 person days	YES			
Dislocation of people	<100 person months	YES			
Dislocation of businesses	<20 business months	YES	•	•	•
Employment affected	<100 jobs lost	YES	·	•	•
Loss of heritage	Local facility	YES	•	•	•
Loss of recreational facility HEALTH and SOCIAL IMPACTS damage and loss severi	Local facility	YES	MIN		•
NATURAL ENVIRONMENT	ly level		IVITIN	UK	
Area of Impact	<1km2	YES			
Duration of Impact	<1 (wet) year	YES	•	•	•
Stock and Fauna		120	•	•	•
	Discharge from dambreak would contaminate water supplies		YES		
	used by stock and fauna. Health impacts not expected.		0	•	•
Ecosystems	Discharge from dambreak would have short term impacts on				
	ecosystems with natural recovery expected after 1 wet season.		YES		
	Remediation possible.				
Rare and endangered fauna and flora	Species exist but minimal damage expected. Recovery within	YES			
	one year.	160		•	•
NATURAL ENVIRONMENT damage and loss severity lev	el		MED	NUI	
HIGHEST DAMAGE AND LOSS SEVERITY LEV	/EL		MED	NUI	
Population at Risk (PAR)	<1				
PAR includes all those persons who would be directly					
exposed to flood waters within the dam break affected zone	if CONSEQUENCE CATEGORY =		LO	vv	
the set as the set of	OUNDEROLINOL CATLOURT =				

they took no action to evacuate.	
Completed and Reviewed By	Tom Ridgway
Date	14/09/2021

Appendix C RSF2 Geotechnical Investigation Report



Tomingley Gold Operations Pty Ltd

TGO RSF 2 Geotechnical investigation Geotechnical Factual Report

October 2021

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Appendices

Appendix A – Site Plan

- Appendix B Geology Plan
- Appendix C Geotechnical Logs and Photographs
- Appendix D Laboratory Testing

1. Introduction

1.1 Background

TGO are preparing a project modification for the development of a proposed satellite pit south of the existing mine to extend the production by 10 Mt, for nominally 7 years at a production rate of 1.5 Mtpa, which is increased from the existing 1 Mtpa.

GHD have previously undertaken an options study for TGO, utilising a multi-criteria assessment to identify the preferred site for the new Residue Storage Facility (RSF) (refer GHD report – TGO Residue Storage Facility 2 Site Options Study Report, dated December 2019). The study determined the preferred site for the new RSF to be located immediately to the south of the existing storage facility.

GHD has also undertaken a concept design for the new RSF which is comprises of a turkeys nest arrangement and utilises the southern wall of the existing embankment to optimise earthwork requirements, and assist in developing a stable landform arrangement and allow for a single closure landform at the end of mine life.

1.2 Purpose of this report

The purpose of this report is to provide factual results from the geotechnical investigation, facilitate the preliminary design of the proposed second RSF.

1.3 Scope of works

The scope of the geotechnical investigation referenced in the report comprised:

- A desktop review of publicly available information and previous geotechnical investigations nearby the proposed site.
- A programme of geotechnical fieldwork, consisting of test pit investigation and borehole investigation, with associated insitu and laboratory testing.
- Preparation of a factual geotechnical report (this report) to summarise the findings of the investigation

Additional to the geotechnical scope of works, a series of hydrogeological work items were original proposed to be undertaken in conjunction with the geotechnical scope, including;

- Installation of 6 No groundwater monitoring bores on conclusion of the drilling investigation and subsequent slug testing following development of the wells.
- Preparation of a hydrogeological assessment section within the geotechnical report (this report) to summarise any relevant hydrogeological observations, comparison to existing information, bore construction details and reporting on the initial baseline groundwater chemistry results.

The hydrogeological scope items were not completed due to cadastral ownership issues at the time of the site works and therefore delayed to a later time.

1.4 Limitations

This report has been prepared by GHD for Tomingley Gold Operations Pty Ltd and may only be used and relied on by Tomingley Gold Operations Pty Ltd for the purpose agreed between GHD and the Tomingley Gold Operations Pty Ltd as set out in section 1.2 of this report.

GHD otherwise disclaims responsibility to any person other than Tomingley Gold Operations Pty Ltd arising in connection with this report. GHD also excludes implied warranties and conditions, to the extent legally permissible.

The services undertaken by GHD in connection with preparing this report were limited to those specifically detailed in the report and are subject to the scope limitations set out in the report.

The opinions, conclusions and any recommendations in this report are based on conditions encountered and information reviewed at the date of preparation of the report. GHD has no responsibility or obligation to update this report to account for events or changes occurring subsequent to the date that the report was prepared.

The opinions, conclusions and any recommendations in this report are based on assumptions made by GHD described in this report. GHD disclaims liability arising from any of the assumptions being incorrect.

The opinions, conclusions and any recommendations in this report are based on information obtained from, and testing undertaken at or in connection with, specific sample points. Site conditions at other parts of the site may be different from the site conditions found at the specific sample points.

Investigations undertaken in respect of this report are constrained by the particular site conditions, such as the location of buildings, services and vegetation. As a result, not all relevant site features and conditions may have been identified in this report.

Site conditions (including the presence of hazardous substances and/or site contamination) may change after the date of this Report. GHD does not accept responsibility arising from, or in connection with, any change to the site conditions. GHD is also not responsible for updating this report if the site conditions change.

2. Desktop Review

2.1 General

A desktop study was undertaken to ascertain the anticipated geological conditions and potential geohazards at the site. The study entailed a review of published data available in the public domain (e.g. NSW geological maps and Google Earth imagery), site specific assessments by other consultants, and a GHD previous investigation undertaken in close proximity to the proposed site.

2.2 Geological setting

Geological Survey of NSW (GSNSW) statewide seamless geology on MinView indicates that the site is located on the Junee-Narromine volcanic belt, part of the Palaeozoic Lachlan Orogen composition of sedimentary, volcanic and intrusive rock formations of early Cambrian to early Devonian age. The Ordo-Silurian sequences that comprise the Wyoming/Caloma deposits are tight to isoclinal folding, strong axial planar cleavage with green schist metamorphic assemblages (GHD 2020a).

The area is dominated by alluvial sequences of clays, sands and gravel of Quaternary to Tertiary age, up to 50 m thick. The alluvial material dissipates to the south and north with basement outcropping. There is a well-developed weathering profile which can extend to 70 m below ground level (The Impax Group 2011).

An extract of the geology map is provided in Appendix B.

2.3 Hydrogeological Setting

There are three distinct groundwater systems within the vicinity of TGO's mining leases, as identified by The Impax Group (2011), and discussed in GHD (2020a):

- Shallow alluvium discrete, shallow alluvium (less than 10 to 20 m deep) dissects the plains surrounding the mine site along creek flow paths. These aquifers are likely to be recharged from rainfall infiltration. Groundwater within these systems is of relatively good quality, however yields are relatively low and dependent on rainfall.
- Deep alluvium up to 100 m deep and located approximately 10 km to the northwest and west of TGO. Groundwater yields are expected to be low and of poor quality. These systems may have some interaction with underlying bedrock however are likely to be primarily recharged from rainfall.
- Fractured rock the area surrounding TGO is typically underlain by shale, siltstone and chert with several fractured rock aquifers in the vicinity of the mine. Groundwater yields range from 0-3 L/s, generally less than 1.5 L/s, and water quality is poor with high salinity.

2.4 **Previous investigations**

A database search of existing subsurface information within or in close proximity to the proposed RSF site was conducted.

Mining One (2009) carried out a geotechnical investigation for the feasibility study of the Residue Storage Facility. The investigation comprised excavation of 25 No test pits to maximum 2.6 m depth within the footprint of the RSF, and cored borehole drilling to a maximum 15 m below ground level within the footprint of the Wyoming One Open Cut.

Based on the findings of the Mining One report, Cooper and Associates (2009) prepared a design for the existing RSF. The design entailed a 42 ha facility, comprising embankment with

6 m wide crest, downstream face slope angle of 1V:3H and upstream face slope angle of 1V:1.5H and 3 m wide by 2 m deep keyway.

Three recent boreholes (GHD 2020b) were found with the findings summarised in Table 2-1. The location of these boreholes are shown on the site plan presented in Appendix A.

Hole ID	Depth Encountered (min to max m)	Material Description
GBH01	0.0 to 7.5	Silty/sandy CLAY, medium to high plasticity, firm to hard, dark red/brown to grey
	Overlying	
	7.5 to 10.5	Clayey/gravelly SAND, coarse grained, grey blotched red-orange
	Overlying	
	10.5+	CLAY trace/some gravel, grey blotched orange- brown/dark red, medium plasticity, hard (residual soil)
GBH02	0.0 to 2.0	CLAY/SILT, medium to high plasticity, red spotted orange, hard
	Overlying	
	2.0 to 2.8	Clayey SAND with gravel, spotted grey, orange streaked brown, dense
	Overlying	
	2.8 to 14.0	Sandy/gravelly CLAY, medium plasticity, grey blotched orange, spotted red and white-orange
	Overlying	
	14.0+	CLAY, medium plasticity, blotch red and grey spotted black to dark red, blotched dark orange
GBH03	0.0 to 6.0	Silty/Sandy CLAY, low to medium plasticity, very stiff to hard, red and white, grey and brown
	Overlying	
	6.0 to 6.5	SAND with clay and gravel, dense, red spotted, red-grey
	Overlying	
	6.5 to 8.9	Gravelly/sandy CLAY, medium plasticity, grey spotted red to grey blotch red-brown
	Overlying	
	8.9 to 11.0	CLAY/SILT, medium plasticity, pale brown
	Overlying	
	11.0+	SILTSTONE/MUDSTONE, pale yellow/green, vertical laminations

Table 2-1 Summary of subsurface conditions at RSF1

3.1 General

An intrusive geotechnical ground investigation was carried out between 17 November and 5 December 2020, comprising the following activities:

- Mechanical excavation of 20 No test pits (TP01 to TP20) in vicinity of the proposed reservoir storage area.
- Drilling of 10 No geotechnical boreholes (BH01 to BH10) within the alignment of the proposed embankment foundation to a target depth of 20 m

The location of test pits and boreholes are shown in Appendix A.

Fieldwork was carried out under full time supervision of an experienced engineer/geologist from GHD and in accordance with AS 1726-2017 (Geotechnical Site Investigations) and standard GHD procedures.

Visual and tactile logging of recovered samples was undertaken in accordance with the GHD soil and rock logging procedures.

The locations and elevations for each test location was surveyed by TGO mine surveyors and provided to GHD following completion of the works. The relevant data is provided on the respective geotechnical log sheets.

3.2 Test Pits

Test pits were excavated to a maximum depth of 4.0 m below ground level, using a 21.5 tonne excavator provided and operated by TGO contractors.

On completion of test pit logging and sampling, the test pits were backfilled with the excavated material and tamped with the base of the excavator bucket.

Descriptive logs indicating the observed soil profile in each pit and test pit photographs are presented in Appendix C and are preceded by explanatory notes summarising general nomenclature and symbols used.

A summary of the test pits locations is included in Table 3-1.

Coordin		inates			
Test Pit ID	(MGA 94, Zone 55)		Depth Excavated	Termination Reason	
	Easting (m)	Northing (m)	(m)		
TP01	613678.8	6393170	3.2	Effective refusal	
TP02	613703.7	6393315	3.5	Limit of Hole	
TP03	613505.9	6393348	3.5	Target Depth	
TP04	613534.7	6393509	2.1	Effective refusal	
TP05	613554	6393622	2.0	Effective refusal	
TP06	613299.9	6393550	3.7	Limit of Hole	
TP07	613318.8	6393661	2.0	Effective refusal	
TP08	613085.4	6393690	3.1	Limit of Hole	

Table 3-1 Test Pit Summary

· Test Pit ID	Coordinates (MGA 94, Zone 55)		Depth Excavated	Termination Reason	
	Easting (m)	Northing (m)	(m)	Termination Reason	
TP09	612879.5	6393721	2.4	Limit of Hole	
TP10	612870.5	6393621	3.0	Effective refusal	
TP11	613067.7	6393588	3.2	Effective refusal	
TP12	613276.8	6393385	4.0	Limit of Hole	
TP13	613043.5	6393417	3.5	Effective refusal	
TP14	613252.2	6393237	4.0	Target Depth	
TP15	613482.5	6393207	2.8	Effective refusal	
TP16	613730.6	6393476	3.7	Effective refusal	
TP17	613757.1	6393589	3.3	Effective refusal	
TP18	613015.1	6393279	3.2	Effective refusal	
TP19	612819.5	6393320	3.0	End of Hole, rock encountered	
TP20	612839.7	6393455	3.5	End of hole, rock encountered	

3.1 Boreholes

The boreholes were drilled using a track mounted drill rig own and operated by Numac Drilling Services Australia Pty Ltd. The boreholes were advanced using sonic drilling techniques from ground surface to achieve a maximum depth of 20 m below ground level.

The boreholes were grouted to ground surface on extraction of drilling rods.

The geotechnical logs and core tray photos are presented in Appendix C, and are preceded by explanatory notes summarising general nomenclature and symbols used.

A summary of the boreholes is presented in Table 3-2.

Table 3-2 Borehole Summary

. BH ID	Coordinates (MGA 94, Zone 55)		Collar RL (mAHD)	Depth (mbgL)	
	Easting (m)	Northing (m)	(survey data)		
BH01	613566.3	6393074	264.7	20	
BH02	613269.7	6393126	263.8	20	
BH03	612974.4	6393189	263.4	20	
BH04	612790.1	6393360	263.8	20	
BH05	612841.4	6393664	264.6	20	
BH06	612974.4	6393714	265.1	20	
BH07	613320.9	6393651	265.4	20	
BH08	613552.7	6393609	265.9	20	
BH09	613785.6	6393577	266.8	20	

. BH ID		oordinates A 94, Zone 55)	Collar RL (mAHD)	Depth (mbgL)	
	Easting (m)	Northing (m)	(survey data)		
BH10	613739.8	6393286	265.5	20	

3.2 In-situ Testing

3.2.1 Pocket Penetrometer

Pocket penetrometer testing was undertaken on the ends of undisturbed (U63) samples obtained from the boreholes, and at regular depth intervals in the walls of the test pits to assess the in-situ consistency of the shallow soils and aid in identification of any potentially soft or low strength zones.

The results of pocket penetrometer testing are presented on the associated test pit and borehole logs in Appendix C.

3.2.2 Dynamic Cone Penetration (DCP)

DCP testing was undertaken at select locations to assist in determining the density of the near surface material. The results of the DCP tests are presented on the test pit logs in Appendix C.

3.2.3 Standard Penetration Tests (SPT)

Standard Penetration Tests (SPT) were undertaken within the boreholes at approximately 1.5 m intervals (where appropriate) to assess the consistency of the soils and to recover disturbed samples. The uncorrected results of the SPTs are plotted on Figure 3-1 and presented on the associated borehole logs in Appendix C.

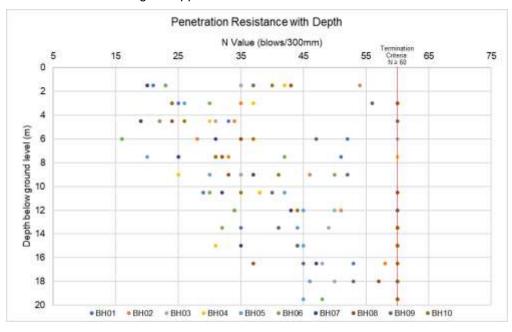


Figure 3-1 Relationship between SPT N value and depth

4. Site Conditions

4.1 Surface Conditions

The proposed RSF2 is located on farmland immediately south of the existing RSF. The vegetation at the site comprises paddocks of knee-high grass and weed-like vegetation. A small farm dam (approx. 0.35 Ha) is located within the south east corner of the proposed RSF. A smaller dam (0.3 Ha) is situated south of the downstream toe in the south west corner.

The RSF area generally had a gentle slope (average of 0.2°) from north east towards the south west.

4.2 Subsurface Conditions

The subsurface conditions encountered within the test pits and boreholes generally comprised topsoil overlying alluvial (sandy) clays with varying proportions of gravels. The material types are summarised in the sections below.

Detailed descriptions of the subsurface materials and conditions encountered during the investigations are provided in the geotechnical test pit and borehole logs, presented in Appendix C and should be referred to for detail at each test location. Photographs of test pits and core trays are included after the corresponding log.

4.2.1 Topsoil

The topsoil encountered across the site typically comprised low plasticity, sandy clay with rootlets. The colour of the topsoil varied between brown and orange brown and with a field moisture condition less than the plastic limit (i.e. dry)

4.2.2 Alluvial

The alluvial material encountered to 20 m depth comprised very stiff to hard, sandy clay/clay with varying proportions of gravels. The clay was typically of low to medium plasticity with a field moisture condition generally equal to, or less than, the plastic limit (i.e. dry to moist).

The alluvial clays were observed to vary in colour between (pale to dark) brown and orange, and grey.

4.3 Groundwater

Groundwater was encountered in BH01, BH04 and BH08 at 7.5 m (RL 257.2 m), 8.0 m (RL 255.8 m) and 7.5 m (RL 258.4 m) below ground level respectively.

5. Laboratory testing

5.1 General

Laboratory testing was undertaken on samples collected from the test sites to confirm field logging and assess material properties.

The following tests were undertaken according to the relevant Australian Standard test methods:

- Particle Size Distribution, including Hydrometer
- Atterberg Limits
- Field Moisture Content
- Bulk Density
- Emerson Class (dispersion)
- Standard Compaction Tests (98% MDD)
- Constant Head Permeability (Triaxial Cell)
- X-ray Computed Tomography (XCT) scan
- Single Stage Triaxial Consolidated Undrained (CU) Strength Tests with Pore Pressure Measurement
- One dimensional consolidation (Oedometer) tests

Testing was conducted in the NATA accredited Fugro AG Laboratory in Perth, and GHD's NATA accredited facility in Sydney.

5.2 Characteristic Tests

Tabulated summaries of the respective soil laboratory test results are presented in Table 5-1 to Table 5-4. Charts plotting all particle size distribution and Atterberg Limit results are presented in Figure 5-1 and Figure 5-2, respectively. The relationships between field moisture content and Atterberg Limits for samples are presented in Figure 5-3.

The laboratory test certificates related to the tests summarised in Table 5-1 to Table 5-4 is included in Appendix D.

Table 5-1 Soil Classification Testing

Test ID	Dopth (m)		MC	Plast	icity (%)	LS	Particle Size Distribution (%)		6)			
Test ID	Depth (m)	Soil Description	(%)	LL	PI	(%)	С	lay	Silt	Sand	Gravel	Emerson Class
TP01	1.0 - 2.0	Sandy CLAY, trace gravel	10.0	31	17	-	5	59	3	37	4	2
TP03	2.0 - 3.0	CLAY with sand, trace gravel	16.9	50	34	-	32	56	1	1	1	2
TP04	1.0 - 2.0	CLAY with sand, trace gravel	13.9	56	38	-	30	33	2	28	9	2
TP07	1.0 - 2.0	CLAY with sand, trace gravel	17.7	48	30	-	6	69	2	23	8	2
TP10	0.0 - 1.0	Sandy CLAY, trace gravel	11.5	29	15	-	28	40	3	31	1	2
TP11	1.0 - 2.0	Sandy CLAY	17.4	56	40	—	3	85	27	36	2	2
TP11	2.0 - 3.0	Sandy CLAY with gravel	-	-	-	-		41		46	13	-
TP12	3.0 - 4.0	Sandy CLAY	14.1	31	17	-		64		36	0	2
TP14	1.0 - 2.0	Sandy CLAY, trace gravel	14.1	44	29	-		44		54	2	2
TP17	1.0 - 2.0	Sandy CLAY, trace gravel	14.4	50	34	-		66		29	5	2
TP18	1.0 - 2.0	CLAY with sand, trace gravel	14.7	54	37	-	3	85	35	27	3	2
TP19	1.0 - 2.0	Sandy CLAY, trace gravel	8.1	41	27	-		39		54	7	-
BH01	1.5 and 3.0*	Sandy CLAY trace gravel	-	-	-	-		95		35	5	-
BH01	3.7 - 3.89	CLAY with sand	22.7	48	29	9.0	1	7	68	15	-	-
BH01	9.0 and 13.5*	CLAY, trace sand & gravel	-	-	-	-		86		12	2	-
BH01	9.68 - 9.84	CLAY with sand	17.0	54	48	10	3	81	48	19	2	
BH02	1.5 and 3.0*	CLAY with sand, trace gravel	-	-	-	_		77		22	1	-

Test ID	Depth (m)		MC	Plast	icity (%)	LS	Particle Size Distribution (%)		6)		
Testid	Depth (m)	Soil Description	(%)	LL	PI	(%)	Clay	Silt	Sand	Gravel	Emerson Class
BH02	6.0, 7.5 and 9.0*	CLAY with sand, trace gravel	-	-	-	-	81		17	2	-
BH03	5.8, 9.0 and 11.8*	Sandy CLAY, trace gravel	-	-	-	-	40	26	30	4	-
BH03	15.0, 18.0 and 19.5*	CLAY with sand, trace gravel	-	-	-	-	45	38	14	3	-
BH04	6.0, 9.0 and 15*	CLAY trace sand	-	-	-	-	86		13	1	-
BH05	3.0 and 4.5*	CLAY with sand, trace gravel	-	-	-	-	71		27	2	-
BH05	5.06 - 5.22	CLAY with sand	19.1	34	19	6	20	65	15	-	-
BH05	7.5, 10.0 and 16.5*	Sandy CLAY, trace gravel	-	-	-	-	67		32	1	-
BH06	1.5 and 3.0*	CLAY with sand, trace gravel	-	-	-	-	70		27	3	-
BH06	6.65 - 6.84	Sandy CLAY	12.2	28	13	6	13	42	44	-	-
BH06	10.7, 12.0 and 13.5*	CLAY with sand, trace gravel	-	-	-	-	83		16	1	-
BH07	3.52 - 3.71	CLAY with sand	17.5	44	27	10	26	48	26	-	-
BH07	4.5 and 6.0*	CLAY trace sand	-	-	-	-	91		9	0	-
BH07	13.5, 16.5 and 19.5*	CLAY with sand, trace gravel	-	-	-	_	34	29	29	-	-
BH08	1.5 and 3.0*	Clayey SAND with gravel	_	_	-	_	31		50	19	-

Test ID	Depth (m)	Soil Description	MC	Plast	icity (%)	LS	Particle Size Distribution (%)		6)	Emerson Class	
Test ID			(%)	LL	PI	(%)	Clay	Silt	Sand	Gravel	
BH08	5.16 - 5.35	CLAY trace sand	20.1	43	26	11.5	30	59	11	-	-
BH08	7.5 - 7.95	Sandy CLAY trace gravel	-	-	-	-	62		35	3	-
BH09	1.5, 3.0, 4.5 and 6.0*	Sandy CLAY trace gravel	-	-	-	_	44		46	10	-
BH09	15.0,18. 0 and 19.5*	CLAY with sand, trace gravel	-	-	-	-	78		20	2	-
BH10	1.5, 3.0, 6.0 and 9.0*	Sandy CLAY trace gravel	-	-	-	_	41		46	13	-
BH10	13.5,15. 0 and 18.0*	CLAY trace sand and gravel	-	-	-	_	88		11	1	-

Note: MC = Moisture Content; LL = Liquid Limit; PI = Plasticity Index; LS = Linear Shrinkage, * Multiple depths indicate blended samples

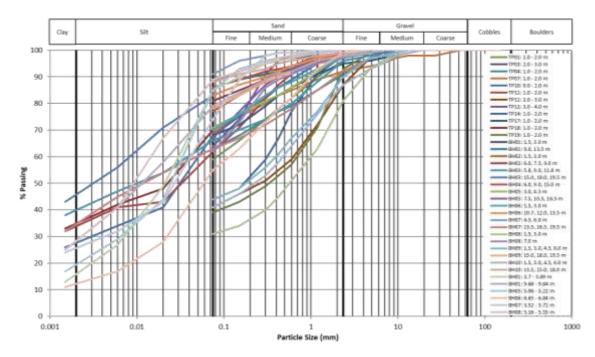


Figure 5-1 Particle Size Distribution Plot

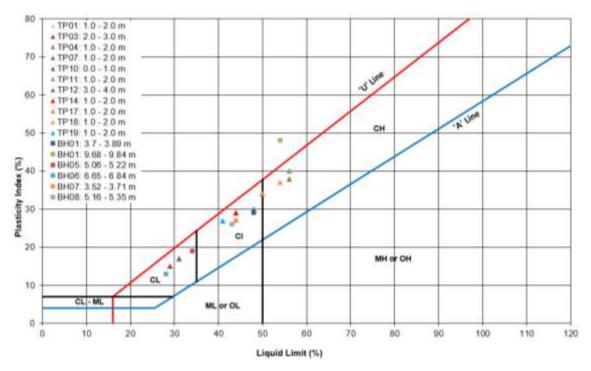
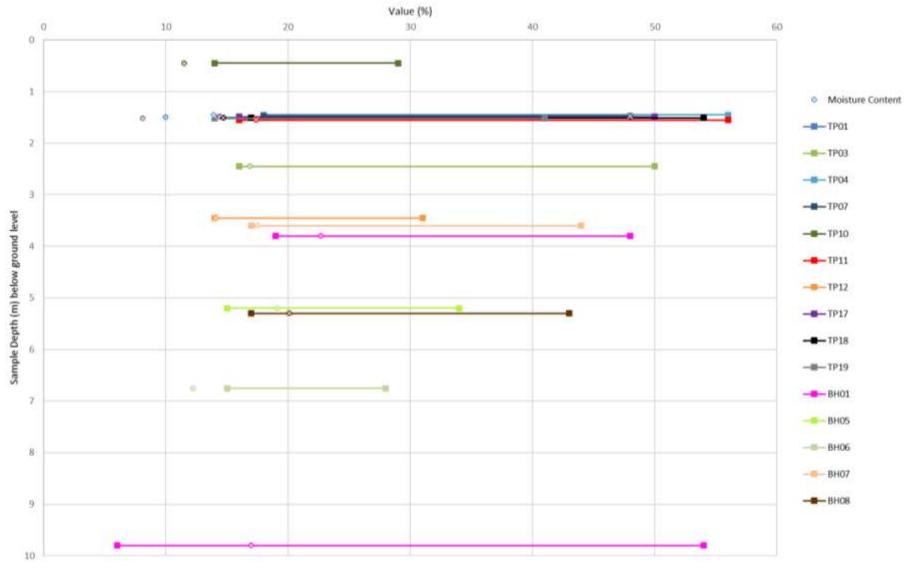


Figure 5-2 Atterberg Limits Plot



Relationship between Atterberg Limits (LL & PL) and Moisture Content

Figure 5-3 Field moisture content relative to Atterberg Limits with depth

Test ID	Depth (m)	Maximum Dry Density (t/m³)	Optimum Moisture Content (%)	Field Moisture Content (%)
TP01	1.0 - 2.0	1.84	14.5	10.0
TP03	2.0 - 3.0	1.66	20.0	16.9
TP04	1.0 - 2.0	1.70	18.0	13.9
TP07	1.0 - 2.0	1.70	21.0	17.7
TP10	0.0 - 1.0	1.81	15.0	11.5
TP11	1.0 - 2.0	1.65	20.0	17.4
TP12	3.0 - 4.0	1.77	16.5	14.1
TP14	1.0 - 2.0	1.79	16.0	14.1
TP17	1.0 - 2.0	1.67	18.0	14.4
TP18	1.0 - 2.0	1.62	18.5	14.7

Table 5-2 Standard Compaction Results

Table 5-3 Material Density

Test ID	Depth (m)	Soil Particle Density (g/cm ³)	Bulk Density (t/m ³)	Dry Density (t/m³)	Field Moisture Content (%)
TP01	1.0 - 2.0	2.53	2.038	1.852	10.0
TP03	2.0 - 3.0	2.55	1.962	1.679	16.9
TP04	1.0 - 2.0	2.50	2.075	1.822	13.9
TP07	1.0 - 2.0	2.48	1.829	1.554	17.7
TP10	0.0 - 1.0	2.46	1.892	1.697	11.5
TP12	3.0 - 4.0	2.50	1.936	1.697	14.1
TP14	1.0 - 2.0	2.61	2.051	1.798	14.1
TP18	1.0 - 2.0	2.50	1.932	1.685	14.7
BH01	3.51 - 3.7	2.68	-	1.72	19.7
BH01	3.7 - 3.89	2.70	-	1.62	22.7
BH01	9.68 - 9.87	2.75	-	1.82	17.0
BH05	5.03 - 5.22	2.71	-	1.71	19.1
BH06	6.65 - 6.84	2.68	-	1.86	12.2
BH07	3.52 - 3.71	2.70	-	1.71	17.5
BH08	5.16 - 5.35	2.71	-	1.40	20.1

Table 5-4 Laboratory Permeability Testing

Test ID	Depth (m)	Description	Coefficient of permeability, k (m/sec)
TP03	2.0 - 3.0	CLAY trace sand	2 x 10 ⁻¹⁰
TP10	0.0 – 1.0	Sandy CLAY, trace gravel	2 x 10 ⁻¹⁰
TP11	1.0 – 2.0	Sandy CLAY	4 x 10 ⁻¹¹
TP18	1.0 – 2.0	Sandy CLAY, trace gravel	1 x 10 ⁻¹⁰

5.3 Oedometer Tests

Consolidation testing was undertaken on four undisturbed samples. The results of the tests are summarised in Figure 5-4 and available test certificates are included in Appendix D.

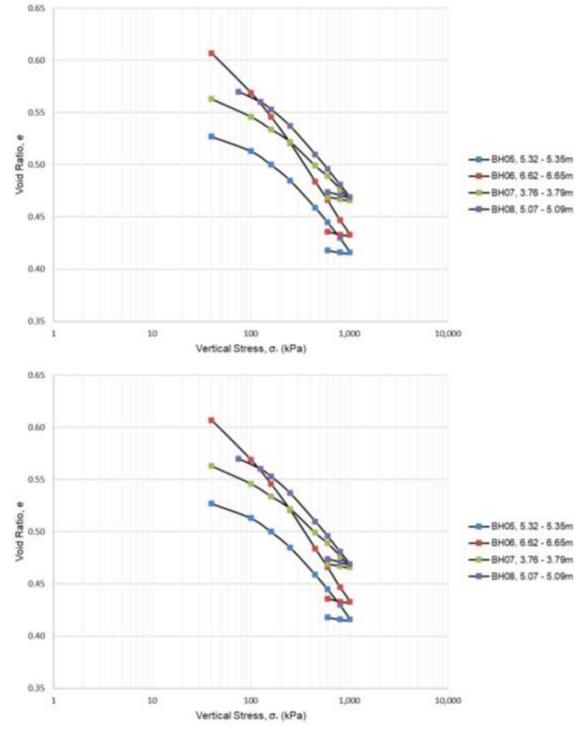


Figure 5-4 Consolidation test compressibility curves

5.4 Triaxial Tests

Four multi-stage and three single stage isotropically consolidated triaxial (CIU) tests with measurement of pore water pressure, have been completed on undisturbed samples. Samples were tested with the consideration of the future consolidation pressure. Available test certificates are included in Appendix D and test results are summarised in Table 5-5.

Table 5-5 Triaxial Test Results

Sample Locations	Consolidati on Pressure, po' (kPa)	Maximum Stress Ratio (M=q/p')	Effective Peak Friction Angle, φ (°)	Peak Undrained Shear Strength, Su	Shear Strength Ratio, s₀/po'
	po (a)			(kPa)	
	130	1.61	39.40	126	0.97
BH01 9.68 - 9.84 m	330	1.24	30.92	234	0.71
	530	1.09	27.47	318	0.6
	120	1.39	34.35	102	0.85
BH05 5.06 - 5.22 m	300	1.21	30.23	180	0.6
	480	1.15	28.85	269	0.56
	120	1.51	37.10	119	0.99
BH06 6.65 - 6.84 m	300	1.38	34.12	240	0.8
	480	1.33	32.98	379	0.79
D U la c	120	1.66	40.55	134	1.12
BH08 5.16 - 5.35 m	300	1.33	32.98	228	0.76
	480	1.18	29.54	302	0.63
BH01 3.51 - 3.70 m	100	1.5	36.87	119	1.19
BH01 3.70 - 3.89 m	200	1.39	34.35	160	0.8
BH07 3.52 - 3.71 m	400	1.23	30.69	240	0.6

6. Information about this report

The report contains the results of a geotechnical investigation conducted for a specific purpose and client. The results should not be used by other parties, or for other purposes, as they may contain neither adequate nor appropriate information. In particular, the investigation does not cover contamination issues unless specifically required to do so by the client.

Test hole logging

The information on the test hole logs (boreholes, test pits, exposures etc.) is based on a visual and tactile assessment, except at the discrete locations where test information is available (field and/or laboratory results). The test hole logs include both factual data and inferred information. Reference should be made to the relevant sheets for the explanation of logging procedures (Soil and Rock Descriptions, Core Log Sheet Notes etc.).

Groundwater

Unless otherwise indicated, the water levels presented on the test hole logs are the levels of free water or seepage in the test hole recorded at the given time of measuring. The actual groundwater level may differ from this recorded level depending on material permeability (i.e. depending on response time of the measuring instrument). Further, variations of this level could occur with time due to such effects as seasonal, environmental and tidal fluctuations or construction activities. Confirmation of groundwater levels, phreatic surfaces or piezometric pressures can only be made by appropriate instrumentation techniques and monitoring programmes.

Interpretation of results

The discussion or recommendations contained within this report normally are based on a site evaluation from discrete test hole data. Generalised, idealised or inferred subsurface conditions (including any geotechnical cross-sections) have been assumed or prepared by interpolation and/or extrapolation of these data. As such these conditions are an interpretation and must be considered as a guide only.

Change in conditions

Local variations or anomalies in the generalised ground conditions do occur in the natural environment, particularly between discrete test hole locations. Additionally, certain design or construction procedures may have been assumed in assessing the soil structure interaction behaviour of the site. Furthermore, conditions may change at the site from those encountered at the time of the geotechnical investigation through construction activities and constantly changing natural forces.

Any change in design, in construction methods, or in ground conditions as noted during construction, from those assumed or reported should be referred to GHD for appropriate assessment and comment.

Geotechnical verification

Verification of the geotechnical assumptions and/or model is an integral part of the design process - investigation, construction verification, and performance monitoring. Variability is a feature of the natural environment and, in many instances, verification of soil or rock quality, or foundation levels, is required. There may be a requirement to extend foundation depths, to modify a foundation system or to conduct monitoring as a result of this natural variability. Allowance for verification by geotechnical personnel accordingly should be recognised and programmed during construction.

Foundations

Where referred to in the report, the soil or rock quality, or the recommended depth of any foundation (piles, footings etc.) is an engineering estimate. The estimate is influenced, and perhaps limited, by the fieldwork method and testing carried out in connection with the site investigation, and other pertinent information as has been made available. The material quality and/or foundation depth remains, however, an estimate and therefore liable to variation. Foundation drawings, designs and specifications should provide for variations in the final depth, depending upon the ground conditions at each point of support, and allow for geotechnical verification.

7. References

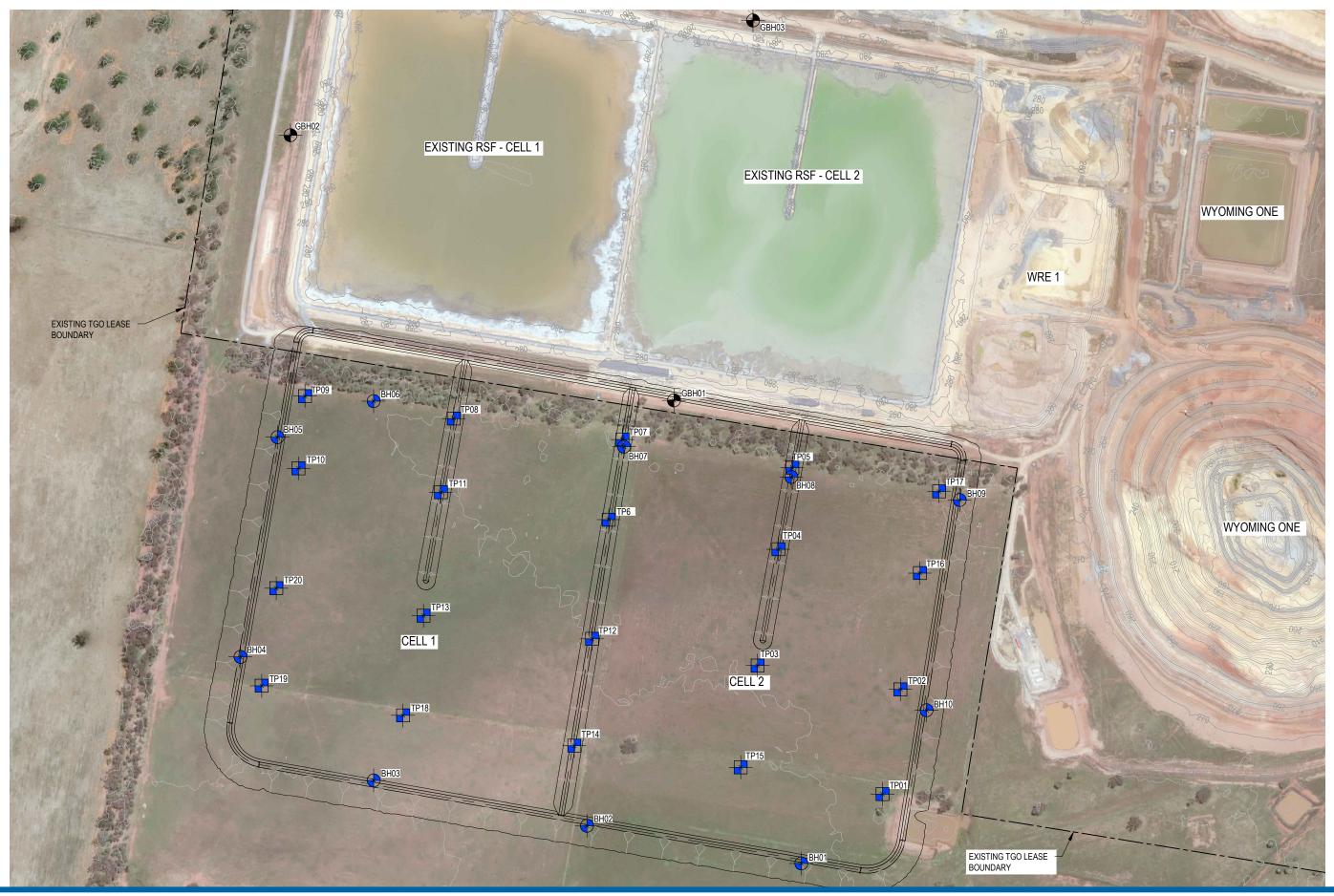
ANCOLD 2019, Guidelines on Tailings Dams, Planning, Design, Construction Operation and Closure
Australian Standard, AS 1726-2017. Geotechnical site investigations. Standards Australia
Australian Standard, AS 1289 – Method of testing soils for engineering purposes. Standards Australia
Cooper and Associates, 2009 Tomingley Gold Project Residue Management Design Report
GHD 2020a, Tomingley Gold Operations Pty Ltd Processing Plant 2020 Groundwater Review
GHD 2020b, TGO RSF Stage 6 Raise Detailed Design Geotechnical Investigation Factual Report
Mining One 2009, Tomingley Gold Project Feasibility Study Factual Geotechnical Report for Residue

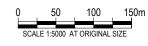
The Impax Group 2011, *Tomingley Gold Project Groundwater Assessment,* prepared for R.W Corkery & Co Pty Ltd on behalf of Alkane Resources Ltd

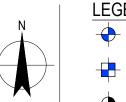
Appendices

GHD | Report for Tomingley Gold Operations Pty Ltd - TGO RSF 2 Geotechnical investigation, 12538404 | 8

Appendix A – Site Plan







LEGEND: BOREHOLE

> TEST PIT EXISTING BOREHOLE



TOMINGLEY GOLD OPERATIONS RSF 2 GEOTECHNICAL INVESTIGATION

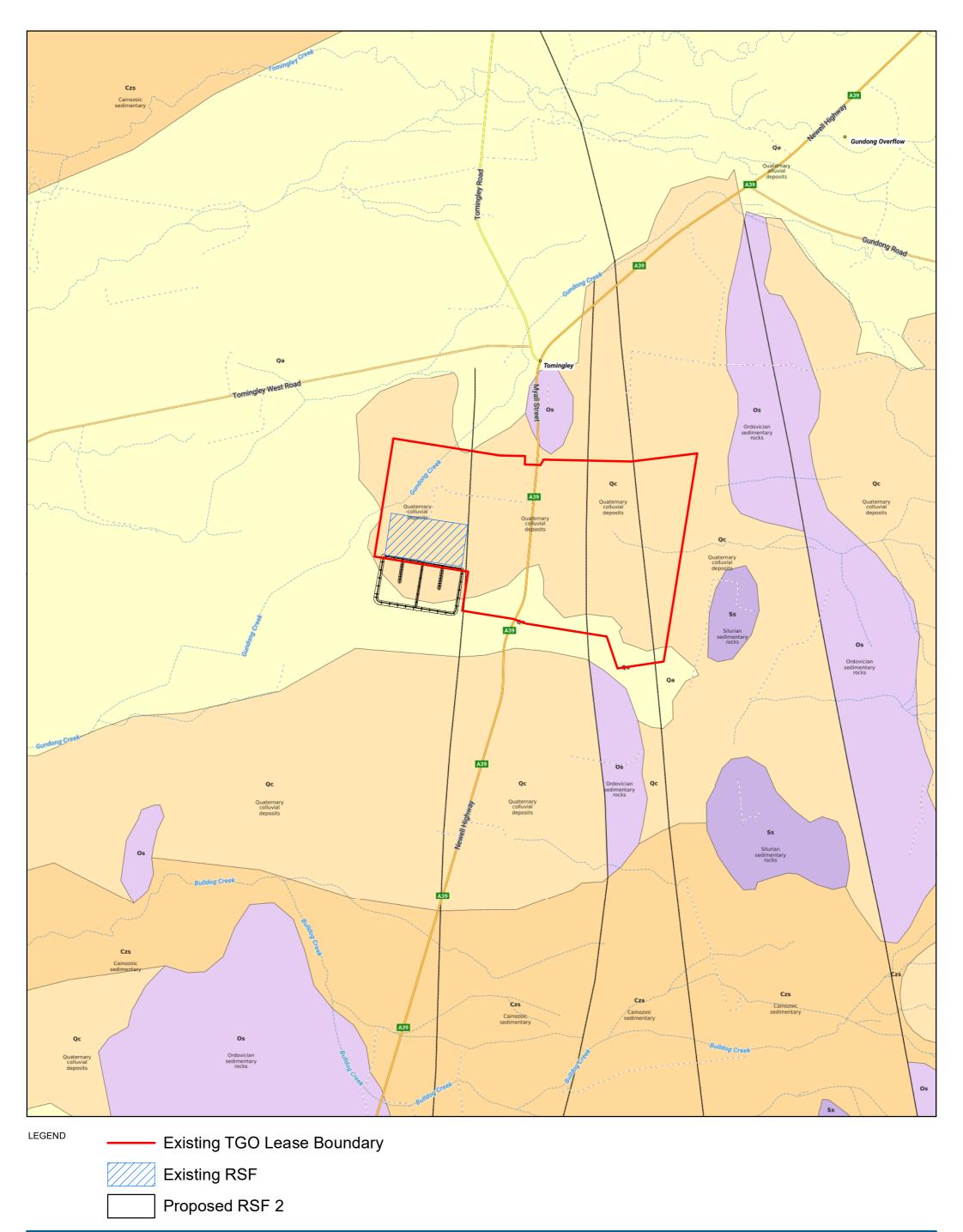
TEST LOCATIONS

Cad File No: G:\

Job Number | 12538404 Revision A Date MAR 2021 Appendix A

2 Salamanca Square Hobart TAS 7000 Australia T 61 3 6210 0600 F 61 3 6210 0601 E hbamail@ghd.com W www.ghd.com

Appendix B – Geology Plan





2 Salamanca Square, Hobart Tasmania 7000 Australia T 61 3 6210 0600 E hbamail@ghd.com W www.ghd.com

G:322/12517944/GISWaps\Deliverables\12517944_Geology_20200720.mxd 2 Salamanca Square, Hobart Tasmania 7000 Australia T © 2021. Whilst every care has been taken to prepare this map, GHD (and DATA CUSTODIAN) make no representations or warranties about its accuracy, reliability, completeness or suitability for any particular purpose and cannot accept liability and responsibility of any kind (whether in contract, tort or otherwise) for any expenses, losses, damages and/or costs (including indirect or consequential damage) which are or may be incurred by any party as a result of the map being inaccurate, incomplete or unsuitable in any way and for any reason. Data source: Maps SIX NSW, Basemap Image, 2020. GHD, Borehole and Test Pit Locations, 2020; Porposed Tailings Dam, 2020. Created by:npolmear

 $\label{eq:product} \textbf{Appendix} \ \textbf{C} - \text{Geotechnical Logs and Photographs}$

Pr	ien oje ocat	ct :		G	eote	chni		Operations vestigation V						LOC	ATION	NO. BHO SHEET 1 OF	
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1					- - - - - - - - - - - - - - - - - - -	1.0		Sandy CLAY, orange brown, lo medium grained sand (Alluvial)	w to medium plasticity, organics, fine to	CI	D	F					
								Sandy CLAY, trace gravel, med coarse grained sand, w ~ PL (/	dium plasticity, orange brown, fine to Alluvial)	- - - -	D-M	St	PP SPT	1.5m: PP = 4 1.5-1.95m: S N = 21	100 kPa (UCS) 3PT = 5,11,10	1	
2					-	2 <u>.0</u>		pale brown to brown				St- VSt	U75	U75-01 @ 1. R: 350 / 400	.95 - 2.35 m mm		
					- - - - 262.0								PP PP		600 kPa (UCS 450 kPa (UCS		
3					-	3 <u>.0</u> 	· · · · · · · · · · · · · · · · · · ·						PP SPT		800 kPa (UCS) 9PT = 9,11,14	1	
					- 261.0	1							U75	U75-02 @ 3. R: 400/400 n	.5 - 3.9 m nm		
4					-	4 <u>.0</u>	· ·	pale orange brown mottled pale	e grey			VSt	PP	3.9m: PP >60	00 kPa (UCS)		
					- - - - - - - -		· · · · · · · · · · · · · · · · · · ·	CLAY with sand, trace gravel, or grained, angular to sub angular	orange brown, low plasticity, fine to mediu r gravels, polymictic, w ~ PL (Alluvial)	m T CL -	D-M	н	SPT	4.5-4.95m: S N = 33	SPT = 12,14,19)	
5	Sonic				- - - - - - - - - - - - - - - - - - -	5.0	· · · · · · · · · · · · · · · · · · ·	CLAY with sand and gravel, gre and gravel is fine to coarse gra gravels up to 40 mm, no gradin	ey and orange mottled, low plasticity, sand ained, angular to sub angular polymictic ng observed, w ~ PL (Alluvial)		D-M	н	PP	5.0 m: PP >6	600 kPa (UCS)		
6					- - - -	- 6 <u>.0</u> -							PP SPT	6.0 m: PP > 6 6.0-6.45 m: S N = 52	600 kPa (UCS SPT =15,22,30)	
				020	- - - 258.0 -			CLAY, medium brown mottled g	grey and orange, w < PL (Alluvial)		D-M	Н	PP	6.5 m: PP >6	600 kPa (UCS)	•	
7				¹ 24/11/2020	- - - -	7 <u>.0</u> - - -							PP	7.0 m: PP >6	600 kPa (UCS)	•	
				¥	- <u>25</u> 7.0 -			orange mottled, w > PL (Alluvia	· 		w	н	SPT	7.5-7.95 m: S N = 51	SPT = 9,20,31		
8					- - - - - - - - - - - - - - - - - - -			CLAY with sand, trace gravels coarse grained angular to sub- mm, w ~ PL (Alluvial)	and cobbles, brown, high plasticity, fine to angular gravels, polymictic, cobbles to 70	CH	M	Н	PP	8.0 m: PP > 6	600 kPa (UCS)	
9					- - - -	9 <u>.0</u>							PP SPT	9.0 m: PP > 6 9.0-9.45 m: 5 N =- 35	600 kPa (UCS SPT = 7,16,19)	
					- 255.0 -	111							U75	U75-03 9.5 - R: 400/400 n	nm		
10					- - - -	10 <u>.0</u>		CLAY, trace sand and gravel, p coarse grained sand, fine grain	pale grey, medium to high plasticity, fine to ned gravel, w ~ PL (Alluvial)	- Сн	M	VSt	PP		600 kPa (UCS		
11					254.0 - -	- - 11.0							PP SPT	10.5 m: PP > 10.5-10.95 m N = 29	• 600 kPa (UC n: SPT = 7,11,	18	
					sheet		-	GHD P	ty Ltd nca Square Hobart TAS 7001 2100 600 F: 61 3 6210060	Hohart	_				Job N	0.	
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SCALE (m) Method	Hole Support	Run	Water	KT (m)	Depth (m)	Graphic Log	Soil N Other Minor Components	Description ame (USC Symbol) s, Plasticity or Particle Characteristic ondition, Consistency, Structure	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Insitu	s/Observations test results	
12				- <u>225</u> 3.0 - <u>225</u> 3.0 						M	VSt	SPT U75 PP PP SPT	11.0 m: PP > 600 kPa (12.0 m: PP > 600 kPa (12.0-12.45 m: SPT = 1(N = 34 U75-04 12.6 - 12.9 m R: 250/300 mm 12.9 m: PP > 600 kPa (13.25 m: PP = 500 kPa 13.5 m: SPT = 1' N = 34	UCS)),15,19 UCS) (UCS)	
14 15 16	250			250.0 2 2 2 2 2 2 2 2 4 9.0 2 4 9.0	14 <u>.0</u> 		CLAY with sand, trace gravels fine to coarse grained gravels cobbles up to 40 mm, no grad PL (Alluvial)	s and cobbles, orange brown, high plastic dominantly quartz, angular to sub angui ing observed, polymictic, grey mottled, w	city, CH ar /~	M	н	PP PP SPT	14.0 m: PP > 600 kPa (14.5 m: PP > 600 kPa (15.0-15.45 m: SPT = 2 ⁻ 16.5-16.95 m: SPT = 15	UCS) 1,31,30*/135mm	
17 18 19				<u>24</u> 8.0 2 <u>4</u> 7.0 <u>2</u> 46.0	17 <u>.0</u> - - - - - - - - - - - - - - - - - - -		18.0 m becomes very brittle/fr	riable				SPT	N = 53 18.0-18.45 m: SPT = 18		
20 —				<u>24</u> 5.0 <u>24</u> 5.0 <u>24</u> 5.0 <u>24</u> 4.0	20.0		END OF HOLE at 20.0 m, red	quired depth							
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SCALE (m)	Method	Hole Support	Run	Water	RL (m)	Depth (m)	Graphic Log	Soil Na Other Minor Components	Description ame (USC Symbol) , Plasticity or Particle Characteristics, ondition, Consistency, Structure	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests		s/Observations test results	
					-			TOPSOIL: CLAY with sand, br fine to medium grained sand, c	own to orange brown, medium plasticity, organics, grass, rootlets, w < PL	CI	D	F				
					- - 263.0		× × ×	CLAY with Sand, trace gravel, grained sand, fine grained gravel	brown, medium plasticity, fine to medium vel w < PL (Alluvial)	- - - -		Η-				
1					-	1 <u>.0</u>	· · · ·	thin interbedded sandy layers a	at 1.0 m and 1.5 m				PP	1.0 m: PP > 600 kPa (U	ICS)	
					-		· ·						PP SPT	1.4 m: PP > 600 kPa (U 1.5-1.95 m: SPT = 18,2	ICS)	
					- <u>26</u> 2.0 -								5-1	N = 54	5,29	
2					- - - - -	2 <u>.0</u>	· · · · · · · · · · · · · · · · · · ·	plasticity, fine to medium grain	s and Cobbles, brown grey, medium ed sand, sub rounded to sub angular ig, cobbles to 30 mm, polymictic, orange	- - - -	м	H H		2.0 m: PP > 600 kPa (U	ICS)	
3					261.0 - - - - -	3 <u>.0</u> 	· · · · · · · · · · · · · · · · · · ·						SPT	3.0-3.45 m: SPT = 10,1 N = 35	6,19	
4					- - 2 <u>6</u> 0.0 - -	4.0	· · · · · · · · · · · · · · · · · · ·	CLAY with Sand, brown to oran grained, w ~ PL (Alluvial)	nge brown, medium plasticity, sand is fine	- - c i -	м	H	PP	3.6 m: PP > 600 kPa (U	ICS)	
-				-		5.0	· · · · · · · · · · · · · · · · · · ·	Sandy CLAY, orange brown, m sand, w ~ PL (Alluvial)	edium plasticity, fine to medium grained	- - - c i -	м	H.	SPT	4.5-4.95 m: SPT = 14,1 N = 34	6,18	
5	0				-		· · · · · · · · · · · · · · · · · · ·		nge brown, medium plasticity, fine to coarse grained angular to sub angular grey mottled, w ~ PL (Alluvial)		М	VSt- H	U75 PP	U75-01 5.0 - 5.3 m R: 300/300 5.3 m: PP = 450 kPa (U	(CS)	
	Sonic				- - 2 <u>5</u> 8.0		· · ·						PP	5.6 m: PP = 450 kPa (L	,	
6						6 <u>.0</u>							PP SPT	6.0 m: PP = 500 kPa (l 6.0-6.45 m: SPT = 7,11 N = 28	JCS) ,17	
7					2 <u>5</u> 7.0 - - - - -	- 7 <u>.0</u> - - -	· · · · · · · · · · · · · · · · · · ·						PP	7.0 m: PP = 550 kPa (L	ICS)	
					- - 2 <u>5</u> 6.0		· · ·						PP SPT	7.5 m: PP = 600 kPa (U 7.5-7.95 m: SPT = 8,13 N = 33	CS) ,20	
8					-	8 <u>.0</u> - -	· · · ·						PP	8.0 m: PP = 600 kPa (l	JCS)	
					- - 2 <u>5</u> 5.0	1111	· · ·						PP	8.5 m: PP > 600 kPa (U	ICS)	
9					-	9 <u>.0</u> - -	· · ·						SPT	9.0-9.45 m: SPT = 10,2 N = 46	0,26	
					- - 2 <u>5</u> 4.0	- - - - 10.0							PP	9.5 m: PP = 550 kPa (U	ICS)	
10					- - - - - -								PP SPT	10.0 m: PP > 600 kPa (10.4 m: PP > 600 kPa (10.4-10.85 m: SPT = 6, N = 38	UCS)	
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DF	RILL		3				N	ATERIAL		1				ADDITIONAL	DATA	
SCALE (m) Method	Hole Support	Run	Water	KL (M)	Depth (m)	Graphic Log	Other Minor Componen	Description Name (USC Symbol) ts, Plasticity or Particle Condition, Consistency		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests		s/Observations test results	
12				- 52.0 - 51.0	12.0		CLAY with sand, pale yellow brittle, w < PL (Alluvial)	brown, high plasticity, fine	grained sand,	— сн-	м 	VSt- H H	SPT	12.0 - 12.45 m: SPT = 1 N = 51	1,23,28	
13				50.0									SPT	13.5-13.95 m: SPT = 8,	25,30*/110mm	
15 Sonic				<u>1</u> 8.0	15.0								SPT	15.0-15.45 m: SPT = 16	3,27,30°/110mm	
16				47.0	16 <u>.0</u> 								SPT	16.5-16.95 m: SPT = 15 N = 58	5,27,31	
18				<u>4</u> 6.0	18.0								SPT	18.0-18.45 m: SPT = 18	3,17,30*/95mm	
20				14.0	- 19 <u>.0</u> - - - - - 20.0	· · · · · · ·	END OF HOLE @ 20.0 m, r	souired denth					SPT	19.5-19.95 m: SPT = 22	2, 27, 30*115mm	
21					- - - - 21 <u>.0</u> - - - - -		END OF HOLE @ 20.0 m, r	admuaa aebtu								
22			E		- 22.0			DL - 1 4 1							Na	
See si detail:	s o	fab		viat	tion	s	GHD 2 Salam T: 61 3	P ty Ltd anca Square Hob 62100 600 F: 6 S PEOPLE PEF	art TAS 7001, H 61 3 62100601	lobart				Job	^{No.} 12538404	

Cli Pro	ojec			Geote	chn	-	Operations vestigation /						LOCAT	TION NO. I SHEET	BH03
		on :				-	93178.2 N GDA94	Surface RL : 263.4m		Incli	natio	n\Bea	ring : -90 \ 00	0 Processo	ed : CP
Co	ntra	acto	r:	Numa	ac			Rig Type : Sonic						Checked	I : TAS
Dat	te S	Start	ed	:26 No	ov 20	0		Date Completed : 27 Nov	20			Logge	ed by : SG	Date : 22	2 Mar 21
	DR	ILLI	NG				MA	ATERIAL					ADDITION	NAL DATA	
SCALE (m)	Method	Hole Support	Kun	RL (m)	Depth (m)	Graphic Log	Soil Na Other Minor Components	Description ame (USC Symbol) , Plasticity or Particle Characteristics, ondition, Consistency, Structure	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	_	nents/Observatio	ns
				E	111		TOPSOIL: CLAY with sand, ro grained, w < PL	otlets, orange brown, medium plasticity, fin	e Cl	D	F				
				<u>26</u> 3.0 	. .		CLAY with sand, pale orange b grained, brittle, w < PL (Alluvial	prown mottled yellow, medium plasticity, fine		┝┍	-н-				
1				Ē	- 1 <u>.0</u>		0	,							
				262.0	111		1.0 m: colour chage to pale be	ige yellow							
				Ē								SPT	1.5-1.95 m: SPT = N = 35	= 12,17,18	
2				E	2 <u>.0</u>							U75	U75-01 2.0 - 2.4 I	m	
				<u>-</u> <u>2</u> 61.0	111								R: 180/200 mm		
				Ē											
3				 -	3 <u>.0</u>		Sandy CLAY with Gravel, low p	plasticity, fine to coarse grained sand, fine		м	н.	SPT	3.0-3.45 m: SPT =	= 28, 30*/110mm	
				- <u>26</u> 0.0			to medium grained angular to s ~ PL (Alluvial)	sub angular gravel, no grading, polymictic, v	v						
				F											
4				Ē	4 <u>.0</u>										
				<u>25</u> 9.0								SPT	4.5-4.95 m: SPT =	= 6.12.19	
				Ē	50								N = 31	.,,	
5												U75	U75-02 4.95 - 5.4 R: refusal	h m	
	Sonic			<u>25</u> 8.0 	- -		Sandy CLAY, trace gravel, ora	ange brown mottled grey, medium plasticity,		- <u>-</u>	н.				
6				Ē	- 6 <u>.0</u>		sand is fine to coarse grained, (Alluvial)	, gravel is fine grained, brittle, w < PL				PP SPT	5.8 m: PP = 500 k 5.8-6.25 m: SPT =	:Pa (UCS) = 25, 30, 30*/130mm	
				- - 											
				F											
7				Ē	7 <u>.0</u>		Sandy CLAY with Gravel, low p	plasticity, fine to coarse grained sand, fine		м	н.				
				- 2 <u>5</u> 6.0			to medium grained angular to s ~ PL (Alluvial)	sub angular gravel, no grading, polymictic, v	v			PP	7.25 m: PP = 200		
				F								SPT	7.5-7.95 m: SPT =	= 22,29,30*/130mm	
8				Ē	8 <u>.0</u> -		Sandy CLAY, trace gravel, pal	e brown, medium to high plasticity, sand is is fine grained, brittle, w < PL (Alluvial)		Тм	н.	PP	8.0 m: PP > 600 k	Pa (UCS)	
				<u>25</u> 5.0	1.1										
				È.	9.0										
9												PP SPT	9.0 m: PP > 600 k 9.0-9.45 m: SPT = N = 35	Pa (UCS) = 7,15,20	
				<u>25</u> 4.0											
10				Ē	- 10 <u>.0</u>										
10				- - 2 <u>5</u> 3.0	11										
				Ē								SPT	10.5-10.95 m: SP	T = 14,23,30*/130mm	
11				<u>F</u>	11.0			th / 1 th d						ob No.	
				shee previa			GHD 2 Salama	ty Ltd nca Square Hobart TAS 7001, 2100 600 F: 61 3 6210060	Hobart				J		^ 4
				scrip				PEOPLE PERFORMANCE						125384	·U4

Client : Project : Location :		-	chnic	cal Inv	Operations vestigation						LOCATIO	N NO. BHO SHEET 2 OF	
Position :					3178.2 N GDA94	Surface RL : 263.4m		Incli	natio	n\Bea	ring : -90 \ 000	Processed : C	P
Contracto	r :	Numa	с			Rig Type : Sonic						Checked : TAS	S
Date Start	ed :	26 Nov	v 20			Date Completed : 27 Nov	20			Logge	ed by : SG	Date : 22 Mar	21
DRILLI	١G				MA	ATERIAL	1				ADDITIONAL	DATA	
SCALE (m) Method Hole Support Bun	Water	RL (m)	Depth (m)	Graphic Log	Soil Na Other Minor Components	Description ame (USC Symbol) 6, Plasticity or Particle Characteristics, ondition, Consistency, Structure	Group Symbol		Consistency / Relative Density		Insitu	s/Observations test results	
		- - 2 <u>5</u> 2.0	ŧ	<u> </u>				М	Н	PP	11.0 m: PP > 600 kPa (UCS)	
		Ē								PP	11.5 m: PP > 600 kPa (UCS)	
12		- -	- 12 <u>.0</u> -							SPT	11.8-12.25 m: SPT = 12 N = 50	2,20,30	
		_ 		•									
		Ę	+										
13		E.	13 <u>.0</u>										
		- <u>25</u> 0.0											
		Ē								PP SPT	13.5-13.95 m: SPT = 11 N = 49	1,22,27	
14		F.	14 <u>.0</u>		CLAY with sand, trace gravel,	red brown, high plasticity, fine to coarse vel, w ~ PL (Alluvial)		м	н.				
		_ <u>24</u> 9.0			grained sand, fine grained gra	vei, w ~ PL (Alluvial)							
		È											
15		F ·	15 <u>.0</u> -							SPT	15.0 m: PP > 600 kPa (15.0-15.45 m: SPT = 13	UCS)	
Sonic		<u>24</u> 8.0											
Ň		Ę								PP	15.5 m: PP > 600 kPa (UCS)	
16		Ē	16 <u>.0</u>							PP	16.0 m: PP > 600 kPa (UCS)	
		<u>24</u> 7.0	1							SPT	16.5-16.95 m: SPT = 13	3.20.28	
			17.0								N = 48	,,	
17		Ē								PP	17.0 m: PP > 600 kPa (UCS)	
		<u>24</u> 6.0 											
18		Ę.	- 18 <u>.0</u> -]									
		 								PP SPT	18.0 m: PP > 600 kPa (18.0-18.45 m: SPT = 12 N = 50	UCS) 2,24,26	
			ŧ										
19		<u> </u>	19 <u>.0</u>										
		- <u>24</u> 4.0											
		Ę		_						SPT	19.5-19.95 m: SPT = 13	3,19,30*/130mm	
20	_	<u>-</u>	20.0		END OF HOLE @ 20 m, requ	ired depth							
		- <u>24</u> 3.0	-										
		Ē											
21		Ē	21 <u>.0</u>										
		- <u>24</u> 2.0 -	-										
		Ē	-										
₂₂ See standa	ard	-	22.0 S fo	r	GHD P	'ty Ltd anca Square Hobart TAS 7001					Job	No.	
details of a					GHD 2 Salama	anča Square Hobart TAS 7001 62100 600 F: 61 3 6210060	, Hobart					12538404	

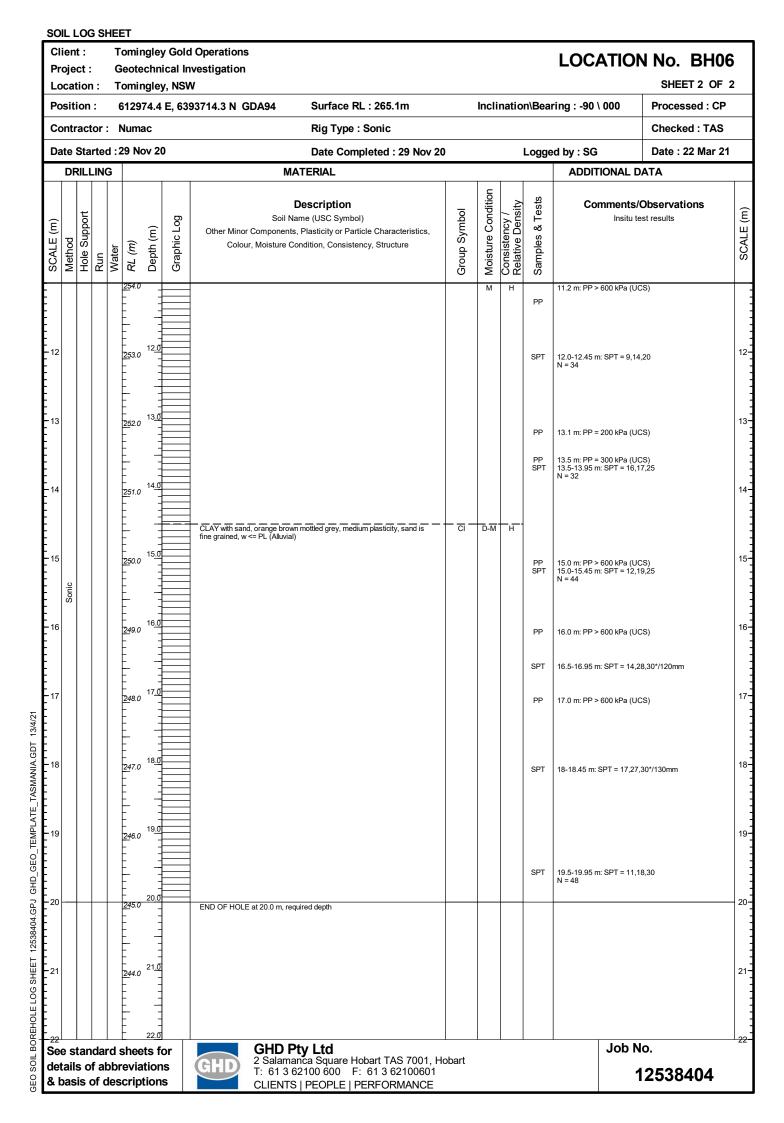
Clie Proj Loca	ject	t :	C	Geote	chn	-	l Operations vestigation V						LOCATIO	ON NO. BH SHEET 1 O	
Posi					-		93360.1 N GDA94	Surface RL : 263.8m		Incli	natio	n\Bea	ring : -90 \ 000	Processed : 0	CP
Con	tra	ctor	:	Numa	ac			Rig Type : Sonic						Checked : TA	S
Date	e St	tarte	ed :	27 No	ov 2	0		Date Completed : 27 No	ov 20			Logge	ed by : SG	Date : 22 Mar	[.] 21
	DRII	LLIN	IG				M	ATERIAL	1				ADDITIONA	L DATA	
SCALE (m) Method		Run Run	Water	KL (m)	Depth (m)	Graphic Log	Soil N Other Minor Component Colour, Moisture C	Description lame (USC Symbol) s, Plasticity or Particle Characteristic Condition, Consistency, Structure	Group S	Moisture Condition	Consistency / Relative Density	Samples & Tests		ts/Observations tu test results	
1				263.0	1 <u>.0</u> - - -		(Alluvial)	n plasticity, organics, grass, rootlets, w < , beige brown, high plasticity, fine to coar b angular to angular gravel, polymictic, nc	50 CH-	D	F	SPT	1.5-1.95 m: SPT = 14 N = 42	,21,21	
·2				<u>26</u> 2.0 - - - - - - - - - - - - - - - - - - -	2 <u>.0</u> - - -		grading, w < PL (Alluvial)	city, pale orange mottling, brittle, w<= PL		D-M	н	PP	N - 42 2.6 m: PP > 600 kPa	(UCS)	
·3 ·4				- - - - - - - - - - - - - - - - - - -	-							SPT PP PP	3.0-3.45 m: SPT = 15 N = 37 3.5 m: PP > 600 kPa 4.0 m: PP > 600 kPa	(UCS)	
5	2			 	- - - 5 <u>.0</u> -		CLAY trace sand, occasional grained sand, fine to coarse grading, polymictic, black and	l gravel, orange brown, medium to coarse grained sub angular to angular gravel, no i beige mottling, w ~ PL (Alluvial)	<u> </u>		н	PP SPT	4.5 m: PP > 600 kPa 4.5-4.95 m: SPT = 7, N = 30		
6 Sinco:				258.0 - - - - - - - - - - - - - - - - - - -	6 <u>.0</u> 							SPT	6.0-6.45 m: SPT = 10 N = 35	.21,14	
8			H 27/11/2020	- - - - - - - - - - - - - - - - - - -	7 <u>.0</u> - - - - 8 <u>.0</u> -							SPT PP	7.5-7.95 m: SPT = 25 8.0 m: PP = 300 kPa		
9				- - - - - - - - - - - - - - - - - - -								PP SPT U75 PP	9.0 m: PP = 500 kPa 9.0-9.45 m: SPT = 7,5 N = 25 U75-01 9.5 - 9.7 m R: 200/200 mm		
10				 	10 <u>.0</u> - - - -							SPT	9.7 m: PP = 400 kPa 10.5-10.95 m: SPT = N = 38		
deta	ils	of a	ıbb	shee revia scrip	tior	าร	T: 61 3 0	Pty Ltd anca Square Hobart TAS 700 62100 600 F: 61 3 621006 S PEOPLE PERFORMANG	501				Jot	No. 12538404	

OIL LC Client : Project	:	эп	Tor	ningle	-	Operations vestigation						LOCATIO	N No. E	3H04
Locatio					y, NSV									2 OF 2
Positio					E, 639	93360.1 N GDA94	Surface RL : 263.8m		Incli	natio	n\Bea	ring : -90 \ 000	Processe	
Contra					•		Rig Type : Sonic						Checked	
Date St DRII			_	NOV 2	0		Date Completed : 27 Not IATERIAL	v 20			Logge	ed by : SG	Date : 22	Mar 21
Method Lolo Support				Depth (m)	Graphic Log	Soil Other Minor Componen	Description Name (USC Symbol) ts, Plasticity or Particle Characteristics Condition, Consistency, Structure	Group Symbol	: Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/		ns
12				2.0 12 <u>.0</u> 11 <u>.0</u> 11.0 11.0 11.0					М	н	SPT	12.0-12.45 m: SPT = 8,15 N = 43 13.5-13.95 m: SPT = 11,		1
14 15 9ico 16			24	14 <u>.0</u> 							SPT	15.0-15.45 m: SPT = 12,2 N = 31	21,30	
17			- - - - - - - - - - - - - - - - - - -	17 <u>.0</u> 							SPT			
20				20.0 3.0 21 <u>.0</u> 		END OF HOLE at 20.0 m, r	equired depth							
Bee sta letails & basis	of	ab	brev	viatior	ns	1.013	Pty Ltd lanca Square Hobart TAS 7001 62100 600 F: 61 3 6210060 S PEOPLE PERFORMANC	1	<u> </u>	<u> </u>		Job N	^{io.} 125384	04

Clie Proj Loca	ect	:	(Geote	chr	-	d Operations avestigation V						LOCATIO	ON NO. BHO SHEET 1 OF	
Posi					-	-	93663.9 N GDA94	Surface RL : 264.6m		Incli	natio	n\Bea	ring : -90 \ 000	Processed : C	P
Con	tra	ctor	:	Numa	ас			Rig Type : Sonic						Checked : TAS	s
Date	e St	arte	d :	28 No	ov 2	0		Date Completed : 28 Nov	v 20			Logge	ed by : SG	Date : 22 Mar	21
D	DRIL	LIN	G				М	IATERIAL					ADDITIONA	L DATA	
SCALE (m) Method	Metriou Hola Support	Run	Water	RL (m)	Depth (m)	Graphic Log	Soil N Other Minor Component	Description Name (USC Symbol) ts, Plasticity or Particle Characteristics Condition, Consistency, Structure	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests		ts/Observations tu test results	
1				<u>26</u> 4.0	- - - - - - - - - - - - - - - - - - -		grained sub angular to angula	ange brown, medium plasticity, fine to mediu ar, no grading, polymictic, w < PL (Alluvial)		D	St	SPT	1.5-1.95 m: SPT = 10 N = 23	,8,15	
3				<u>26</u> 2.0 <u>26</u> 2.0 <u>2</u> <u>2</u> <u>2</u> <u>2</u> <u>2</u> <u>2</u>	2 <u>.0</u> 		mottled, high plasticity, fine to w ~ PL (Alluvial)	I, pale yellow beige brown, orange brown o coarse grained sand, gravel is fine graine	d,			SPT	3.0-3.45 m: SPT = 9, N = 26	11,15	
т 5 . <u>ч</u>				260.0	4 <u>.0</u> - - - - - - - - - - - - - - - - - - -							PP SPT U75	4.0 m: PP = 500 kPa 4.5-4.95 m: SPT = 11 N = 22 U75-01 5.0 - 5.4 m R: 400/400	,13,9	
5 6 7				259.0 - - - - - - - - - - - - - - - - - - -	6 <u>.0</u> - - - 7 <u>.0</u>		medium plasticity, fine to med	range brown, grey and beige mottling, dium grained sand, fine to coarse grained s grading, polymictic, w ~ PL (Alluvial)		м	VSt	PP PP SPT U75 PP	5.4 m: PP = 500 kPa 5.6 m: PP = 300 kPa 6.0 m: PP = 300 kPa 6.0-6.45 m: SPT = 5,8 N = 16 U75-02 6.5 - 6.9 m R: 400/400 6.9 m: PP = 300 kPa	(UCS) (UCS) 3,8	
3				257.0 - - - - - - - - - - - - - - - - - - -	8 <u>.0</u>							PP SPT PP	7.5 m: PP = 250 kPa 7.5-7.95 m: SPT = 6, N = 20 8.0 m: PP = 300 kPa	10,10	
9				 	9 <u>.0</u> - - - - - - - - - - - - - - - - - - -		9.0 m: mottling becomes mor yellow	re prevalent, orange brown, pale grey and			VSt	PP SPT PP	9.0 m: PP = 300 kPa 9.0-9.45 m: SPT = 4, N = 30 10.0 m: PP > 600 kPa	12,18	
11				_ <u>25</u> 4.0 _ _			10.5 m: colour change to pak				н	PP SPT	10.5 m: PP > 600 kPa 10.5-10.95 m: SPT = N = 42	10,19,23	
leta	ils	of a	bb	shee revia scrip	tio	ns	GHD GHD F 2 Salam T: 61 3 CHENT	Pty Ltd anca Square Hobart TAS 7001 62100 600 F: 61 3 6210060 S PEOPLE PERFORMANC	∣, Hobart)1 ⊏				Jot	No. 12538404	

Clier Proje Loca			-	hnical	Id Operations Investigation SW						LOCATIO	N NO. BHO SHEET 2 OF	
	tion :				393663.9 N GDA94	Surface RL : 264.6m		Incli	natio	n\Bea	ring : -90 \ 000	Processed : Cl	Р
Cont	racto	or:	Numa	•		Rig Type : Sonic						Checked : TAS	3
Date	Star	ted	28 Nov	/ 20		Date Completed : 28 Nov	20			Logge	ed by : SG	Date : 22 Mar 2	21
D	RILLI	ING				MATERIAL					ADDITIONAL	DATA	
SCALE (m) Method	Hole Support	Run	RL (m)	Depth (m) Graphic Log	Soil Other Minor Compone Colour, Moisture	Description Name (USC Symbol) nts, Plasticity or Particle Characteristics, Condition, Consistency, Structure	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests		s/Observations test results	
12			 					М	VSt	SPT	12.0-12.45 m: SPT = 9, N = 45	19,26	
13			 	13 <u>.0</u> 						SPT	13.5-13.95 m: SPT = 14 N = 44	1,19,25	
15 Sonic			- 2 <u>5</u> 0.0 - - - - - - - - - - - - - - - - - -		15.0 m: colour change to or	range brown with pale grey mottling			н	SPT	15.0-15.45 m: SPT = 8, N = 45	19,26	
16			 							SPT	16.5-16.95 m: SPT = 18	,25,30*/135mm	
18			2 <u>4</u> 7.0							SPT	18.0-18.45 m: SPT = 15 N = 46	5,21,25	
20			- - - - - - - - - -							SPT	19.5-19.95 m: SPT = 8, N = 45	19,26	
21				- - - 21 <u>.0</u> - -	END OF HOLE at 20.0 m,	required depth							
			<u>24</u> 3.0	22.0									
			sheet	s for	GHD 2 Salar T: 61 3	Pty Ltd nanca Square Hobart TAS 7001, 3 62100 600 F: 61 3 6210060	Hobart				Job		
			scripti		T: 61 3	8 62100 600 F: 61 3 6210060 FS PEOPLE PERFORMANCE	1					12538404	

Clie Proj Loca	jec	t :		Geo	tec	hni		Operations vestigation /							LOC	ATION	NO. BHO	
Posi					-			3714.3 N GDA94	Surface RL : 265	.1m		Incli	natio	n\Bea	ring : -90	000	Processed : CF	P
Con	tra	cto	r:	Nur	nac	;			Rig Type : Sonic								Checked : TAS	;
Date	e S	tart	ed	:29	Nov	/ 20)		Date Completed	: 29 Nov 20				Logge	ed by : SG		Date : 22 Mar 2	21
D	DRI	LLI	NG					M	ATERIAL						-		TA	
SCALE (M) Method		Hole Support	Run	Water RL (m)		Depth (m)	Graphic Log	Soil Na Other Minor Components	Description ame (USC Symbol) s, Plasticity or Particle Char ondition, Consistency, Stru		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Co	mments/O Insitu tes	bservations t results	
				<u>26</u> 5 - - - - - -	.0				ange brown, medium plasticity, angular gravels, no grading, p		CI	D	St					
1				264 - - - - -				CLAY with sand, trace gravel, fine to medium grained sand, angular gravels, no grading, p	beige brown with orange, med fine to coarse grained angular olymictic, w < PL (Alluvial)	ium plasticity, to sub	<u> </u>	D	VSt	SPT	1.5-1.95 m: 5 N = 23	SPT = 9,11,12		
2				- 263 - - - - - -	.0	2.0	· · · · · · · · · · · · · · · · · · ·							U75	U75-01 2.0 - R: 300/300 n	2.3 m nm		
3				- 262 - - - - - - - -	.0	3.0		3.0 m: slight colour change to	pale orange beige					SPT	3.0-3.45 m: 5 N = 30	SPT = 10,14,16		
5 5	onic			261 	.0	4.0		Sandy CLAY, beige brown, hig ~ PL (Alluvial)	jh plasticity, fine to medium gra	ined sand, w	— сн —	M	VSt	SPT	4.5-4.95 m: S N = 22	SPT = 7,10,12		
6	ō			- - - - - - - -	.0	6 <u>.0</u>	· · · · ·							PP SPT	6.0-6.45 m: 5 N = 16			
7				 	.0	7.0	· · · · · · · · · · · · · · · · · · ·							U75 PP	U75-02 6.5 - R: 400/400 n 7.0 m: PP = 5	6.9 m nm 500 kPa (UCS)		
8				 	.0	8.0	0 0 0 0	Sandy Gravelly CLAY, orange sand, fine to medium grained a polymictic, w ~ PL (Alluvial)	brown, low plasticity, fine to cc angular to sub angular gravel, r	barse grained to grading,		м	н	SPT PP	N = 42	SPT = 18,20,22		
9				 	.0	9.0	<i>•</i>	CLAY with sand, trace gravel, is fine to coarse grained, grav	beige brown, medium to high p el is fine grained, w ~ PL (Alluv	asticity, sand ial)	СІ-СН	M	н	PP SPT PP	9.0-9.45 m: 8 N = 50	600 kPa (UCS) SPT = 20,23,27 600 kPa (UCS)		
10				- 1255 - - - - - - -	.0 1	0.0 - - - - -								PP SPT	10.7-11.15 m	600 kPa (UCS n: SPT = 9,13,1		
₁₁∟ See	sta	hne	larr	t she		1.0 s fc		GHD P	Ptv Ltd						N = 30	Job No).	
deta	ils	of	abl	brev escri	iati	ion	s	GHD 2 Salama T: 61 3 6 CLIENTS	P ty Ltd anca Square Hobart Ta 52100 600 F: 61 3 6 PEOPLE PERFOR	AS 7001, Ho 62100601	bart						2538404	



Clio Pro Loc	ojec	ct :	:	Ge	otec	hni		Operations vestigation /							LOCA	TION	No. BHO	
Pos	siti	on		61	332	0.9	E, 639	3650.8 N GDA94	Surface RL : 265.4n	n		Incli	natio	n\Bea	ring : -90 \ 0	00	Processed : C	P
Coi	ntra	acto	or:	Nu	uma	;			Rig Type : Sonic								Checked : TAS	5
Dat	te S	Star	ted	: 30	Nov	/ 20)		Date Completed : 2	Dec 20			I	Logge	d by : IG		Date : 22 Mar	21
	DR	ILL	ING	i				M	ATERIAL						ADDITIO	NAL DA	ТА	
SCALE (m)	Method	Hole Support	Run	Water	KL (m)	Depth (m)	Graphic Log	Soil N Other Minor Components	Description lame (USC Symbol) s, Plasticity or Particle Characte condition, Consistency, Structure	eristics,	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comr	ments/O Insitu test	bservations results	
1					<u>5</u> 5.0 - <u>-</u>	1.0		sand is fine grained, w < PL (.				D	F 					
2				2 1 1 1 1 1 2 1 1 1 1 1	- 53.0	2.0			eige brown, low plasticity, coarse gr angular to sub angular gravel, no ob Jluvial)					SPT	1.5-1.95 m: SPT N = 46			
5					<u>5</u> 2.0		· · · · · · · · · · · · · · · · · · ·	CLAY with sand, pale beige b grained sand, grey and orang	rown, medium plasticity, fine to mec le brown mottling, w ~ PL (Alluvial)	lium	CI	M	VSt	PP SPT U75	PP @ 3.0 m 200 3.0-3.45 m: SPT N = 20 U75-01 3.5 - 3.9 R: 400/400			
4				1 1 12	- <u>5</u> 1.0	4.0	· · · · ·							PP PP SPT	PP @ 3.9 m 425 PP @ 4.0 m 200 4.5 - 4.95 m: SP N = 24) kPa		
5	Sonic			1 1 1 2	- <u>5</u> 0.0	5.0	· · · · · · · · · · · · · · · · · · ·							PP	5.5 m: PP = 300	kPa (UCS)		
6					- 59.0	6 <u>.0</u> 	· · · · · · · · · · · · · · · · · · ·							SPT	6.0-6.45 m: SPT N = 19	. ,		
7					- 58.0		· · · · · · · · · · · · · · · · · · ·		ale brown, low plasticity, medium to o grained angular to sub angular grav			M	VSt					
8					-	8.0	· · · · · · · · · · · · · · · · · · ·	7.5 m: becoming mottled with	pale grey					PP SPT PP	7.5 m: PP = 150 7.5-7.95 m: SPT N = 31 8.0 m: PP = 350	ī = 11,14,17		
9					57.0 - 56.0	9.0								SPT	9.0-9.45 m: SPT N = 25	T = 8,12,13		
10					- ,	- - - - - - - - - - - - - - - - - - -	· · · · · · · · · · · · · · · · · · ·											
11					<u>5</u> 5.0	- - - - 11.0	· · · · · · · · · · · · · · · · · · ·							SPT	10.5-10.95 m: S N = 37			
					neet: viati			GHD F 2 Salama T: 61 3 0	Pty Ltd anca Square Hobart TAS	7001, Hoba	art				•	Job No		
					ripti			1. 0100	62100 600 F: 61 3 621 S PEOPLE PERFORM							11	2538404	

Clien Proje Loca	ect :			ech	nica	l Inv	Operations vestigation							LOCATIO	ON NO. BH SHEET 2 (
Posit							3650.8 N GDA94	Surface RL :	265.4m		Incli	natio	n\Bea	ring : -90 \ 000	Processed :	СР
Cont	ract	or :	Num	ac				Rig Type : Se	onic						Checked : T	AS
Date	Sta	rted	:30 N	lov 2	20			Date Comple	eted : 2 Dec 20)			Logge	ed by : IG	Date : 22 Ma	r 21
D	RILL	ING	i		-		М	ATERIAL		1	1			ADDITIONAL	DATA	
SCALE (m) Method	Hole Support	Run	Water RL (m)	Depth (m)	Crashie Loca	GIAPIIIC LOU	Soil N Other Minor Component	Description lame (USC Symbol) is, Plasticity or Particle Condition, Consistency		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests		ts/Observations u test results	
12			254.0 - - - - - - 253.0 - - -	- 12 <u>.</u>)			CLAY with sand, trace grave sand is fine to coarse graine 12.0 m: some grey mottling	I, orange brown, medium i d, gravel is fine grained, v	to high plasticity, v < PL (Alluvial)	СІ-СН	D	VSt- H	PP SPT	12.0m: PP > 600 kPa 12.0-12.45 m: SPT = 1 N = 32	(UCS) 5,15,17	
13 14			252.0 - - - - - - - - - - - - - - - - - - -	- 14 <u>.</u>		· · · · · · · · · · · · · · · · · · ·							SPT	13.5-13.95 m: SPT = 1 N = 43	3,13,30	
15 Sonic			- - - - - - - - - - - - - - - - - - -	-		· · · · · · · · · · · · · · · · · · ·							PP SPT	15.0 m: PP > 600 kPa 15.0-15.45 m: SPT = 1	1,21,30*/120mm	
16			- 249.0 - - - - - - - - - - - - - - - - - - -	- 17 <u>.</u>									PP SPT	16.5 m: PP = 400 kPa 16.5-16.95 m: SPT = 1 N = 35	(UCS) 6,15,20	
18			- - - - - - - - - - - - - - - - - - -	18 <u>.</u>) -		· · · · · · · · · · · · · · · · · · ·							PP SPT	18.0 m: PP > 600 kPa 18.0-18.45 m: SPT = 1 N = 47	(UCS) 7,20,27	
19			 	19 <u>.</u>)									PP SPT	19m: PP > 600 kPa (U 19.5-19.95 m: SPT = 2		
20			- - - - - - - - - - - - - - - - - - -	20.			END OF HOLE at 20.0 m, at	required depth								
21			 	21 <u>.</u>) 22.												
			d she	ets	for			P ty Ltd anca Square Hoba 62100 600 F: 6						Job	No.	
detail			brevia escrip		-		GHD 2 Salam T: 61 3	anca Square Hoba 62100 600 F: 6	ait 1AS 7001, F 51 3 62100601	TIBUUR					12538404	

Cli Pro	-	ct :	:	Ge	ote	chn		Operations vestigation /							LOCATIO	ON NO. BHO SHEET 1 O	
	siti							93608.7 N GDA94	Surface RL : 265.	9m		Incli	natio	n\Bea	ring : -90 \ 000	Processed : C	P
Co	ntra	acto	or :	Nu	uma	C			Rig Type : Sonic							Checked : TA	S
Dat	te S	Star	ted	:2	Dec	: 20			Date Completed :	3 Dec 20				Logge	ed by : IG	Date : 22 Mar	21
	DR	ILL	ING	;				м	ATERIAL						ADDITIONAL	DATA	
SCALE (m)	Method	Hole Support	Run	Water	KL (m)	Depth (m)	Graphic Log	Soil N Other Minor Component	Description lame (USC Symbol) s, Plasticity or Particle Chara Condition, Consistency, Struct		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests		ts/Observations u test results	
				Ē		-		Sandy CLAY, with gravel, ora sand, angular to sub angular	nge brown, low plasticity, fine to gravel, some small roots, w < PL	coarse _ (Alluvial)	CL	D	F				
				F	-			0.5 m: becoming pale orange									
1					<u>5</u> 5.0	1.0		1.0 m: becoming grey					н	SPT	1.5-1.95 m: SPT = 16,	22,21	
2					<u>6</u> 4.0	2.0									N = 43		
·3					<u>5</u> 3.0 - <u>5</u> 2.0	3.0 								SPT	3.0-3.45 m: SPT = 18,	30*/85mm	
5	Sonic				- <u>5</u> 1.0	5.0		CLAY trace sand, orange bro sand is fine grained, w ~ PL (own some grey mottling, medium Alluvial)	plasticity, — -	<u>- cı</u> -	M	VSt	PP SPT U75	4.5 m: PP = 350 kPa (4.5-4.95 m: SPT = 6,1 N = 24 U75-01 5.0 - 5.4 m R: 400/400		
6	Ň			2	<u>5</u> 0.0 -	6 <u>.0</u> 1								PP SPT	6.0 m: PP > 600 kPa (6.0-6.45 m: SPT = 12, N = 35	18,17	
7				N 3/11/2020	<u>5</u> 9.0 -	- 7 <u>.0</u> - -		Gravelly CLAY, orange brown gravel, w ~ PL (Alluvial)	n, medium plasticity, angular to si	uo angular	CI	М	H	PP	6.5 m: PP = 450 kPa (UCS)	
			[F	58.0		· · ·	Sandy CLAY, trace gravel, pa grained sand, gravel is fine g	ale brown, medium plasticity, fine rained w > PL (Alluvial)	to medium	CI	W	Н	PP SPT	7.5 m: PP = 500 kPa (7.5-7.95 m: SPT = 6,1 N = 32	UCS) 2,20	
8					-	8.0		Sandy CLAY with gravel, brov gravel, some cobbles (larger	wn, medium plasticity, angular to gravels), w ~ PL (Alluvial)	sub angular	— с і —	M	VSt- H	PP	8.0 m: PP = 200 kPa (UCS)	
9					57.0 - 56.0	9.0		CLAY with sand, orange brov grained, w ~ PL (Alluvial)	vn, low to medium plasticity, sand	l is fine — — -	<u>- ci</u> -	M	H H	PP SPT	9.0 m: PP > 600 kPa (9.0-9.45 m: SPT = 8,1 N = 33	UCS) 3,20	
10					- 55.0	10 <u>.0</u> - - - - 11.0								SPT	10.5-10.95 m: SPT = 1	10,21,30*/115mm	
11 See	e st	and	dar	d sł			or	GHD F	P ty Ltd anca Square Hobart TA 62100 600 F: 61 3 6	0.700/					Job	No.	
deta	ails	s of		bre	viat	tion	s	GHD 2 Salama T: 61 3	anca Square Hobart TA	S 7001, Ho	bart					12538404	

Client Projec Locati	ct:	C	-	nical Inv	Operations vestigation						LOCATIC	N NO. BHO SHEET 2 OF	
Positio	on :			-	3608.7 N GDA94	Surface RL : 265.9m		Incli	natio	n\Bea	ring : -90 \ 000	Processed : Cl	P
Contra	actor	:	Numac			Rig Type : Sonic						Checked : TAS	3
Date S	Starte	d :	2 Dec 20			Date Completed : 3 Dec	20			Logge	ed by : IG	Date : 22 Mar 2	21
DR	ILLIN	G			MA	TERIAL					ADDITIONAL	DATA	
SCALE (m) Method	Hole Support Run	Water	<i>RL (m)</i> Depth (m)	Graphic Log	Soil Na Other Minor Components,	Description me (USC Symbol) Plasticity or Particle Characteristics ndition, Consistency, Structure	Group Symbol	Moisture Condition	 T Consistency / Relative Density 	Samples & Tests		5/Observations test results	
12			254.0 		11.5 m: dark orange brown col	bur				PP SPT	12.0 m: PP > 600 kPa (12.0-12.45 m: SPT = 27	UCS) 7,301/95mm	
13			253.0 - 13.0 252.0 - 14.0 		12.8 m: orange brown colour					SPT	13.5-13.95 m: SPT = 16	5,24,30*/125mm	
15 Sonic			251.0 - 15 <u>.0</u> 							PP SPT	15.0 m: PP > 600 kPa (15.0-15.45 m: SPT = 14	UCS) ,25,30*/125mm	
16			<u>25</u> 0.0 - 16 <u>.0</u> - 16 <u>.0</u> - 16 <u>.0</u>							SPT	16.5-16.95 m: SPT = 11 N = 37	1,12,25	
18			248.0 18 <u>.0</u> 							SPT	18.0-18.45 m: SPT = 17 N = 57	27,30	
19			247.0 - - 19.0 							PP SPT	19.0 m: PP > 600 kPa (19.0-19.45 m: SPT = 20	UCS) ,30°/10mm	
20			246.0 20.0 20.0 245.0 245.0 245.0		END OF HOLE at 20.0 m, requ	uired depth							
			244.0 2240 22.0 Sheets f	for	GHD P 2 Salamar	ty Ltd nca Square Hobart TAS 7001 2100 600 F: 61 3 6210060	, Hobart				Job	No.	
			reviatior scription		GHD 2 Salamar T: 61 3 6	nca Square Hobart TAS 7001 2100 600 F: 61 3 6210060 PEOPLE PERFORMANC	, nobart)1					12538404	

Clie Proj Loc	ject	t :	C	Geote	echr	-	Operations vestigation V							LOCAT	TON NO. SHEE	BH09
Pos					-	-	93577.4 N GDA94	Surface RL : 26	6.8m		Incli	natio	n\Bea	ring : -90 \ 00	0 Proces	sed : CP
Con	ntra	ctor	• :	Num	ac			Rig Type : Soni	с						Check	ed : TAS
Date	e Si	tarte	ed :	3 De	c 20			Date Completed	d : 4 Dec 20				Logge	ed by : IG	Date : 2	22 Mar 21
0	DRI	LLIN	IG				M	ATERIAL						ADDITION	IAL DATA	
SCALE (m)	Method	Run Run	Water	KL (m)	Depth (m)	Graphic Log	Soil N Other Minor Components	Description ame (USC Symbol) s, Plasticity or Particle Cha condition, Consistency, Str		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests		ents/Observat	ions
				E	-		CLAY with sand, dark orange grained, w < PL	brown, medium plasticity, sar	nd is fine	CI	D	Fb				
1				- - - - - - - -			Gravelly CLAY, dark brown, n gravel, w <= PL (Alluvial) Sandy CLAY trace gravel, pal coarse grained sand, angular		-	<u>- ci</u> -	D-M D-M	— _Н —-	PP	0.5 m: PP > 600 k	Pa (UCS)	
2				- - - 265.0	2 <u>.0</u>	0							SPT	1.5-1.95 m: SPT = N = 37	: 15,17,20	
-				- - - - - - - - - - - - - - - - - - -									PP	2.4 m: PP > 600 k	Pa (UCS)	
3				 <u>26</u> 3.0		0 0 0							SPT	3.0-3.45 m: SPT = N = 56	: 18,27,28	
4				 <u>26</u> 2.0	4 <u>.0</u> - - - -		4.5 m: grey colour, increasing			— <u></u> –	M		PP SPT	4.4 m: PP = 500 k 4.5-4.95 m: SPT =	(Pa (UCS) : 18,30*/15mm	
5	Sonic			 	5 <u>.0</u> - - - -		grained sand, angular to sub a	angular gravel, w ~ PL (Alluvia	al)	0.			PP	5.2 m: PP > 600 k	Pa (UCS)	
6				<u>-</u> - - - - - - - - - - - - - - - - - -	6 <u>.0</u> - - - -		6.0 m: orange brown colour						PP SPT	6.0 m: PP > 600 k 6.0-6.45 m: SPT = N= 47	Pa (UCS) : 12,19,28	
8				- - - - - - - - - - - - - - - - - - -			Sandy CLAY, orange brown, i sand, w ~ PL (Alluvial)	medium plasticity, fine to med	ium grained	— <u></u> –	M		PP SPT	7.5 m: PP > 600 k 7.5-7.95 m: SPT = N = 31	Pa (UCS) 6,14,17	
9				 	9 <u>.0</u>		CLAY, orange brown, medium	n plasticity (Alluvial)		— <u>-</u> -	м		PP SPT	9.0 m: PP > 600 k 9.0-9.45 m: SPT = N = 52	Pa (UCS) - 22,22,30	
10				- - - 2 <u>5</u> 7.0 - - - -												
				- - - 	_								SPT	10.5-10.95 m: SP N = 40	T = 9,17,23	
11⊥ See	sta		 ard	⊦ shee	<u>11.0</u> ets f		GHD F	Ptv Ltd			L			 	ob No.	
deta	ails	of a	abb	revia scrip	atior	าร	T: 6136	P ty Ltd anca Square Hobart ⁻ 62100 600 F: 61 3 S PEOPLE PERFO	62100601	bart					12538	404

Clie Proj Loca	ect	::	G	Geote	echn	-	Operations vestigation						LOCATIC	N NO. BHO	
Posi							- 93577.4 N GDA94	Surface RL : 266.8m		Inc	linatio	on\Bea	aring : -90 \ 000	Processed : C	
Con	trad	ctor	:	Num	ac			Rig Type : Sonic						Checked : TAS	5
Date	e St	arte	d :	3 De	c 20			Date Completed : 4 D)ec 20			Logg	ed by : IG	Date : 22 Mar	21
D	RIL	LIN	G				M	ATERIAL					ADDITIONAL	DATA	
SCALE (m) Method	Hole Sunnort	Run	Water	RL (m)	Depth (m)	Graphic Log	Soil N Other Minor Components	Description Jame (USC Symbol) Is, Plasticity or Particle Characteris Condition, Consistency, Structure	stics, Stics, Solution	Moisture Condition		Samples & Tests		s/Observations test results	
12				- - - - - - - - - - - - - - - - - - -	12 <u>.0</u> - - -		12.0 m: some grey mottling					PP SPT	12.0 m: PP > 600 kPa (12.0-12.45 m: SPT = 9	UCS) 25,30*/20mm	
13				- - - - - - - - - - - - - - - - - - -	13 <u>.0</u> 							PP SPT	13.5 m: PP > 600 kPa 13.5-13.95 m: SPT = 1 N = 41	UCS) 2,14,27	
15 	20110			 252.0 251.0	15 <u>.0</u> - - - - -		Gravelly Sandy CLAY, dark o to coarse grained, gravel is fi ~ PL (Alluvial)	prange brown, medium plasticity, sand i ine grained, angular to sub angular gra	is fine CI avel, w	M		SPT	15.0-15.45 m: SPT = 1 N = 44	9,19,25	
16				- - - - - - - - - - - - - - - - - - -	16 <u>.0</u> - - - - - - - - - - - - - - - - - - -		Sandy CLAY, dark orange bro grained sand, rare gravels, w	own, medium plasticity, medium to coa /~ PL (Alluvial)	irse CI	M		SPT	16.5-16.95 m: SPT = 1 N = 45	0,20,25	
18				 249.0 248.0	18 <u>.0</u> - - -							SPT	18.0-18.45 m: SPT = 1 N = 53	5,27,26	
19				240.0 	19 <u>.0</u> - - -							SPT	19.5-19.95 m: SPT = 1	5,30*/10mm	
21				 	- - - 21 <u>.0</u> - - - - - -		END OF HOLE at 20.0 m, re	quired depth							
-22				<u>24</u> 5.0	22.0										
See : detai & ba	ils	of a	bb	revia	ets f atior	or 1s		Pty Ltd anca Square Hobart TAS 7/ 62100 600 F: 61 3 6210 S PEOPLE PERFORMA		_			Job	No. 12538404	

Pro	•	t: ct: tion		G	eote	chni		Operations vestigation V						LOCA	ATION	NO. BH1 SHEET 1 OF	
Ро	siti	ion	:	6	1373	89.8	E, 639	93286.2 N GDA94	Surface RL : 265.5m		Incli	natio	n\Bea	ring : -90 \	000	Processed : CP	כ
Co	ontr	act	or :	N	luma	C			Rig Type : Sonic							Checked : TAS	;
Da	te :	Sta	rtec	1:4	Dec	: 20			Date Completed : 4 Dec	20			Logge	ed by : IG		Date : 22 Mar 2	21
	DF	RILL	INC	6				MA	TERIAL					ADDIT	IONAL DA	ATA	
SCALE (m)	Method	Hole Support	Run	Water	KT (m)	Depth (m)	Graphic Log	Soil Na Other Minor Components,	Description Ime (USC Symbol) , Plasticity or Particle Characteristics ondition, Consistency, Structure	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Cor	nments/C	Observations tresults	
					-	-	°0	Gravelly Sandy CLAY, orange coarse grained, rare angular to	brown, medium plasticity, sand is fine to o sub angular gravels, w < PL	CI	D-M	F					
1				-	<u>26</u> 5.0 - - - - -	- - 1 <u>.0</u>	 	CLAY with sand, dark orange b PL (Alluvial)	rown, medium plasticity, rare gravels, w	<u>-</u>	D-M	F.					
				-	- - 2 <u>6</u> 4.0 - - -	2.0	· · · · · · · · · · · · · · · · · · ·	Sandy CLAY, trace gravel, gre grained sand, fine grained grav	y, medium plasticity, medium to coarse vel, w < PL (Alluvial)		<u>р-</u> м	н.	SPT	1.5-1.95 m: S N = 40	PT = 15,20,20)	
3				-	 	2.0											
4				-	 262.0 	4.0		Sandy CLAY with gravel, orang coarse grained sand, angular t	je brown, medium plasticity, medium to o sub angular gravels, w ∼ PL (Alluvial)		м	VSt	PP SPT	3.0 m: PP > 6 3.0-3.45 m: S N = 24	00 kPa (UCS) PT = 8,9,15		
5				-	- - - 2 <u>6</u> 1.0 - - - - -	5.0							SPT	4.5-4.95 m: S N = 26	PT = 11,10,16	5	
6	Sonic				- - 2 <u>6</u> 0.0 - - -	6.0								5.5 m: PP = 3			
0				-	- - 2 <u>5</u> 9.0 - - -		•0 -0 -0 	Sandy CLAY, orange brown, m PL (Alluvial)	edium plasticity, medium grained sand, w	v~	м	н	PP SPT	6.0 m: PP > 6 6.0-6.45 m: S N = 37	00 kPa (UCS) PT = 12,15,22		
7 8						7.0		7.0 m: some grey mottling					PP SPT	7.5 m: PP > 6 7.5-7.95 m: S N = 31	00 kPa (UCS) PT = 9,15,16		
9				-	- - 2 <u>5</u> 7.0 - - - - - - 2 <u>5</u> 6.0	9.0							PP SPT	9.0 m: PP > 6 9.0-9.45 m: S N = 41	00 kPa (UCS PT = 10,22,19	5	
10				-	<u>-</u> - - - - - 2 <u>5</u> 5.0			CLAY, orange brown, medium	plasticity, w~PL (Alluvial)				SPT	10.5-10.95 m N = 35	: SPT = 6,12,2	13	
11					-	- 11.0											
det	ails	s of	f ab	bro	sheet eviat cript	tion	IS		ty Ltd nca Square Hobart TAS 7001 2100 600 F: 61 3 6210060 PEOPLE PERFORMANC						Job No	o. 2538404	

Clier Proje Loca	ect		G	ieote	chn	-	l Operations vestigation V						LOCATIO	N NO. BH1 SHEET 2 OF	
Posit	tion	1:					93286.2 N GDA94	Surface RL : 265.5m		Incli	natio	n\Bea	ring : -90 \ 000	Processed : C	Ρ
Cont	rac	tor	: 1	Numa	ac			Rig Type : Sonic						Checked : TAS	3
Date	Sta	arte	d :4	1 Dec	: 20			Date Completed : 4 Dec	: 20			Logge	ed by : IG	Date : 22 Mar 2	21
D	RIL		G				M	ATERIAL		1			ADDITIONAL I	DATA	
SCALE (m) Method	Hole Support	Run	Water	RL (m)	Depth (m)	Graphic Log	Soil N Other Minor Components	Description lame (USC Symbol) s, Plasticity or Particle Characteristic: condition, Consistency, Structure	^{s,} Group Symbol	Moisture Condition	[±] Consistency / Relative Density	Samples & Tests		/Observations rest results	
12				254.0 - - - - - - - - - - - - - - - - - - -		· · ·		own, medium plasticity, medium to coarse)		D	н.	SPT	12.0-12,45 m: SPT = 9,1 N = 44	8.26	
13				252.0	13 <u>.0</u> 14 <u>.0</u> 		CLAY, trace sand and gravel, medium to coarse grained sa	pale orange brown, medium plasticity, nd and fine grained gravels, w < PL (Alluv		D	н	SPT	13.5-13.95 m: SPT = 19,	30*/30mm	
15 Sonic				251.0 - - - - - - - - - - - - - - - - - - -								SPT	15.0-15.45 m: SPT = 14,	30*/20mm	
16				- - - - - - - - - - - - - - - - - - -	16 <u>.0</u> 17 <u>.0</u> 							SPT	16.5-16.95 m: SPT = 14,	28,30*/40mm	
18				248.0 - - - - - 247.0	 18 <u>.0</u> 							SPT	18.0-18.45 m: SPT = 20,	30*/70mm	
19				 	- 19 <u>.0</u> - - - - - - - - - - - - - - - - - - -							SPT	19.5-19.95 m: SPT = 16,	30°/10mm	
20				- - - - - - - - - - - - - - - - - - -	20.0 		END OF HOLE at 20.0 m, rea	quired depth				<u></u>			
²² See s			rd :	shee		or	GHD F	Pty Ltd anca Square Hobart TAS 700 62100 600 F: 61 3 621006	1 Hobert				Job I	No.	
detail				evia crip			CLIENTS	62100 600 F: 61 3 621006	01					12538404	

Pr	ient oje ocat		Geot	ingley Gold Operations technical Investigation ingley, NSW						LOCAT	ION NO. TPO SHEET 1 OF	
		on :		678.8 E, 6393170.0 N GDA94	Surface RL	: 265	.5m			Pit Width : 1.5	Processed : TA	S
Сс	ontr	actor	: TGC)	Machine :	Doosa	n DX	225L	.C	Pit Length : 3.5	Checked : TAS	
Da	te :		17 N	Nov 20						Logged by : SG	Date : 22 Mar 2	1
EX	CA			MATERIAL	-					ADDITIONAL I	DATA	
Scale (m)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure	ency,	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ob		
		0.60		TOPSOIL: Sandy Clay with rootlet: plasticity, orange brown, w < PL	s, low	CL	D	F	DCP	DCP (blows/100mm): 7,11,15,15	,15,18,32	
		0.60 264.91		Sandy CLAY, low plasticity, orange < PL (Alluvial)	e brown, w	CL	D	Н	BS	BS (0.5 - 1.0m)		
1									PP	1.0m: PP > 600 kPa (UCS)		
									BS	BS (1.0 - 2.0m)		
2		2.00 263.51		Sandy CLAY with Gravel, low plas brown, gravel is fine to coarse grai subround to subangular, polymictio (Alluvial)	ined,	CL	D	н	PP	2.0m: PP > 600 kPa (UCS)		
	GNE								BS	nodules of white weathered rock of brittle, 20cm diam BS (2.0 - 3.0m)	(feldspar, plagioclase),	
3		3 20										
		<u>3.20</u> 262.31		Effective refusal with 0.6m wide, to bucket	oothed							
4												
5												
				ets for ations GHD Pt 2 Salamar T: 61.3 62	nča Square Ho	bart TA	AS 70	01, ⊢	lobart	Jol	b No.	
				T: 61 3 62	2100.600 F: PEOPLE PI	6136	52100	601			12538404	

Clieı Proje	nt :	Tom Geo	TION LOG SHEET ingley Gold Operations technical Investigation ingley, NSW						LOCATIO	ON No. TP02 SHEET 1 OF	
Posi	tion :		703.7 E, 6393315.0 N GDA94	Surface RL	: 265.	7m			Pit Width : 1.3	Processed : TAS	3
Cont	tracto	r: TG	0	Machine : D	Doosa	n DX	225L	.C	Pit Length : 3.5	Checked : TAS	
Date	:	17 I	Nov 20						Logged by : SG	Date : 22 Mar 21	
XCA			MATERIAL	-					ADDITIONAL D	ATA	
Scale (m) Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Obs		
			TOPSOIL: Sandy Clay with rootlet plasticity, orange brown, w < PL	s, low	CL	D	H	BS PP	DCP (blows/100mm): 3,3,14,14,9,8,9,9,9,12,14,16,14,27 BS (0.0 - 0.5m) 0.5m: PP > 600 kPa	7	
1	1.00 264.67		CLAY with sand, trace gravel, low brown, gravel is fine to coarse grai subround to subangular, w < PL (A	ined,	CL	D	Η	PP BS	1.1m: PP > 600 kPa 1.5m: white inclusions possible we	athered rock? feldspar	-
2 ENC									plagioclase, brittle, <20 cm diam BS (1.0 - 2.0m)		
3	3.00 262.67 3.50 262.17	· · · ·	Sandy CLAY, medium plasticity, pa with occasional orange mottles, w (Alluvial)	ale grey <= PL	CI	D- M	Н	PP BS	3.0m: PP > 600 kPa BS (3.0 - 3.5m)		
4	202.11		Limit of hole								
etai	ls of a	abbrev	T: 61 3 6	ty Ltd nca Square Hob 2100 600 F: PEOPLE PE	6136	62100	601	lobart	Job	No. 12538404	

Clie Proj Loca	ect	t :	Geot	ingley Gold Operations echnical Investigation ingley, NSW						LOCATI	ON NO. TPO SHEET 1 OF	
Posi				505.9 E, 6393348.0 N GDA94	Surface RL	: 265	.1m			Pit Width : 1.3	Processed : TA	٩S
Con	tra	ctor :	TGC)	Machine : D	Doosa	n DX	225L	.C	Pit Length : 2.1	Checked : TAS	5
Date	ə :		17 N	lov 20						Logged by : SG	Date : 22 Mar 2	21
XCA	4			MATERIAL						ADDITIONAL [DATA	
Scale (m) Water		Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure	ency,	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ob Insitu test r		
		0.50		TOPSOIL: Sandy Clay with rootlet: plasticity, orange brown, w < PL	s, Iow	CL	D	Т	DCP PP BS	DCP (blows/100mm): 5,7,10,12,14,16,15,12,11,11,11,1 BS (0.0 - 0.5m)	13,10,10,10	
	26	0.50		CLAY, medium plasticity, orange b PL, reduced rootlets (Alluvial)	orown, w <	CI	D	Η	PP	0.5m: PP > 600 kPa (UCS)		
1									PP	1.0m: PP > 600 kPa (UCS)		
									BS	BS (1.0 - 2.0m)		
2		2.00 63.07	· · · · · · · · · · · · · · · · · · ·	CLAY with sand, medium plasticity brown-yellow, w < PL (Alluvial)	/, pale	CI	D	Н				
									BS	BS (2.0 - 3.0m)		
	26	<u>3.50</u> 61.57										
				ets for GHD P	hy I tel					 		
eta	ils	of at	obrevi	ations GHD 2 Salamar T: 61 3 62	t y ∟to nca Square Hot 2100 600 F: PEOPLE PE	6136	62100	601	lobart		12538404	

Clie Proj Loca	nt : ect	:	Tom Geot	FION LOG SHEET ingley Gold Operations technical Investigation ingley, NSW						LOCAT	ON NO. TP	
Posi				534.7 E, 6393509.0 N GDA94	Surface R	L : 265	.8m			Pit Width : 1.5	Processed : T	AS
Con	tra	ctor :	TGC)	Machine :	Doosa	n DX	225L	.c	Pit Length : 2	Checked : TA	s
Date):		17 N	Nov 20						Logged by : SG	Date : 22 Mar	21
XCA	1			MATERIAL	•					ADDITIONAL I	DATA	
Water		Deptn / (KL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity of Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ob Insitu test r		
				TOPSOIL: Sandy Clay with rootlets plasticity, orange brown, w < PL	s, low	CL	D	Н	DCP	DCP (blows/100mm): 3,3,4,15,26	3	
									PP	0.2m: PP > 600 kPa (UCS)		
	26	0.50		CLAY with sand, trace gravel, high orange brown, gravel is fine to med grained (Alluvial)	n plasticity, dium	СН	D	Н	BS PP	0.5m: PP > 600 kPa (UCS) BS (0.0-1.0m)		
1		-	· · · · · · · · · · · · · · · · · · ·						PP	1.0m: PP > 600 kPa (UCS)		
			· · · · · · · · · · · · · · · · · · ·						BS	BS (1.0 - 2.0m)		
2	26	1.80 64.04	· · ·	with gravel, fine grained, subround subangular, black mottled orange b (Alluvial)	l to prown	СН	D	н				:
	26	<u>2.10</u> 63.74		Effective refusal with 0.6m wide, to bucket	oothed							
GNF	5											
3												
Ļ												
			!	ets for GHD Pt	by I to							
etai	ils	of ab	obrevi	ations GHD 2 Salamar T: 61 3 62	L Y LLO nca Square Ho 2100 600 F: PEOPLE P	6136	62100	601	lobart		12538404	

CI Pr	ien oje		Tom Geot	FION LOG SHEET ingley Gold Operations technical Investigation ingley, NSW						LOCAT	ION NO. TP(SHEET 1 OI	
		ion :		554.0 E, 6393622.0 N GDA94	Surface R	L : 266	.0m			Pit Width : 1.3	Processed : T	
Сс	ontr	actor			Machine :	Doosa	n DX	225L	.c	Pit Length : 3.7	Checked : TAS	S
	te			Nov 20						Logged by : SG	Date : 22 Mar	21
EX	CA			MATERIA	L					ADDITIONAL I	DATA	
Scale (m)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consis Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ot		SCALE (m)
			\boxtimes	TOPSOIL: Sandy Clay with rootle plasticity, brown, w < PL	ts, low	CL	D	Н	DCP	DCP (blows/100mm): 4,17,18,25		
		0.50							BS PP	BS (0.0 - 0.5m) 0.3m: PP > 600 kPa (UCS)		
		0.50 265.45		Sandy CLAY with gravel, high pla orange mottled brown, gravel is fi subround to subangular, w < PL (ne grained,	CH	D	Η	PP	0.5m: PP > 600 kPa (UCS)		
1									PP	1.0m: PP > 600 kPa (UCS)		1
			· · · ·						PP	1.3m: PP > 600 kPa		
		2.00	· · · · · · · · · · · · · · · · · · ·						PP	1.9m: PP > 600 kPa		
-2-	GNE	<u>2.00</u> 263.95		Effective refusal								2
-3												3
4												4
<u>-</u>												
det	tails	s of a	bbrevi	T: 6136	Pty Ltd anca Square H 62100 600 F 6 PEOPLE F	: 6130	62100	601	lobart	Jo	^{b No.} 12538404	

Clier Proje _oca		Geo	ingley Gold Operations technical Investigation ingley, NSW						LOCA	TION NO. T SHEET 1	
	tion :		299.9 E, 6393550.0 N GDA94	Surface R	L : 265	.2m			Pit Width : 1.5	Processed	: TAS
Cont	racto	r: TG	D	Machine :	Doosa	n DX	225L	.C	Pit Length : 2	Checked :	TAS
Date	:	17	Nov 20						Logged by : SG	Date : 22 N	lar 21
XCA			MATERIAL	_					ADDITIONA		
Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests		Observations st results	
2	0.6(CLAY with sand, trace gravel, med plasticity, brown mottled black and gravel is fine to coarse grained, su subangular (Alluvial)	l orange,	CI	D	Н	DCP SV PP PP	DCP (blows/100mm): 7,7,9,9,2 0.1m: SV = 145/30 kPa 0.3m: PP > 600 kPa (UCS) 0.5m: PP > 600 kPa (UCS) 1.0m: PP > 600 kPa (UCS)		
GNE								PP PP	2.5m: PP > 600 kPa (UCS) 3.0m: PP > 600 kPa (UCS)		
ŀ	<u>3.7(</u> 261.54		Limit of hole					PP	3.7m: PP > 600 kPa (UCS)		
etai	ls of a	ard she abbrevi descri		ty Ltd nca Square Ho 2100 600 F: PEOPLE P	6136	52100	0601	lobart	U	ob No. 1253840	4

Clie Pro	ent	::	Tomi	FION LOG SHEET ingley Gold Operations rechnical Investigation						LOCAT	ION No. TP0	7
	-	ion :		ingley, NSW							SHEET 1 OF	1
Position : Contractor : Date :		613	318.8 E, 6393661.0 N GDA94	Surface R	Pit Width : 1.4 Processed :		TAS					
		: TGO Machine : Doosan DX 225LC					Pit Length : 3.1	Checked : TAS	TAS			
		18 Nov 20							Logged by : SG	Date : 22 Mar 2	Mar 21	
XC	A			MATERIAL	-					ADDITIONAL	DATA	
	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure	ency,	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ot Insitu test		SCALE (m)
				TOPSOIL: Sandy Clay with rootlet plasticity, orange brown, w < PL	s, low	CL	D	St	DCP SV PP	DCP (blows/100mm): 5,3,8,20,20,21,24,21,20,17,15,1 0.1m: SV = 70/8 kPa 0.2m: PP > 600 kPa	8,13,18,18,34,24,18,1	7,19, ⁻
									PP BS	0.5m: PP > 600 kPa BS (0.5 - 1.0m)		
1									PP SV	1.0m: PP > 600 kPa 1.0m: SV unable to penetrate		1
	2	1.50 264.11		CLAY with sand, trace gravel, med plasticity orange brown occasional mottles, gravel is fine to coarse gra subround to subangular (Alluvial)	black	CI	D	Н	BS	BS (1.0 - 2.0m)		
<u>-</u>	GNE	<u>2.00</u> 263.61	· ·	Effective refusal with 0.6m wide, to bucket	oothed							_ 2
3												3
												4
-												5
leta	ails	s of al	bbrevi	1: 61 3 6	ty Ltd nca Square Ho 2100 600 F: PEOPLE P	: 6130	62100	0601	lobart	JO	^{b No.} 12538404	

Pr	ien oje ocat		Geot	ingley Gold Operations technical Investigation ingley, NSW						LOCATI	ON NO. TPO SHEET 1 OF	
		on :		085.4 E, 6393690.0 N GDA94	Surface RL	: 265.	4m			Pit Width : 1.4	Processed : TA	S
Сс	ontr	actor	: TGC)	Machine : D	oosa	n DX	225L	.C	Pit Length : 2.4	Checked : TAS	
Da	ate	1	18 N	Nov 20						Logged by : SG	Date : 22 Mar 2 ⁴	1
EX	CA			MATERIAL						ADDITIONAL D	ATA	
Scale (m)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity of Particle Characteristics, Colour, Moisture Condition, Consister Structure	ncy,	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Obs		
				TOPSOIL: Sandy Clay with rootlets plasticity, orange brown, w < PL	s, low	CL	D	St	DCP SV PP PP BS	DCP (blows/100mm): 9,12,9,9,11 0.1m: SV = 68/24 kPa 0.2m: PP > 300 kPa 0.5m: PP > 600 kPa BS (0.0 - 1.0m)	13,16,16,15,17,32	
1		0.70 264.72		CLAY with sand, trace gravel, medi plasticity, brown occasionally black orange mottled, gravel is coarse gra subround to sub angular (Alluvial)	and	CI	D	Н	PP	1.0m: PP > 600 kPa		
2		2.00 263.42		CLAY with sand and gravel, mediur fine to coarse grained sand, pale br (Alluvial)	m plasticity, rown	CI	D	D	PP	2.0m: PP > 600 kPa		
3	GNE								BS	BS (2.0 - 3.0m)		
		<u>3.10</u> 262.32		Limit of hole								
4												
_												
т Эе	e si	anda	rd she	ets for GHD Pt	y Ltd ica Square Hob					Job	No.	—
				ations GHD 2 Salaman T: 61 3 62	īca Square Hob	art TA	S 70	01, H	lobart		12538404	

INTEGRA & Existance RL: 284.3m PIL Width: 1.4 Processed: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS Dreacked: TAS <	Clien Proje Locat		Geot	ngley Gold Operations echnical Investigation ngley, NSW						LOCAT	ION NO. T SHEET 1	
Initiation: TOO Machine : Docean DX 25LC PH Length : 3 Checked : TAS tei: 18 Nov 20 Date : 32 Mar 21 ADDITIONAL DATA Checked : TAS ADDITIONAL DATA ADDITIONAL DATA ADDITIONAL DATA Image: Standard Company Bit S					Surface RI	. : 264	.9m			Pit Width : 1.4		
In: 18 Nov 20 Logged by: 3G Date: 22 Mar 21 CA MATERIAL ADDITIONAL DAT Display Display Display Comments/Observations mathematics Comments/Observations UP UP Open Mark Constraints, C	Contr	actor	: TGO)	Machine :	Doosa	n DX	225L	.c		Checked :	TAS
CA MATERIAL ADDITIONAL DATA gring and gring and analysis of the second comparison of the detail and a contract of the detail and the detail and a co	Date	:	18 N	lov 20							Date : 22 N	lar 21
Image: State of the set of	ХСА			MATERIAL							DATA	
brown, w < PL brown, w < PL Service Classification in a submitted in the submitted in t		Depth / (RL) metres	Graphic Log	Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste		Group Symbol	Moisture Condition	Consistency / Relative Density	~ð	Comments/O	oservations	
B C I ES ES (0.0 - 1.0m) is indemit to case grained, subround to subengular (Aluxia) I I IS ES (0.0 - 1.0m) I I I I IS ES (0.0 - 1.0m) I I I I IS ES (0.0 - 1.0m) I I I I I IS ES (0.0 - 1.0m) I I I I I IS ES (0.0 - 1.0m) I I I I I I IS IS (0.0 - 1.0m) I I I I I I IS IS (0.0 - 1.0m) I I I I IS IS IS (0.0 - 2.0m) I I I I I I I I I I I I I I I I I I I I I I I I I I I I <t< td=""><td></td><td>0.50</td><td></td><td>TOPSOIL: Sandy Clay, low plastic brown, w < PL</td><td>ity, orange</td><td>CL</td><td>D</td><td>St</td><td>SV PP</td><td>0.1m: SV = 65/15 kPa 0.2m: PP > 600 kPa</td><td></td><td></td></t<>		0.50		TOPSOIL: Sandy Clay, low plastic brown, w < PL	ity, orange	CL	D	St	SV PP	0.1m: SV = 65/15 kPa 0.2m: PP > 600 kPa		
e standard sheets for the de abharante Signal Population (1990) Hobart (1897) Hobart (1990) Hobart		264.44		brown with orange and black stain is medium to coarse grained, subr	ing, gravel	CI	D	Н				
e standard sheets for tip of a bheets for tip of a bheets for tip of a bheets for Salamanda Square Hobart TAS 7001, Hobart Job No.	1								PP	1.0m: PP > 600 kPa		
e standard sheets for Ville of otherwritetrars Salamanca Square Hobart TAS 7001, Hobart									BS	BS (1.0 - 2.0m)		
e standard sheets for Ville of otherwritetrars Salamanca Square Hobart TAS 7001, Hobart	2	2.40										
e standard sheets for tails of abbreviations EHD Pty Ltd 2 Salamanca Square Hobart TAS 7001, Hobart 2 Salamanca	GNE	262.54		Limit of hole								
e standard sheets for tails of abbreviations EXAMPLE AND ADDRESS TO 1, Hobart TAS 7001, Hobart 7001, Ho	3											
e standard sheets for tails of abbreviations GHD Pty Ltd 2 Salamanca Square Hobart TAS 7001, Hobart T: 61 3 62100 600 F: 61 3 62100601	ŀ											
e standard sheets for tails of abbreviations GHD Pty Ltd 5 CHD 2 Salamanca Square Hobart TAS 7001, Hobart 5 CH 3 C2100 600 F: 61 3 C2100601												
tails of abbreviations 2 Salamanca Square Hobart TAS 7001, Hobart	ee s	tanda	rd she	ets for GHD P	ty Ltd					Jo	b No.	
Dasis of descriptions	etail	s of a	bbrevia	T: 6136	nca Square Ho 2100 600 F·	bart TA 61 3 6	AS 70 62100	01, ⊢)601	lobart		1253840	Л

Clie Proj Loca	ect	t :	Geot	ingley Gold Operations technical Investigation ingley, NSW						LOCAT	ON NO. TP1 SHEET 1 OF	
Posi				870.5 E, 6393621.0 N GDA94	Surface RL	. : 264	.6m			Pit Width : 1.3	Processed : T/	AS
Con	tra	ctor :	TGC)	Machine :	Doosa	n DX	225L	.C	Pit Length : 3.2	Checked : TAS	3
Date	:		18 N	Nov 20						Logged by : SG	Date : 22 Mar 2	21
XCA	1			MATERIAL			1			ADDITIONAL I	DATA	
Vuater		Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity of Particle Characteristics, Colour, Moisture Condition, Consister Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ob Insitu test r		
				TOPSOIL: Sandy Clay with rootlets plasticity, orange brown, w < PL	s, low	CL	D	H	DCP SV PP	DCP (blows/100mm): 4,7,15,20,2 0.1m: SV = 95/10 kPa 0.2m: PP > 600 kPa	27	
	2	0.50		Sandy CLAY, trace gravel, low plas orange brown, w < PL (Alluvial)	sticity,	CL	D	Н	PP BS	0.5m: PP > 600 kPa BS (0.0 - 1.0m)		
		-							PP	1.0m: PP > 600 kPa		
	2	1.50 63.07		Sandy CLAY, trace gravel, medium yellow brown with black and orange trace Fe, gravel is medium to coars	e mottled,	CI	D	Н				
		-	· · ·	subround to subangular (Alluvial)					BS	BS (1.5 - 2.0)		
CNF CNF	GIL								PP	2.0m: PP > 600 kPa		
	2	<u>3.00</u> 61.57	· · · · · · · · · · · · · · · · · · ·	Effective refusal with 0.6m wide, to bucket	pothed							
			- I - I		w I te					_	b No.	_
eta	ils	of ab	obrevi	1: 61 3 62	t y Ltd hca Square Ho 2100 600 F: PEOPLE PE	6136	52100	601	lobart	30	12538404	

Pr	ient ojeo cat		Geot	ingley Gold Operations echnical Investigation ingley, NSW						LOCAT	ION NO. TP1 SHEET 1 OF	
Po	siti	on :		067.7 E, 6393588.0 N GDA94	Surface RL	: 264	.9m			Pit Width : 1.4	Processed : TA	AS
Co	ontr	actor	TGC)	Machine : D	oosa	n DX	225L	C	Pit Length : 4	Checked : TAS	;
Da	te :		18 N	lov 20						Logged by : SG	Date : 22 Mar 2	21
X	CA			MATERIAL	-					ADDITIONAL	DATA	
scale (m)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/O Insitu test		
				TOPSOIL: Sandy Clay with rootlet: plasticity, orange brown, w < PL	s, low	CL	D	H	PP	DCP (blows/100mm): 3,6,13,19,18,17,15,13,11,14,14, 0.2m: PP > 600 kPa	13,13,13,12	
		0.60 264.27	· · · · · · · · · · · · · · · · · · ·	Sandy CLAY, trace organics, high orange brown, sand is fine to coars w < PL (Alluvial)	plasticity, se grained,	СН	D	Н	PP BS	0.5m: PP > 600 kPa BS (0.0 - 1.0m)		
1			· · · · ·						PP	1.0m: PP > 600 kPa		
									BS	BS (1.0 - 2.0m)		
2	GNE	2.00 262.87		Sandy CLAY with gravel, high plas grey-brown with orange and black sand is fine to coarse grained, grav grained, sub-rounded, w < PL (All	staining, vel is fine	СН	D	VD				
3	0								BS	BS (2.0 - 3.2m)		
		<u>3.20</u> 261.67		Effective refusal with 0.6m wide, to bucket	oothed							
e	e st	andar	d she	ets for GHD P	ty Ltd					Jo	b No.	
			obrevi	ations GHD 2 Salamar T: 61 3 62 CLIENTS	ty Ltd nca Square Hob 2100 600 F:	oart TA 61 3 6	AS 70	01, ⊢ 0601	lobart		12538404	

Pr	-		Tomi Geot	ION LOG SHEET ngley Gold Operations echnical Investigation ingley, NSW						LOCAT	ION NO. TP1 SHEET 1 OF	
		on :		276.8 E, 6393385.0 N GDA94	Surface RL :	265.	0m			Pit Width : 1.3	Processed : TA	
Сс	ontr	actor	TGO)	Machine : Do	osar	ו DX	225L	c	Pit Length : 3.5	Checked : TAS	;
	te :			lov 20						Logged by : SG	Date : 22 Mar 2	21
EXC	CA			MATERIAL						ADDITIONAL	DATA	
(m)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ot		
				TOPSOIL: Sandy Clay, with rootle plasticity, orange brown, w < PL	ts, low	CL	D	VSt	DCP SV PP	DCP (blows/100mm): 4,7,19,16,17,15,14,11,12,13,14, 0.1m: SV = 78/20 kPa 0.2m: PP = 300 kPa	14,12,14,12	
		0.60 264.43		CLAY with sand, low to medium pl trace organics, orange brown, w < (Alluvial)	asticity, (PL	CL- CI	D	Н	PP BS	0.5m: PP > 600 kPa BS (0.0 - 1.0m)		
1									PP	1.0m: PP > 600 kPa		
									BS	BS (1.0 - 2.0m)		
2									PP	2.0m: PP > 600 kPa		
3	GNE	2.50 262.53		Sandy CLAY, low plasticity, orange brown, sand is fine to coarse grain (Alluvial)	e yellow ed, w < PL	CL	D	Н				
									BS	BS (3.0 - 4.0m)		
T		<u>4.00</u> 261.03		Limit of hole								
-												
				ets for GHD P	ty Ltd nca Square Hoba	art TA	S 70	 01 ⊔	lohart	Jo	b No.	_
h	ails	s of ab	obrevia	ations GHD 2 Salamar T: 61 3 6 CLIENTS	2100 600 F: 6	136	2100	51, I⊓ 601	Juan		12538404	

Pr	ient ojeo cat		Geot	ingley Gold Operations echnical Investigation ingley, NSW						LOCAT	ION NO. TP' SHEET 1 OI	
		on :		043.5 E, 6393417.0 N GDA94	Surface RL	: 264.	1m			Pit Width : 1.4	Processed : T	AS
Co	ontr	actor	: TGC)	Machine : D)oosa	n DX	225L	.C	Pit Length : 4	Checked : TA	s
Da	te :		19 N	lov 20						Logged by : SG	Date : 22 Mar	21
X	CA			MATERIAL	-					ADDITIONAL I	DATA	
ocale (III)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ob Insitu test		
				TOPSOIL: Sandy Clay, low plastic brown, w < PL	ity, orange	CL	D	VSt	DCP	DCP (blows/100mm): 4,4,8,9,29 0.1m: SV = 120/20 kPa		
									PP	0.3m: PP = 300 kPa (UCS)		
									PP	0.5m: PP = 300 kPa (UCS)		
		0.60 263.50		Sandy CLAY, trace organics, medi plasticity, yellow brown, sand is fin medium grained, w < PL (Alluvial)	e to	CI	D	Н	BS	BS (0.0 - 1.0m)		
			· · · · · · · · · · · · · · · · · · ·						PP	1.0m: PP > 600 kPa		
	GNE	2.00 262.10 <u>3.50</u> 260.60		Sandy CLAY with gravel, medium pale yellow brown, sand is fine to o grained, gravel is fine grained, well subangular, polymictic, w < PL (A	coarse I rounded to Iluvial)	CI	D	Н	BS	BS (2.0 - 3.0m)		
		200.00		Effective refusal of 0.6m wide, too	thed bucket							
e	e st	andai	rd she	ets for GHD Pt	t y Ltd nca Square Hob	. –	0 -	0.1		Jo	b No.	
~*	ails	s of al	bbrevi	ations GHD 2 Salamar T: 61 3 62 CLIENTS	nca Square Hob	part TA	AS 70	01, H	Iobart		12538404	

Cli Pro	ent ojec	::	Tomi Geot	ION LOG SHEET ngley Gold Operations echnical Investigation ingley, NSW						LOCAT	ION NO. TP	
		on :		252.2 E, 6393237.0 N GDA94	Surface RL	. : 264	.4m			Pit Width : 1.3	Processed :	
Co	ntr	actor	: TGC		Machine :	Doosa	n DX	2251	С	Pit Length : 2.8	Checked : T	AS
Da				lov 20		20000				Logged by : SG	Date : 22 Ma	
EXC			101	MATERIAL						ADDITIONAL		1 2 1
(m)		Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity of Particle Characteristics, Colour, Moisture Condition, Consiste Structure	ency,	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/O	bservations	
-1	GNE	0.60 263.80 260.40		TOPSOIL: Sandy Clay with rootlets plasticity, orange brown, w < PL Sandy CLAY, trace gravel, black m orange brown, medium plasticity, s to coarse grained, gravel is fine gra rounded to subangular, w < PL (Al	nottled and is fine ained, well	CI	D	H H	DCP SV PP BS BS BS	DCP (blows/100mm): 4,7,14,16,18,13,10,10,12,10,12, 0.1m: SV = 185/45 kPa 0.3m: PP > 600 kPa (UCS) BS (0.0 - 1.0m) clasts consist of quartz, feldspar, 1.0m: PP > 600 kPa (UCS) BS (1.0 - 2.0m) BS (1.0 - 2.0m) BS (2.0 - 2.5m) change to trenching bucket		
det	ails	s of a	rd she bbrevi	T: 61 3 62	t y Ltd nca Square Ho 2100 600 F: PEOPLE PI	6136	62100)601	lobart	ol	b No. 12538404	

	ient oje			ngley Gold Operations echnical Investigation						LOCAT	ION No. TP1	
		ion : on :		ngley, NSW 182.5 E, 6393207.0 N GDA94	Surface RL	· 261	6m				SHEET 1 OF Processed : TA	
			: TGC					0051	•	Pit Width : 0.8	Checked : TAS	
	ntr ite :			, lov 20	Machine : [Joosa		225L		Pit Length : 3.7 Logged by : SG	Date : 22 Mar 2	
			1910	MATERIAL						ADDITIONAL		
	~				-			~	ts	ADDITIONAL		Т
scale (m)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure	ency,	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ol Insitu test	results	
				TOPSOIL: Sandy Clay with rootlet plasticity, orange brown, w < PL	s, low	CL	D	Н	DCP SV	DCP (blows/100mm): 12,16,26,2 0.1m: SV = 148/16 kPa	6,26	
									PP	0.3m: PP > 600 kPa		
		0.50 264.15		Sandy CLAY, trace gravel, low to r plasticity, orange brown, sand is fir grained, w < PL (Alluvial)	medium ne to coarse	CL - CI	D	Н	PP BS	0.5m: PP > 600 kPa BS (0.0 - 1.0m)		
1									PP			
									BS	BS (1.0 - 2.0m)		
	GNE											
		<u>2.80</u> 261.85		Effective refusal with 0.6m wide, to	oothed							
;				bucket								
	e st	anda	rd she	ets for GHD P	ty Ltd nca Square Hol					Jo	b No.	=
				ations GHD 2 Salamar T: 61 3 6	nca Square Hol	oart TA	AS 70	01, ⊢	lobart			

Clier Proje Loca	ect :	Geo	ningley Gold Operations stechnical Investigation ningley, NSW						LOCAT	ON NO. TP1 SHEET 1 OF	
Posi			3730.6 E, 6393476.0 N GDA94	Surface RI	. : 266	.2m			Pit Width : 0.8	Processed : TA	s
Con	tract	tor: TG	0	Machine :	Doosa	n DX	225L	.c	Pit Length : 3.3	Checked : TAS	;
Date):	19	Nov 20						Logged by : SG	Date : 22 Mar 2	21
EXCA			MATERIAL	•					ADDITIONAL I		
Scale (m) Water	Depth / (RL)	metres Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ob Insitu test r		
-1 -1 -2 -3 -4	0.265	.60	Structure TOPSOIL: clay with sand and root sand is fine to medium grained, w Sandy CLAY with gravel, medium brown, sand is fine to coarse grain fine grained, subrounded to suban PL (Alluvial) Effective refusal with 0.6m wide, to bucket	< PL plasticity, led, gravel is gular, w <	CI			DCP PP BS PP BS BS	DCP (blows/100mm): 6,8,13,13,17,17,18,13,14,15,16,7 0.3m: PP > 600 kPa 0.5m: PP > 600 kPa BS (0.0 - 1.0m) 1.0m: PP > 600 kPa 1.0m: white inclusions of possible plagioclase, < 20cm diam BS (1.0 - 2.0m) BS (2.0 - 3.0m)		
letai	ls o	idard she f abbrev of descri	1: 61 3 62	ty Ltd nca Square Hc 2100 600 F: PEOPLE P	6136	52100	1001	lobart	Jol	No. 12538404	

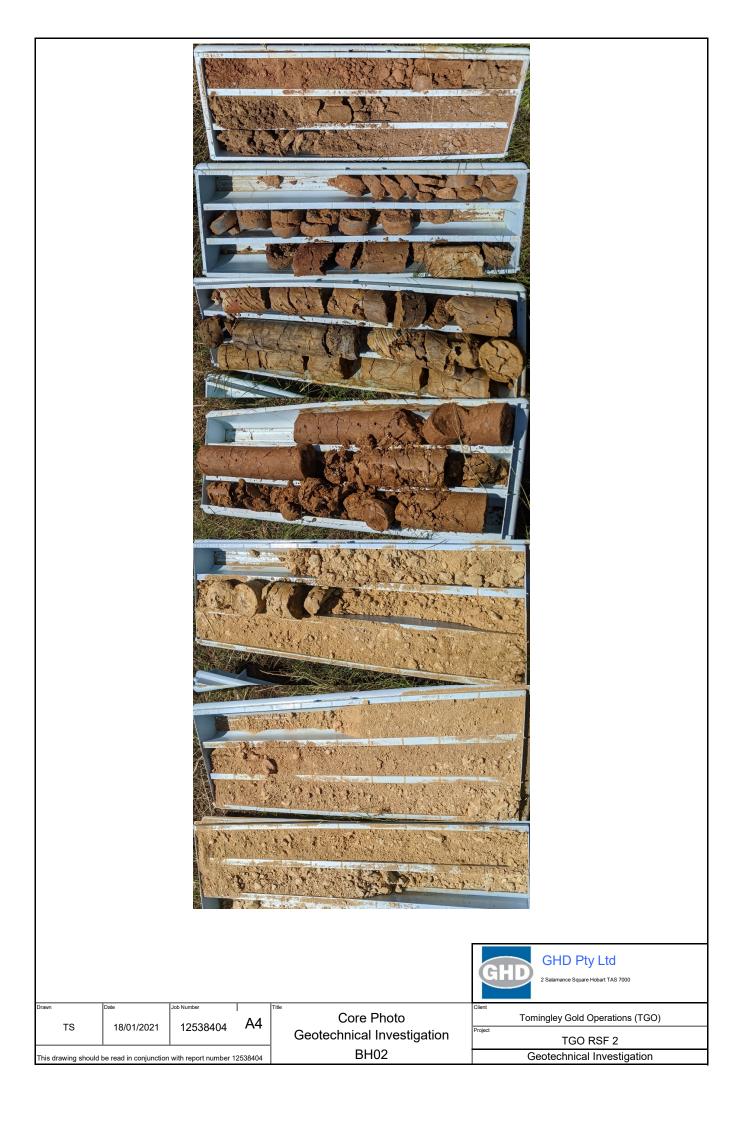
Pr	ient oje ocat		Geot	ingley Gold Operations echnical Investigation ingley, NSW						LOCAT	ION NO. TP1 SHEET 1 OF	
		on :		745.1 E, 6393589.0 N GDA94	Surface R	L : 266	.7m			Pit Width : 0.8	Processed : TA	AS
Сс	ontr	actor	: TGC)	Machine :	Doosa	n DX	225L	.c	Pit Length : 3.2	Checked : TAS	3
Da	te :	1	19 N	lov 20						Logged by : SG	Date : 22 Mar 2	21
EX	CA			MATERIAL	-					ADDITIONAL	DATA	
Scale (m)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/O Insitu test		
				TOPSOIL: Sandy Clay with rootlet: plasticity, orange brown, w < PL	s, Iow	CL	D	Н	DCP	DCP (blows/100mm): 9,6,10,12,16,15,17,13,13,12,9,9 0.3m: PP > 600 kPa	9,20,18,19	
		0.60 266.13		Sandy CLAY, trace gravel, mediun orange brown, sand is fine to coars w < PL (Alluvial)	n plasticity, se grained,	CI	D	Н	PP BS	0.5m: PP > 600 kPa BS (0.0 - 1.0m)		
·1		1.00 265.73		Sandy CLAY, trace gravel, mediun black mottled yellow brown, sand is medium grained, gravel is fine grai PL (Alluvial)	s fine to	CI	D	н		1.0m: PP > 600 kPa		1
									BS	BS (1.0 - 2.0m)		
2	GNE								BS	BS (2.0 - 3.0m)		
3		3 30										
		<u>3.30</u> 263.43		Effective refusal with 0.6m wide, to bucket	oothed							
4												
5												
		s of al	bbrevi	ets for ations otions GHD GHD 2 Salamar T: 61 3 62 CLIENTS	ty Ltd nca Square Ho 2100 600 F:	obart T/ 61 3 6	AS 70 52100	01, ⊢)601	lobart	Jo	b No. 12538404	

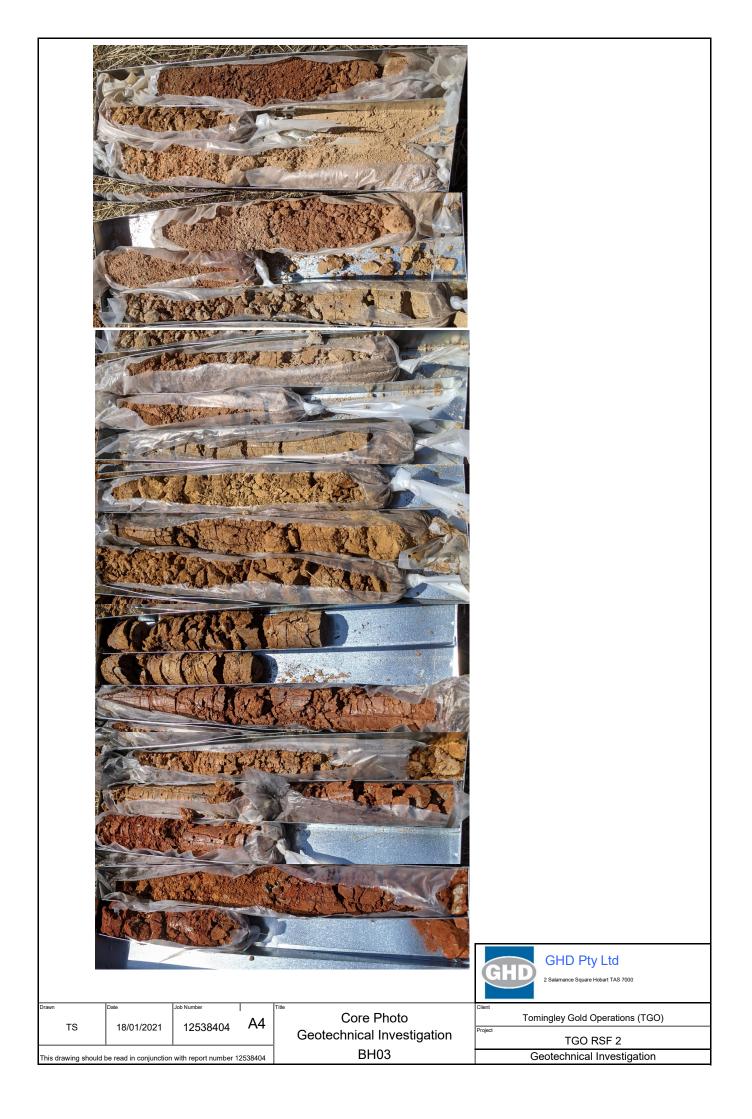
Clier Proje Loca	nt :	Tom Geo	FION LOG SHEET ingley Gold Operations technical Investigation ingley, NSW						LOCAT	ION NO. TP' SHEET 1 OF	
	tion :		015.1 E, 6393279.0 N GDA94	Surface RL	: 263	.9m			Pit Width : 0.8	Processed : T	AS
Cont	ractor	: TG	0	Machine : D	Doosa	n DX	225L	.c	Pit Length : 3	Checked : TAS	s
Date	:	19	Nov 20						Logged by : SG	Date : 22 Mar	21
XCA			MATERIAL						ADDITIONAL	DATA	
Scale (m) Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity of Particle Characteristics, Colour, Moisture Condition, Consiste Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/O		
1	0.80		TOPSOIL: Sandy Clay with rootlets plasticity, orange brown, w < PL CLAY with sand, trace gravel, high orange brown, sand is fine to coars gravel is fine grained, w < PL (Allu	n plasticity, se grained,	CL	D	Н	DCP PP BS PP	DCP (blows/100mm): 5,12,20,26 0.3m: PP = 550 kPa 0.5m: PP = 550 kPa BS (0.0 - 1.0m) 1.0m: PP > 600 kPa	5	
2 2 CNE	2.00 261.88		Sandy CLAY with gravel, high plas orange brown, sand is fine to coars gravel is fine grained, subround to w < PL (Alluvial)	se grained,	СН	D	Н	BS	BS (1.0 - 2.0m) BS (2.0 - 3.0m)		
5	<u>3.20</u> 260.68		Effective refusal with 0.6m wide, to bucket	pothed							
Ļ											
etail	s of a	bbrev	eets for iations ptions GHD Pt 2 Salamar T: 61 3 62 CLIENTS	t y Ltd rca Square Hob 2100 600 F:	Dart T/	AS 70	01, H	lobart	Jo	b No. 12538404	

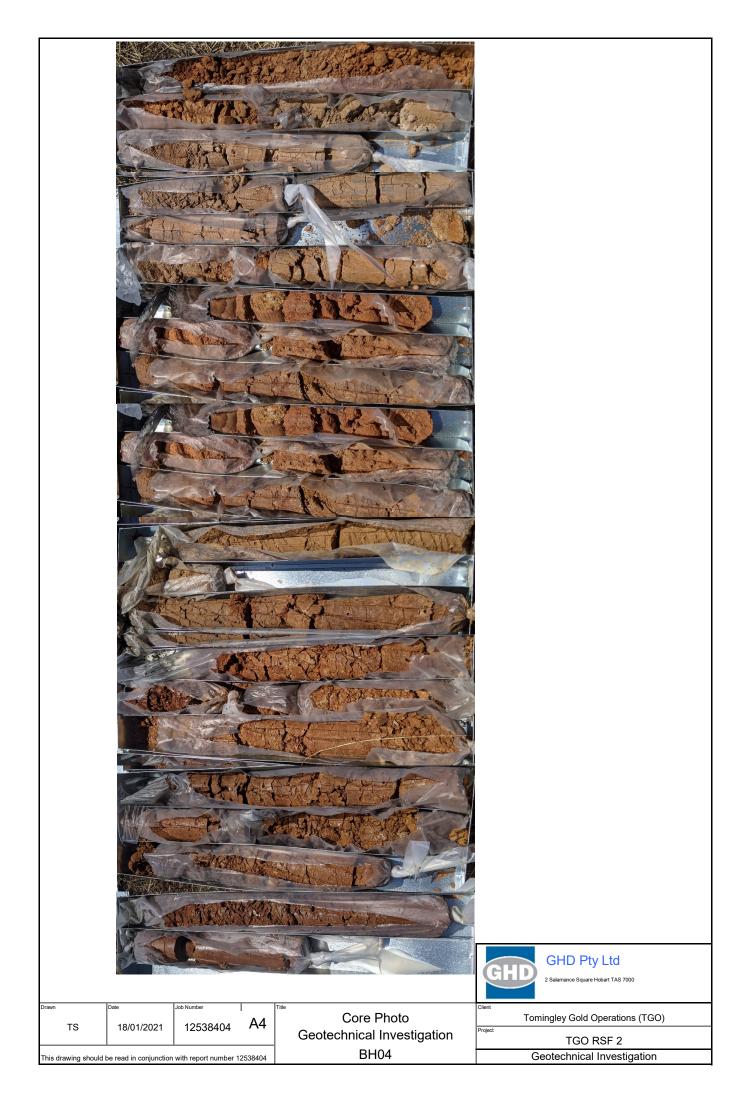
Clie Proj Loca	ec	t:	Geot	ingley Gold Operations technical Investigation ingley, NSW						LOCAT	TON NO. TP SHEET 1 O	
Posi				839.7 E, 6393455.0 N GDA94	Surface RL	. : 264	.1m			Pit Width : 0.8	Processed : C	P
Con	tra	actor	: TGC	0	Machine :	Doosa	n DX	225L	.c	Pit Length : 3	Checked : TA	S
Date	e :		19 1	Nov 20						Logged by : SG	Date : 22 Mar	21
XCA	4			MATERIAL	-		1			ADDITIONAL	DATA	
Scale (m) Water	VVale	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consist Structure		Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/C Insitu tes		
1 2 INC	2	0.60 2.00 262.05		Topsoil: Sandy CLAY with rootlets low to medium plasticity, orange y sand is fine to coarse grained, w < Sandy CLAY, trace gravel, medium pale grey brown, black and orange sand is fine to coarse grained, gra grained, well rounded to sub angu no obvious grading, polymictic (All Sandy CLAY with gravel, medium pale yellow orange brown, fine to r grained well rounded to sub angula no obvious grading, cemented by o (Alluvial)	ellow brown, ← PL n plasticity, e mottled, vel is fine lar gravel, uvial) plasticity, medium ar gravels,		W <pi W<pi< td=""><td></td><td>DCP SV PP BS PP BS BS</td><td>DCP (blows/100mm): 5,5,10,14 SV 113/55 kPa, small blade PP 550 kPa PP 600 kPa SV refused, small blade PP > 600 kPa</td><td>,14,19,25</td><td></td></pi<></pi 		DCP SV PP BS PP BS BS	DCP (blows/100mm): 5,5,10,14 SV 113/55 kPa, small blade PP 550 kPa PP 600 kPa SV refused, small blade PP > 600 kPa	,14,19,25	
3	2	<u>3.00</u> 261.05	· · · · · · · · · · · · · · · · · · ·	END OF HOLE at 3.0 m, effective	e refusal							
ŀ												
eta	ils	of al	bbrevi	eets for iations ptions GHD P 2 Salama T: 61 3 6 CLIENTS	ty Ltd nca Square Ho 2100 600 F:	bart T/ 61 3 (AS 70	01, ⊦ 0601	lobart	Jo	ob No. 12538404	

Clie Pro Loc	ent ojec	::	Tom Geot	FION LOG SHEET ingley Gold Operations technical Investigation ingley, NSW						LOCAT	ION NO. TP2	
		on :		819.5 E, 6393320.0 N GDA94	Surface RL	. : 263.	.7m			Pit Width : 0.8	Processed : CP	
Coi	ntra	actor	: TGC	0	Machine : I	Doosa	n DX	225L	.C	Pit Length : 3.5	Checked : TAS	
Dat	te :		19 N	Nov 20						Logged by : SG	Date : 22 Mar 21	1
хс	A			MATERIAL	-					ADDITIONAL	DATA	
scale (m)	Water	Depth / (RL) metres	Graphic Log	Description Soil Name (USC Symbol) Other Minor Components, Plasticity Particle Characteristics, Colour, Moisture Condition, Consiste Structure	ency,	Group Symbol	Moisture Condition	Consistency / Relative Density	Samples & Tests	Comments/Ot Insitu test	results	
				Topsoil: Sandy Clay with rootlets, or brown, low plasticity, fine to mediu sand, w < PL	orange ım grained	CL	D	F	DCP SV	DCP (blows/100mm): 3,4,7,9,18, SV 90/22 kPa, small blade	26	
									PP	PP >600 kPa		
				From 0.5 m: more compacted, dec proportion of organics	creased				BS PP	PP >600 kPa		
1	;	1.00 262.70		Sandy CLAY, pale orange yellow b medium plasticity, fine to medium sand, w < PL (Alluvial)	prown, grained	CI	D	F- St	SV PP	SV refusal, small blade PP 600 kPa		
									BS			
2	;	2.00 261.70		Sandy CLAY with Gravel, medium yellow brown, fine to medium grair fine to coarse grained well rounde angular gravel, no grading, polymic	ned sand, d to sub	- <u>c</u> i-	D	 F		Possible extremely weathered gra feldspar and plagioclase?	anite clasts with quartz,	:
	GNE			(Alluvial)					BS			
3												
		<u>3.50</u> 260.20	· · · · · · · · · · · · · · · · · · ·						BS			
		∠ou.20		END OF HOLE at 3.5 m, effective	refusal							
L												
-												
				ets for GHD Pt 2 Salamar	nča Square Ho	bart TA	AS 70	01, H	lobart	Jo	b No.	
eta				iations GHD 2 Salamar T: 61 3 6 CLIENTS	2100 600 F:	6136	52100	601			12538404	



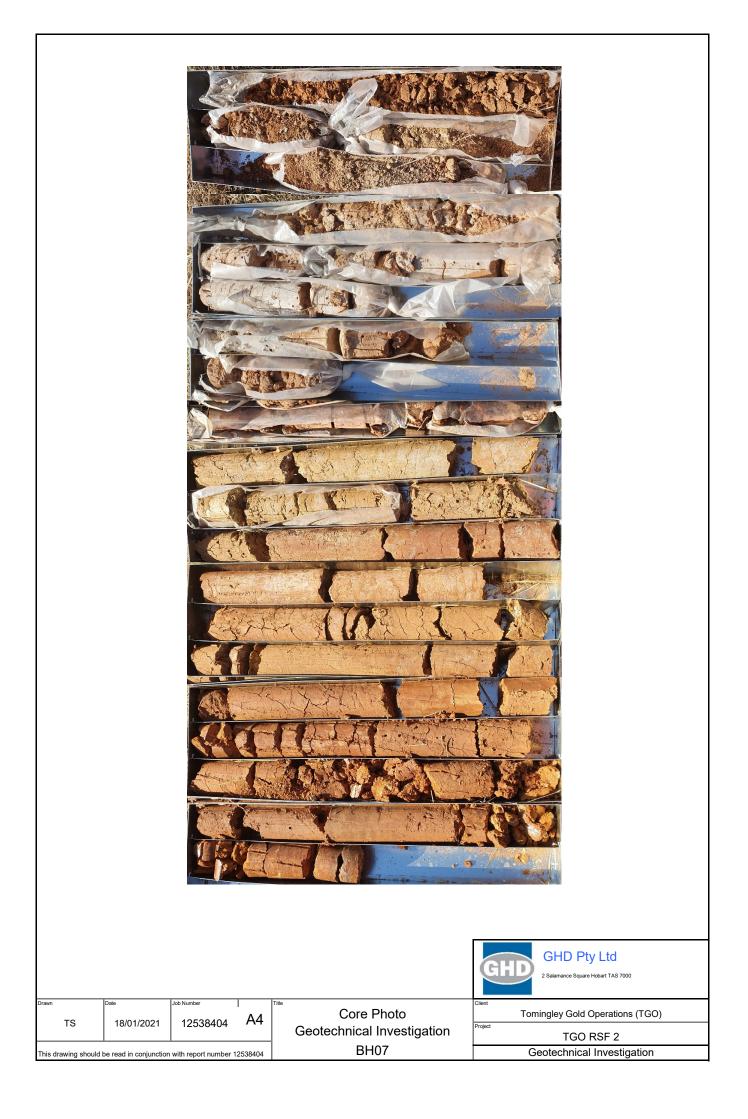


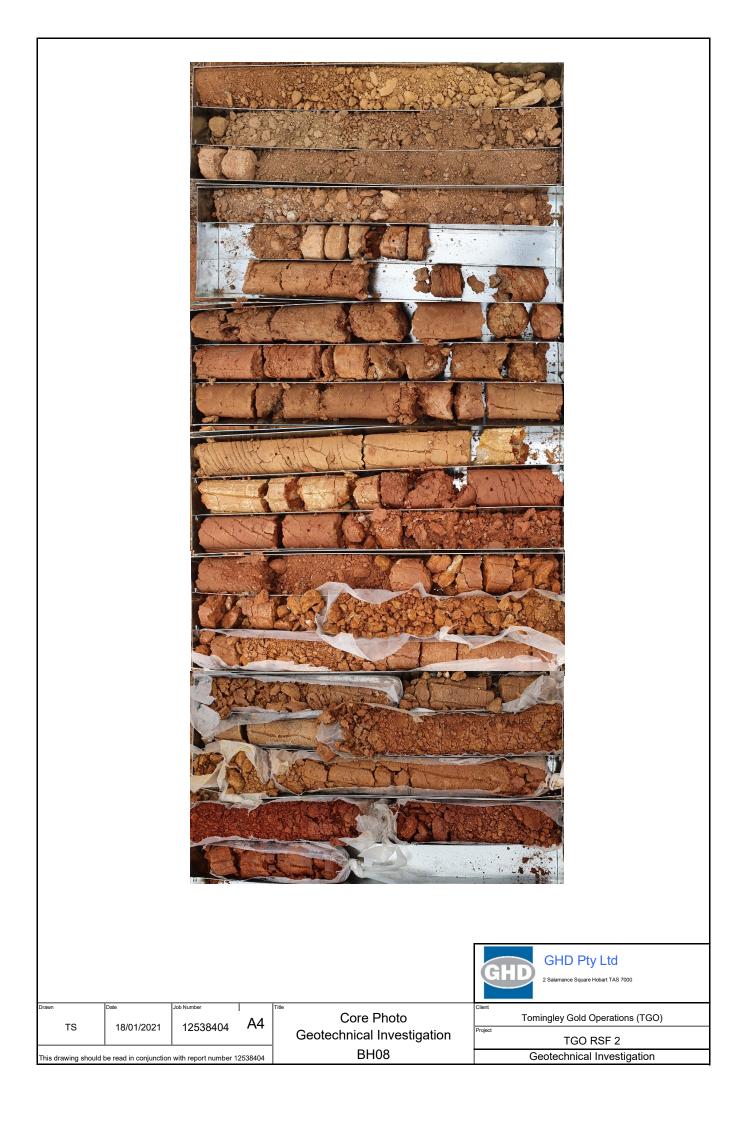




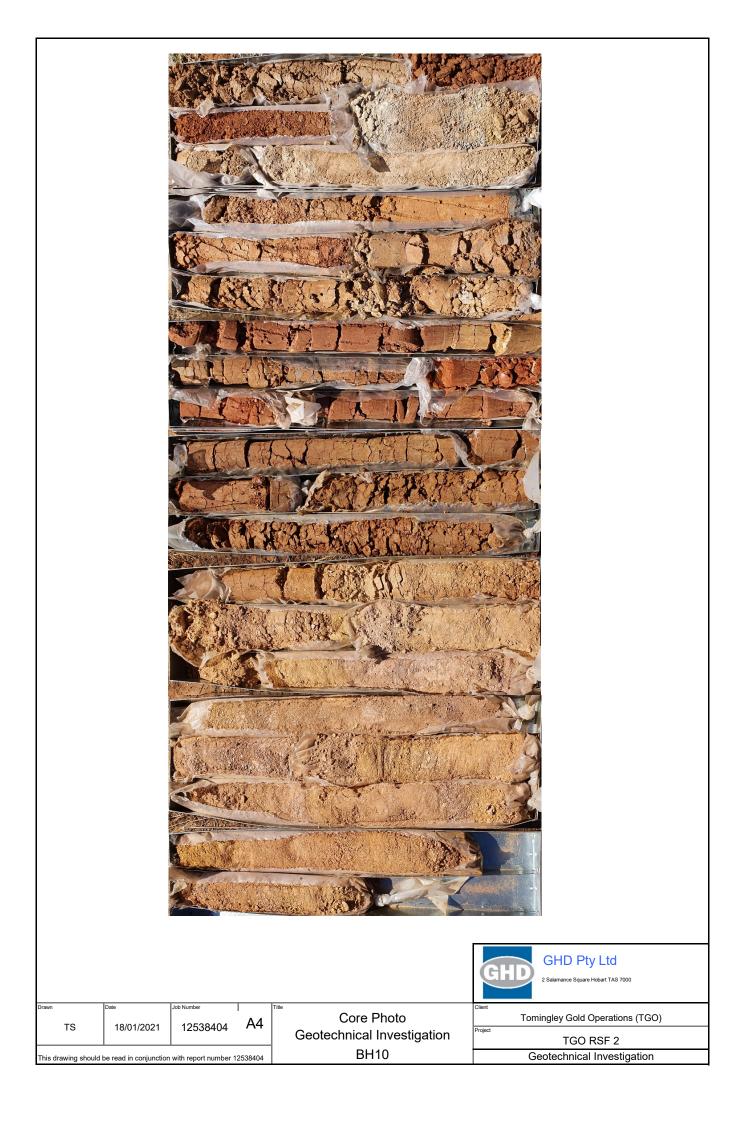


GHD Pty Ltd GHD 2 Salamance Square Hobart TAS 7000 Core Photo Tomingley Gold Operations (TGO) A4 ΤS 18/01/2021 12538404 Geotechnical Investigation TGO RSF 2 BH06 Geotechnical Investigation This drawing should be read in conjunction with report number 12538404









Appendix D – Laboratory Testing

Client : Tomingley Gold Operations Job No: 12538404 Project : RSF 2 GI Borehole No : See below Location : Tomingley, NSW Depth : See below Sample No : See below **TEST METHOD**: Client ID : AS1289.2.1.1 See below Bulk Density Dry Density Depth FMC GHD **BH/Test Pit ID** (m) (%) (t/m³) (t/m³) Sample No TP01 1.0-2.0 10.0 2.038 1.852 SYD21-0001-01 TP03 2.0-3.0 SYD21-0001-02 16.9 1.962 1.679 TP04 1.0-2.0 13.9 2.075 1.822 SYD21-0001-03 TP07 1.0-2.0 17.7 1.829 1.554 SYD21-0001-04 TP10 0.0-1.0 11.5 1.892 1.697 SYD21-0001-05 TP12 3.0-4.0 14.1 1.936 1.697 SYD21-0001-08 TP14 1.0-2.0 2.051 SYD21-0001-09 14.1 1.798 1.0-2.0 TP18 14.7 1.932 1.685 SYD21-0001-11 Comments : Bulk density performed on soil lumps GHD Tested By: DW 5/43 Herbert St Artarmon, NSW 2065 Date Tested: e 15/01/2021 Telephone: (02) 9462 4860 Fax: (02) 9462 4710 **GEOTECHNICAL TESTING SERVICES** Checked By: GV This document is issued in accordance with NATA's accreditation requirements. Approved Þ Accredited for compliance with ISO/IEC 17026 NATA Signatory Laboratory Accreditation Number: 679 D. Brooke

Natural Moisture / Density Report

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Report No:

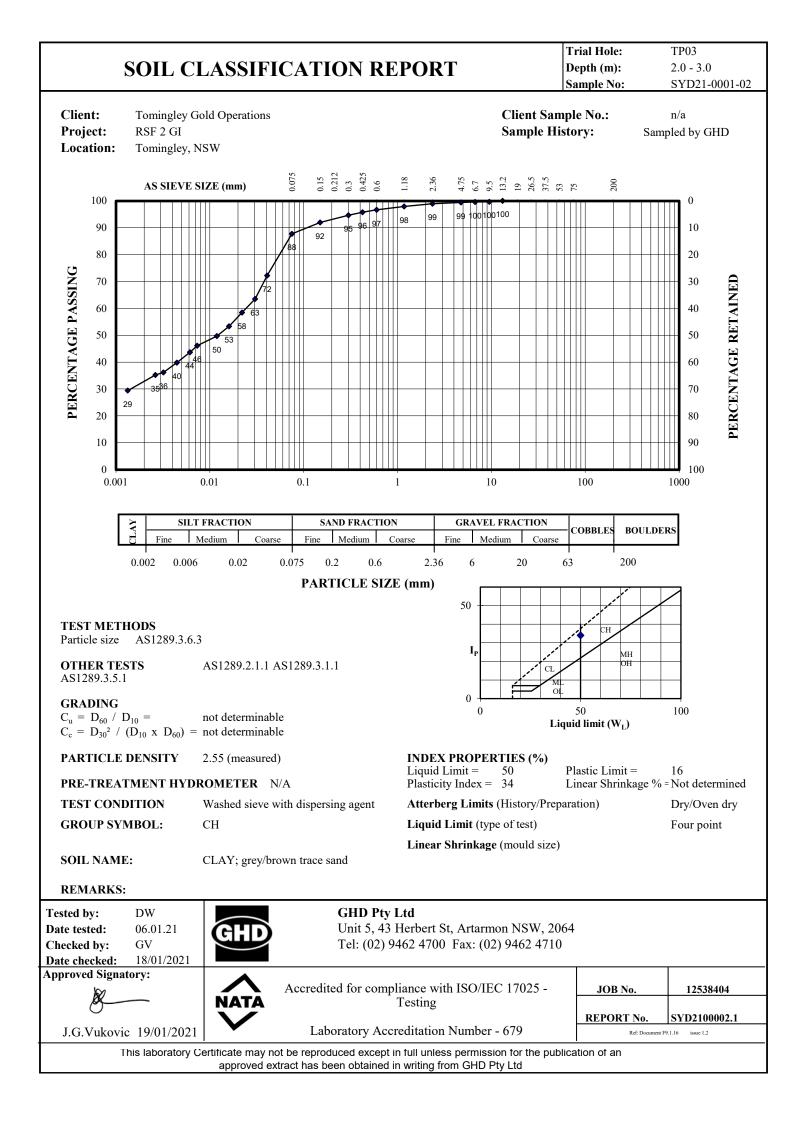
SYD2100076

Date :

19/01/2021



Materi	al Test Rep	ort				Report No: S	YD2100001 Issue No: 1
Client:	Tomingley Gold Ope RSF 2 GI Tomingley NSW	erations Pty Ltd			NATA	Accredited for compliance with IS Testing	SO / IEC 17025 -
Project:	12538404				No: 679	on Approved Signatory: Jure G Date of Issue: 19/01/2021 T SHALL NOT BE REPRODUCED	
Sample D	etails				Particle S	ize Distribution	
GHD Sample Date Sample Sampled By BH / TP No. Depth (m) Soil Descrip	e No SYD21-00 ed 18/11/202 Sampled I TP01 1.0 - 2.0	0	n trace gravel		Method: Drying by: Date Tested: Note: Sieve Size 19.0mm 13.2mm 9.5mm 6.7mm 4.75mm	AS 1289.3.6.1 Oven 5/01/2021 Sample Washed % Passing 100 100 99 99 99	Limits
Other Tes	st Results				2.36mm 1.18mm	96 90	
Description		Method	Result	Limits	600µm	82	
Moisture Cor Date Tested Sample Histo Preparation Linear Shrink	ory	AS 1289.2.1.1 AS 1289.1.1 AS 1289.1.1 AS 1289.3.4.1	4/01/2021 Oven-dried Dry Sieved		425μm 300μm 150μm 75μm	78 74 66 59	
Mould Lengtl Crumbling Curling Cracking Liquid Limit (Method Plastic Limit Plasticity Ind Avg Particle Den	h (mm) %) (%)	AS 1289.3.1.1 AS 1289.3.2.1 AS 1289.3.3.1 AS 1289.3.5.1	No No 31 Four Point 14 17		_		
Avg Particle Den Temperature – r	sity - retained 2.36mm (°C) retained 2.36mm (°C) Density (g/cm ³)		2.57 2.57 26 2.53		Chart		
Emerson Cla Soil Descript Type of Wate Temperature Date Tested	ass Number ion er e of Water (°C)		2 sandy CLAY Distilled 23 13/01/2021		% Passing		
Standard MI Standard OI Retained Sie Oversize Ma Curing Time LL Method Date Tested	MC (%) eve (mm) terial (%)	AS 1289.5.1.1 - 2	2017 1.84 14.5 19 0 79 5 1289.3.1.1 8/01/2021			under state and stat	67mm 86mm 132mm 160mm





wateria	Test Report	Issue No: 1
F	omingley Gold Operations Pty Ltd RSF 2 GI omingley NSW	Accredited for compliance with ISO / IEC 17025 - Testing
Project: 1	2538404	NATA Accreditation Approved Signatory: Jure G Vukovic No: 679 Date of Issue: 19/01/2021 THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FULL

GHD Sample No Date Sampled Sampled By BH / TP No. Depth (m) Soil Description

SYD21-0001-02 18/11/2020 Sampled by GHD TP03 2.0 - 3.0 CLAY; grey/brown trace sand

Test Results

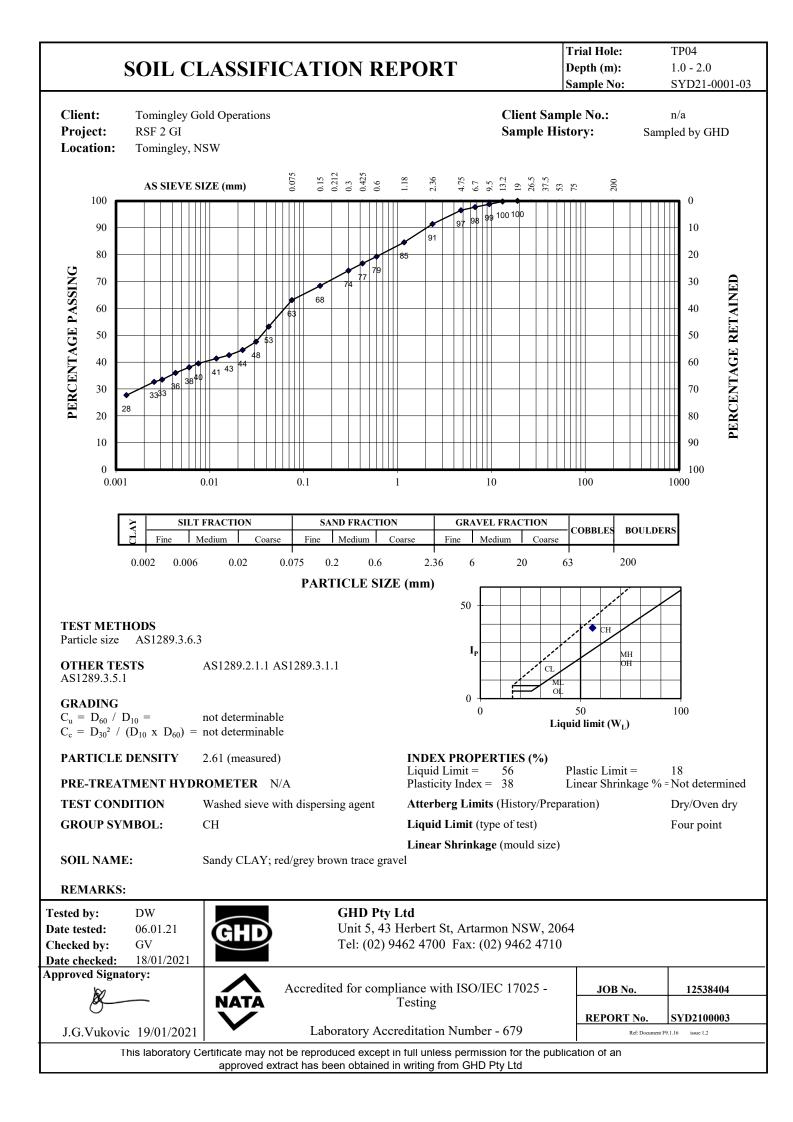
Description	Method	Result	Limit
Moisture Content (%)	AS 1289.2.1.1	16.9	
Date Tested		4/01/2021	
Sample History	AS 1289.1.1	Oven-dried	
Preparation	AS 1289.1.1	Dry Sieved	
Linear Shrinkage (%)	AS 1289.3.4.1	N/A	
Mould Length (mm)			
Crumbling		No	
Curling		No	
Cracking		No	
_iquid Limit (%)	AS 1289.3.1.1	50	
Vethod		Four Point	
Plastic Limit (%)	AS 1289.3.2.1	16	
Plasticity Index (%)	AS 1289.3.3.1	34	
Avg Particle Density - passing 2.36mm (g/cm ³)	AS 1289.3.5.1	2.55	
Temperature - passing 2.36mm (°C)		26	
Avg Particle Density - retained 2.36mm (g/cm ³)		2.57	
Femperature - retained 2.36mm (°C)		26	
Soil Particle Density (g/cm³)		2.57	
Emerson Class Number	AS 1289.3.8.1	2	
Soil Description		CLAY trace sand	
Гуре of Water		Distilled	
Femperature of Water (°C)		23	
Date Tested		13/01/2021	
Standard MDD (t/m³)	AS 1289.5.1.1 - 2017	1.66	
Standard OMC (%)		20.0	
Retained Sieve (mm)		19	
Oversize Material (%)		0	
Curing Time (h)		175	
LL Method		AS 1289.3.1.1	



Materi	al Test Repo	rt	Report No: SYD2100002 Issue No: 1
Client: Project:	Tomingley Gold Operat RSF 2 GI Tomingley NSW 12538404	ons Pty Ltd	Accredited for compliance with ISO / IEC 17025 - Testing NATA Accreditation Approved Signatory: Jure G Vukovic No: 679 Date of Issue: 19/01/2021 THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FULL
Sample D	etails		
GHD Sampl Date Sampl Sampled By BH / TP No. Depth (m) Soil Descrip	ed 18/11/2020 Sampled by G TP03 2.0 - 3.0	_	

Test Results

Description	Method	Result	Limits
Date Tested		7/01/2021	
Coef of Permeability (m/s)	AS 1289.6.7.3	2 E-10	
Mean Stress Level (kPa)		30	
Permeant Used		Syd tap water	
Length (mm)		75.7	
Diameter (mm)		64.3	
Length/Diameter Ratio		1.18	
Laboratory Moisture Ratio (%)		100.5	
Laboratory Density Ratio (%)		97.5	
CompactiveEffort		Standard	
Method of Compaction		Remoulded	
Surcharge Applied (kg)		0.0	
Pressure Applied (kPa)		10	
Oversize Sieve (mm)		9.5	
Percentage Oversize (%)		0.5	
Moisture Content (%)		24.9	
Date Tested		8/01/2021	





Materi	al Test Report	Report No: SYD2100003 Issue No: 1
Client:	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW	Accredited for compliance with ISO / IEC 17025 - Testing
Project:	12538404	NATA Accreditation Approved Signatory: Jure G Vukovic No: 679 Date of Issue: 19/01/2021 THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FULL
Sample D	Details	•
GHD Sampl	e No SYD21-0001-03	

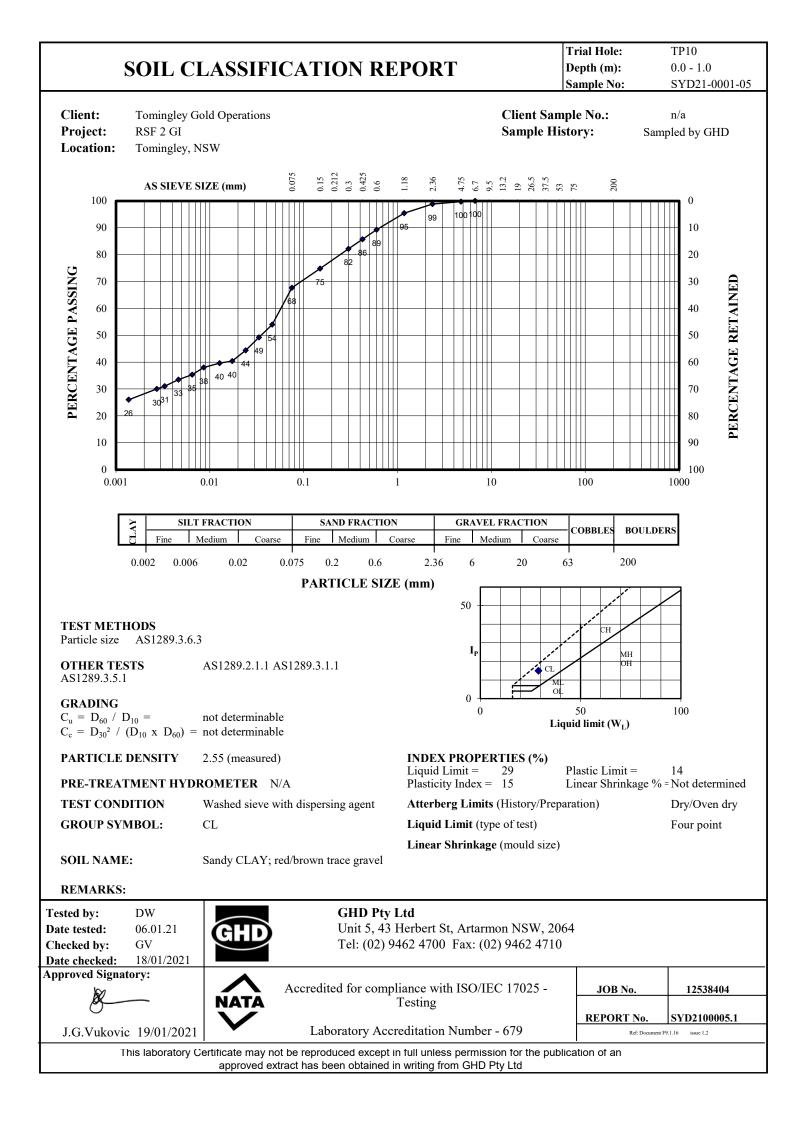
GHD Sample No	SYD21-0001-03
Date Sampled	18/11/2020
Sampled By	Sampled by GHD
BH / TP No.	TP04
Depth (m)	1.0 - 2.0
Soil Description	Sandy CLAY; red/grey brown trace gravel

Test Results

Description	Method	Result	Limits
Moisture Content (%)	AS 1289.2.1.1	13.9	
Date Tested		4/01/2021	
Sample History	AS 1289.1.1	Oven-dried	
Preparation	AS 1289.1.1	Dry Sieved	
Linear Shrinkage (%)	AS 1289.3.4.1	N/A	
Mould Length (mm)			
Crumbling		No	
Curling		No	
Cracking		No	
Liquid Limit (%)	AS 1289.3.1.1	56	
Method		Four Point	
Plastic Limit (%)	AS 1289.3.2.1	18	
Plasticity Index (%)	AS 1289.3.3.1	38	
Avg Particle Density - passing 2.36mm (g/cm ³)	AS 1289.3.5.1	2 <u>.</u> 61	
Temperature - passing 2.36mm (°C)		26	
Avg Particle Density - retained 2.36mm (g/cm ³)		2.50	
Temperature - retained 2.36mm (°C)		26	
Soil Particle Density (g/cm ³)		2.50	
Emerson Class Number	AS 1289.3.8.1	2	
Soil Description		sandy CLAY	
Type of Water		Distilled	
Temperature of Water (°C)		23	
Date Tested		13/01/2021	
Standard MDD (t/m ³)	AS 1289.5.1.1 - 2017	1.70	
Standard OMC (%)		18.0	
Retained Sieve (mm)		19	
Oversize Material (%)		0	
Curing Time (h)		169	
LL Method		AS 1289.3.1.1	



Materia	al Test Rep	ort				•	SYD2100004 Issue No: 1
Client:	Tomingley Gold Ope RSF 2 GI Tomingley NSW	rations Pty Ltd			NATA	Accredited for compliance wit Testing	h ISO / IEC 17025 -
Project:	12538404				No: 679	on Approved Signatory: Jure Date of Issue: 19/01/20 T SHALL NOT BE REPRODUCI	21
Sample De	etails				Particle S	ize Distribution	
GHD Sample Date Sampled Sampled By BH / TP No. Depth (m) Soil Descript	No SYD21-00 d 18/11/202 Sampled t TP07 1.0 - 2.0	0	race gravel		Method: Drying by: Date Tested: Note: Sieve Size 53.0mm 37.5mm 26.5mm 19.0mm 13.2mm	AS 1289.3.6.1 Oven 5/01/2021 Sample Washed % Passing 100 99 98 98 98 98	Limits
Other Test	t Results				9.5mm	97	
Description		Method	Result	Limits	6.7mm 4.75mm	95 94	
Moisture Cont Date Tested Sample Histor	. ,	AS 1289.2.1.1		Liiiits	2.36mm 1.18mm 600µm	94 92 89 85	
Preparation Linear Shrinka Mould Length Crumbling Curling Cracking Liquid Limit (% Method Plastic Limit (% Plasticity Inde	(mm) 6) %) x (%)	AS 1289.1.1 AS 1289.3.4.1 AS 1289.3.1.1 AS 1289.3.2.1 AS 1289.3.2.1 AS 1289.3.3.1	No No 48 Four Point 18 30		425μm 300μm 150μm 75μm	83 80 75 69	
Temperature - pa Avg Particle Densi	ty - passing 2.36mm (g/cm³) assing 2.36mm (°C) ty - retained 2.36mm (g/cm³)	AS 1289.3.5.1	26 2.29		Chart		
Soil Particle D Emerson Clas	s Number	AS 1289.3.8.1			% Passing		
Soil Description Type of Water Temperature of Date Tested	of Water (°C)		Y with sand Distilled 23 13/01/2021		90 80 70 60 50		
Standard MD Standard OM Retained Siev Oversize Mate Curing Time (I LL Method	C (%) re (mm) erial (%)	AS 1289.5.1.1 - 2 AS	21.0 19 2 97 6 1289.3.1.1			under state and	95mm 132mm 132mm 132mm 132mm 132mm 132mm
Date Tested			7/01/2021				





Materi	— al Test Report	Report No: SYD2100005 Issue No: 1
Client:	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW	Accredited for compliance with ISO / IEC 17025 - Testing
Project:	12538404	NATA Accreditation Approved Signatory: Jure G Vukovic No: 679 Date of Issue: 19/01/2021 THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FULL
Sample D		THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FU

GHD Sample No Date Sampled Sampled By BH / TP No. Depth (m) Soil Description

SYD21-0001-05 18/11/2020 Sampled by GHD TP10 0.0 - 1.0 Sandy CLAY; red/brown trace gravel

Test Results

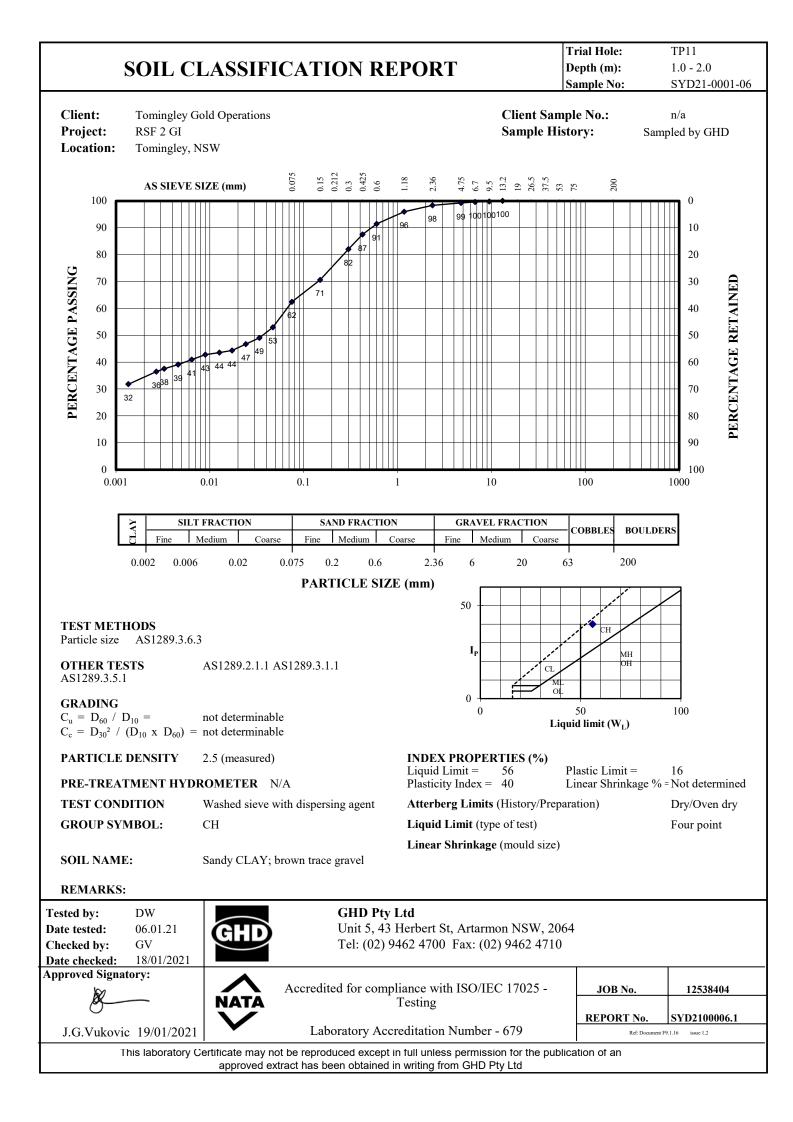
Description	Method	Result	Limits
Moisture Content (%)	AS 1289.2.1.1	11.5	
Date Tested		4/01/2021	
Sample History	AS 1289.1.1	Oven-dried	
Preparation	AS 1289.1.1	Dry Sieved	
Linear Shrinkage (%)	AS 1289.3.4.1	N/A	
Mould Length (mm)			
Crumbling		No	
Curling		No	
Cracking		No	
Liquid Limit (%)	AS 1289.3.1.1	29	
Method		Four Point	
Plastic Limit (%)	AS 1289.3.2.1	14	
Plasticity Index (%)	AS 1289.3.3.1	15	
Avg Particle Density - passing 2.36mm (g/cm ³)	AS 1289.3.5.1	2.55	
Temperature - passing 2.36mm (°C)		26	
Avg Particle Density - retained 2.36mm (g/cm ³)		2.46	
Temperature - retained 2.36mm (°C)		26	
Soil Particle Density (g/cm³)		2.46	
Emerson Class Number	AS 1289.3.8.1	2	
Soil Description		Sandy CLAY	
Type of Water		Distilled	
Temperature of Water (°C)		23	
Date Tested		13/01/2021	
Standard MDD (t/m³)	AS 1289.5.1.1 - 2017	1 <u>.</u> 81	
Standard OMC (%)		15 <u>.</u> 0	
Retained Sieve (mm)		19	
Oversize Material (%)		0	
Curing Time (h)		79	
LL Method		AS 1289.3.1.1	



Materi	al Test Report	Report No: SYD2100005 Issue No: 1
Client: Project:	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW 12538404	Accredited for compliance with ISO / IEC 17025 - Testing
Sample D	etails	
GHD Sampl Date Sampl Sampled By BH / TP No. Depth (m) Soil Descrip	ed 18/11/2020 Sampled by GHD TP10 0.0 - 1.0	

Test Results

Description	Method	Result	Limits
Date Tested		7/01/2021	
Coef of Permeability (m/s)	AS 1289.6.7.3	2 E-10	
Mean Stress Level (kPa)		30	
Permeant Used		Syd tap water	
Length (mm)		74.4	
Diameter (mm)		63.5	
Length/Diameter Ratio		1.17	
Laboratory Moisture Ratio (%)		101.0	
Laboratory Density Ratio (%)		98.0	
CompactiveEffort		Standard	
Method of Compaction		Remoulded	
Surcharge Applied (kg)		0.0	
Pressure Applied (kPa)		10	
Oversize Sieve (mm)		9.5	
Percentage Oversize (%)		0.0	
Moisture Content (%)		17.9	
Date Tested		8/01/2021	





Materi	al Test Report	Report No: SYD2100006 Issue No: 1
Client:	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW	Accredited for compliance with ISO / IEC 17025 - Testing
Project:	12538404	NATA Accreditation Approved Signatory: Jure G Vukovic No: 679 Date of Issue: 19/01/2021 THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FULL

GHD Sample No Date Sampled Sampled By BH / TP No. Depth (m) Soil Description SYD21-0001-06 18/11/2020 Sampled by GHD TP11 1.0 - 2.0 Sandy CLAY; brown

Test Results

Description	Method	Result	Limits
Moisture Content (%)	AS 1289.2.1.1	17.4	
Date Tested		4/01/2021	
Sample History	AS 1289.1.1	Oven-dried	
Preparation	AS 1289.1.1	Dry Sieved	
Linear Shrinkage (%)	AS 1289.3.4.1	N/A	
Mould Length (mm)			
Crumbling		No	
Curling		No	
Cracking		No	
Liquid Limit (%)	AS 1289.3.1.1	56	
Method		Four Point	
Plastic Limit (%)	AS 1289.3.2.1	16	
Plasticity Index (%)	AS 1289.3.3.1	40	
Avg Particle Density - passing 2.36mm (g/cm ³)	AS 1289.3.5.1	2.50	
Temperature - passing 2.36mm (°C)		26	
Avg Particle Density - retained 2.36mm (g/cm ³)		2.41	
Temperature - retained 2.36mm (°C)		26	
Soil Particle Density (g/cm ³)		2.41	
Emerson Class Number	AS 1289.3.8.1	2	
Soil Description		Sandy CLAY	
Type of Water		Distilled	
Temperature of Water (°C)		23	
Date Tested		13/01/2021	
Standard MDD (t/m³)	AS 1289.5.1.1 - 2017	1.65	
Standard OMC (%)		20.0	
Retained Sieve (mm)		19	
Oversize Material (%)		0	
Curing Time (h)		169	
LL Method		AS 1289.3.1.1	



Material Test Report		Report No: SYD2100006 Issue No: 1
Client: Project:	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW 12538404	Accredited for compliance with ISO / IEC 17025 - Testing NATA Accreditation Approved Signatory: Jure G Vukovic No: 679 Date of Issue: 19/01/2021 THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FULL
Sample D	etails	
GHD Sampl Date Sampl Sampled B BH / TP No. Depth (m) Soil Descri	ed 18/11/2020 Sampled by GHD TP11 1.0 - 2.0	

Test Results

Description	Method	Result	Limits
Date Tested		7/01/2021	
Coef of Permeability (m/s)	AS 1289.6.7.3	4 E-11	
Mean Stress Level (kPa)		30	
Permeant Used		Syd tap water	
Length (mm)		74.2	
Diameter (mm)		63.5	
Length/Diameter Ratio		1.17	
Laboratory Moisture Ratio (%)		100.0	
Laboratory Density Ratio (%)		98.5	
CompactiveEffort		Standard	
Method of Compaction		Remoulded	
Surcharge Applied (kg)		0.0	
Pressure Applied (kPa)		10	
Oversize Sieve (mm)		9.5	
Percentage Oversize (%)		0.5	
Moisture Content (%)		23.8	
Date Tested		8/01/2021	



ateri	al Te	st R	еро	rt					Report No: SY	Issue No:
lient:	RSF 2 0 Toming	GI ley NSV		ons Pty Ltd				NATA	8	
roject:	125384	04							roved Signatory: Jure G e of Issue: 18/01/2021 _ NOT BE REPRODUCED E	
ample	Details					Othe	r Test I	Results		
HD Sampl ate Sampl ampled By H / TP No. epth (m) oil Descrip	ed , otion		2020 ed by GF 0 LAY; red/		n gravel	<u>Descrip</u>		Metho	od Resul	t Limits
vrticle S	ize Dis [.]	tributi	on					Method: Drying by: Date Tested:	AS 1289.3.6.1 Oven 5/01/2021	
100 I · · ·					••••		· · · · · ·	Note: Sample	Washed	
90 - · · · · 80 - · · · 60 - · · · 50 - · · · 40 - · · · 30 - · · · 10 - · · ·								Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600μm 425μm 300μm 150μm 75μm	% Passing 100 99 97 87 72 59 55 51 46 41	Limits
0	150µm	300µm	425µm	1.18mm	2.36mm	4.75mm	9.5mm			



Materia	al Test Rep	oort				-	: SYD210000 Issue No:
Client:	Tomingley Gold Ope	erations Pty Ltd				Accredited for compliance Testing	with ISO / IEC 17025 -
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					NAIA	×	
	Tomingley NSW					C)	
Project:	12538404					on Approved Signatory: Ju	re G Vukovic
Fiojeci.	12000404				No: 679		
					THIS DOCUMENT	Date of Issue: 19/01/2 T SHALL NOT BE REPRODU	
Sample D	etails				Particle S	ize Distributio	n
GHD Sample		01-08			Method:	AS 1289.3.6.1	
Date Sample					Drying by:	Oven	
Sampled By					Date Tested:	5/01/2021	
BH / TP No.	TP12						
Depth (m)	3.0 - 4.0				Note:	Sample Washed	
Soil Descrip		AY; grey/brown					
•					Sieve Size	% Passing	Limits
					4.75mm	100	
					2.36mm	100	
					1.18mm	99	
					600µm	98	
					425µm	96	
Other Tes	t Results				300µm 150µm	91 74	
Description		Method	Result	Limits	75µm	64	
Moisture Con	ntent (%)	AS 1289.2.1.1	14.1				
Date Tested			4/01/2021				
Sample Histo	ory		Oven-dried				
Preparation	(0()		Dry Sieved				
Linear Shrink		AS 1289.3.4.1	N/A				
Mould Length Crumbling	1 (mm)		No				
Curling			No No				
Cracking			No				
Liquid Limit (%)	AS 1289.3.1.1	31				
Method	70)		Four Point				
Plastic Limit ((%)	AS 1289.3.2.1	14				
Plasticity Inde		AS 1289 3 3 1	17				
	sity - passing 2.36mm (g/cm³)	AS 1289.3.5.1	2.50		-		
	bassing 2.36mm (°C)		26				
Avg Particle Dens	sity - retained 2.36mm (g/cm³)				Chart		
	etained 2.36mm (°C)						
	Density (g/cm³)		2.50		% Passing		
Date Tested			18/01/2021		100 F		
Emerson Cla		AS 1289.3.8.1	2		90		
Soil Descripti		Sa	andy CLAY		70		
Type of Wate			Distilled 23		60		
Date Tested	of Water (°C)		23 13/01/2021		50		
Standard MI)D (t/m³)	AS 1289.5.1.1 - 20					
Standard ML		AO 1203.0.1.1 - 20	16.5		20		
Retained Sie			19		10		
Oversize Mat			0		٥	mit mit mit	uu uu
Curing Time			79		R	년 응 왕 응 후 Sieve	2.33)
LL Method	× /	AS	1289.3.1.1				
I I Melnoo							

N/A



Materi	al Test Report				Report No:	SYD2100008 Issue No: 1
Client:	Tomingley Gold Operations Pty RSF 2 GI Tomingley NSW	Ltd		NATA	Accredited for compliance wi Testing	th ISO / IEC 17025 -
Project:	12538404			No: 679	n Approved Signatory: Jure Date of Issue: 19/01/20 SHALL NOT BE REPRODUC	21
Sample D	etails			Particle Si	ize Distribution	
GHD Sample Date Sample Sampled By BH / TP No. Depth (m) Soil Descrip	e No SYD21-0001-08 ed 18/11/2020 Sampled by GHD TP12 3.0 - 4.0	own		Method: Drying by: Date Tested: Note: Sieve Size 4.75mm 2.36mm 1.18mm 600µm 425µm	AS 1289.3.6.1 Oven	Limits
Other Tes				300µm 150µm	91 74	
Mean Stress Permeant Us Length (mm) Diameter (mm Length/Diam Laboratory M Laboratory D CompactiveE Method of Co Surcharge A Pressure App Oversize Sie Percentage O	m) eter Ratio loisture Ratio (%) ensity Ratio (%) Effort ompaction oplied (kg) olied (kPa) ve (mm) Oversize (%)	30 Syd tap water 74.8 63.5 1.18 101.0 97.5 Standard Remoulded 0.0 10 9.5 0.0	Limits	_ 75μm -	64	
Moisture Cor		19.6		Chart		
Date Tested		8/01/2021		% Passing		3.30000



Materi	al Test Rep	ort					SYD2100009 Issue No: 1
Client:	Tomingley Gold Ope RSF 2 GI Tomingley NSW	rations Pty Ltd			NATA	Accredited for compliance wi	th ISO / IEC 17025 -
Project:	12538404				No: 679	on Approved Signatory: Jure Date of Issue: 19/01/20 I SHALL NOT BE REPRODUC	21
Sample D	etails				Particle S	ize Distribution	
GHD Sample Date Sample Sampled By BH / TP No. Depth (m) Soil Descrip	e No SYD21-00 ed 18/11/202 Sampled t TP14 1.0 - 2.0	0	own trace grave	el	Method: Drying by: Date Tested: Note: Sieve Size 13.2mm 9.5mm 6.7mm 4.75mm 2.36mm	AS 1289.3.6.1 Oven 5/01/2021 Sample Washed % Passing 100 100 100 100 98	Limits
Other Tes	st Results				1.18mm 600µm	91 77	
Description		Method	Result	Limits	425µm	67	
Temperature - p Avg Particle Den	ory (age (%) n (mm) %) (%) ex (%) sity - passing 2.36mm (g/cm³) passing 2.36mm (°C) sity - retained 2.36mm (g/cm³)	AS 1289.2.1.1 AS 1289.1.1 AS 1289.1.1 AS 1289.3.4.1 AS 1289.3.4.1 AS 1289.3.1.1 AS 1289.3.2.1 AS 1289.3.3.1 AS 1289.3.5.1	4/01/2021 Oven-dried Dry Sieved N/A No No 44 Four Point 15 29 2.61 26 2.47		300μm 150μm 75μm	59 48 44	
	etained 2.36mm (°C) Density (g/cm³)		26 2.61				
Emerson Cla Soil Descript Type of Wate	ss Number ion of Water (°C) DD (t/m³) MC (%) ve (mm) terial (%)	AS 1289.5.1.1 - 2	andy CLAY Distilled 23 13/01/2021				Comm

Comments

N/A



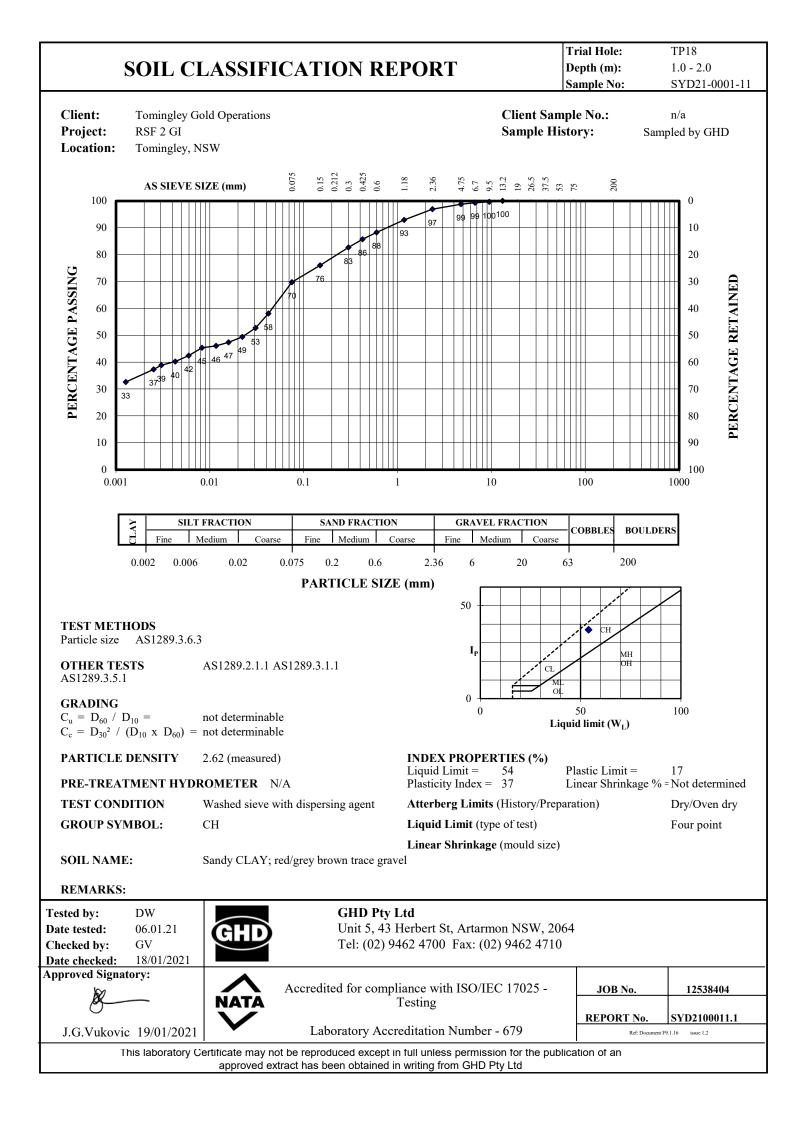
Materia	l Test Rep	ort				Report No:	SYD2100010
Client:	Tomingley Gold Ope RSF 2 GI Tomingley NSW	rations Pty Ltd			NATA	Testing	
Project:	12538404				No: 679	n Approved Signatory: Jur Date of Issue: 19/01/20 SHALL NOT BE REPRODUC	21
Sample De GHD Sample Date Sampled Sampled By BH / TP No. Depth (m) Soil Descripti	No SYD21-00 1 18/11/202 Sampled b TP17 1.0 - 2.0	0	trace gravel		Particle S Method: Drying by: Date Tested: Note: Sieve Size 19.0mm 13.2mm 9.5mm 6.7mm 4.75mm	ize Distribution AS 1289.3.6.1 Oven 5/01/2021 Sample Washed % Passing 100 99 98 97 96	Limits
Dther Test Description Moisture Conte Date Tested Sample History Preparation Linear Shrinka Mould Length Crumbling Curling Cracking Liquid Limit (%	ent (%) y ige (%) (mm)		Result 14.4 4/01/2021 Oven-dried Dry Sieved N/A No No No S0	Limits	2.36mm 1.18mm 600µm 425µm 300µm 150µm 75µm	95 91 86 83 79 72 66	
Method Plastic Limit (% Plasticity Inde> Date Tested Avg Particle Densit Temperature - pas Avg Particle Densit	%) × (%) y - passing 2.36mm (g/cm ³) ssing 2.36mm (°C) y - retained 2.36mm (°C) ensity (g/cm ³) s Number n of Water (°C) D (t/m ³) C (%) e (mm) prial (%)	AS 1289.3.2.1 AS 1289.3.3.1 AS 1289.3.5.1 AS 1289.3.5.1	Four Point 16 34 18/01/2021 2.62 26 2.42 26 2.61 2 andy CLAY Distilled 23 13/01/2021		Chart % Passing		

Comments

N/A



Materi	al Tes	st Report				Report No:	SYD2100010 Issue No: 1
Client:		ey Gold Operations Pty Lt	d		NATA	Accredited for compliance w Testing	th ISO / IEC 17025 -
Project:	1253840	14			No: 679	Date of Issue: 19/01/20 T SHALL NOT BE REPRODUC	21
Sample D	etails				Particle S	ize Distribution	ĭ
GHD Sample Date Sample Sampled By BH / TP No. Depth (m) Soil Descrip	e No ed	SYD21-0001-10 18/11/2020 Sampled by GHD TP17 1.0 - 2.0 Sandy CLAY; grey/brow	n trace gravel		Method: Drying by: Date Tested: Note: Sieve Size	AS 1289.3.6.1 Oven	Limits
Other Tes	st Resul	ts			19.0mm 13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm	100 99 98 97 96 95 91	
Description Date Tested		Method	Result 7/01/2021	Limits	600µm 425µm	86 83	
					150μm 75μm	72 66	
					Chart		
					% Passing		4.55mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.57mm B.





Materi	al Test Report	Report No: SYD2100011 Issue No: 1
Client:	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW	Accredited for compliance with ISO / IEC 17025 - Testing
Project:	12538404	NATA Accreditation Approved Signatory: Jure G Vukovic No: 679 Date of Issue: 19/01/2021 THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FULL

CHD Sample N

GHD Sample No	SYD21-0001-11
Date Sampled	18/11/2020
Sampled By	Sampled by GHD
BH / TP No.	TP18
Depth (m)	1.0 - 2.0
Soil Description	Sandy CLAY; red/grey brown trace gravel
-	

Test Results

Description	Method	Result	Limits
Moisture Content (%)	AS 1289.2.1.1	14.7	
Date Tested		4/01/2021	
Sample History	AS 1289.1.1	Oven-dried	
Preparation	AS 1289.1.1	Dry Sieved	
₋inear Shrinkage (%)	AS 1289.3.4.1	N/A	
Mould Length (mm)			
Crumbling		No	
Curling		No	
Cracking		No	
Liquid Limit (%)	AS 1289.3.1.1	54	
Vlethod		Four Point	
Plastic Limit (%)	AS 1289.3.2.1	17	
Plasticity Index (%)	AS 1289.3.3.1	37	
Avg Particle Density - passing 2.36mm (g/cm ³)	AS 1289.3.5.1	2.62	
Temperature - passing 2.36mm (°C)		26	
Avg Particle Density - retained 2.36mm (g/cm ³)		2.50	
Femperature - retained 2.36mm (°C)		26	
Soil Particle Density (g/cm³)		2.50	
Emerson Class Number	AS 1289.3.8.1	2	
Soil Description		Sandy CLAY	
Гуре of Water		Distilled	
Femperature of Water (°C)		23	
Date Tested		13/01/2021	
Standard MDD (t/m³)	AS 1289.5.1.1 - 2017	1.62	
Standard OMC (%)		18.5	
Retained Sieve (mm)		19	
Oversize Material (%)		0	
Curing Time (h)		169	
L Method		AS 1289.3.1.1	



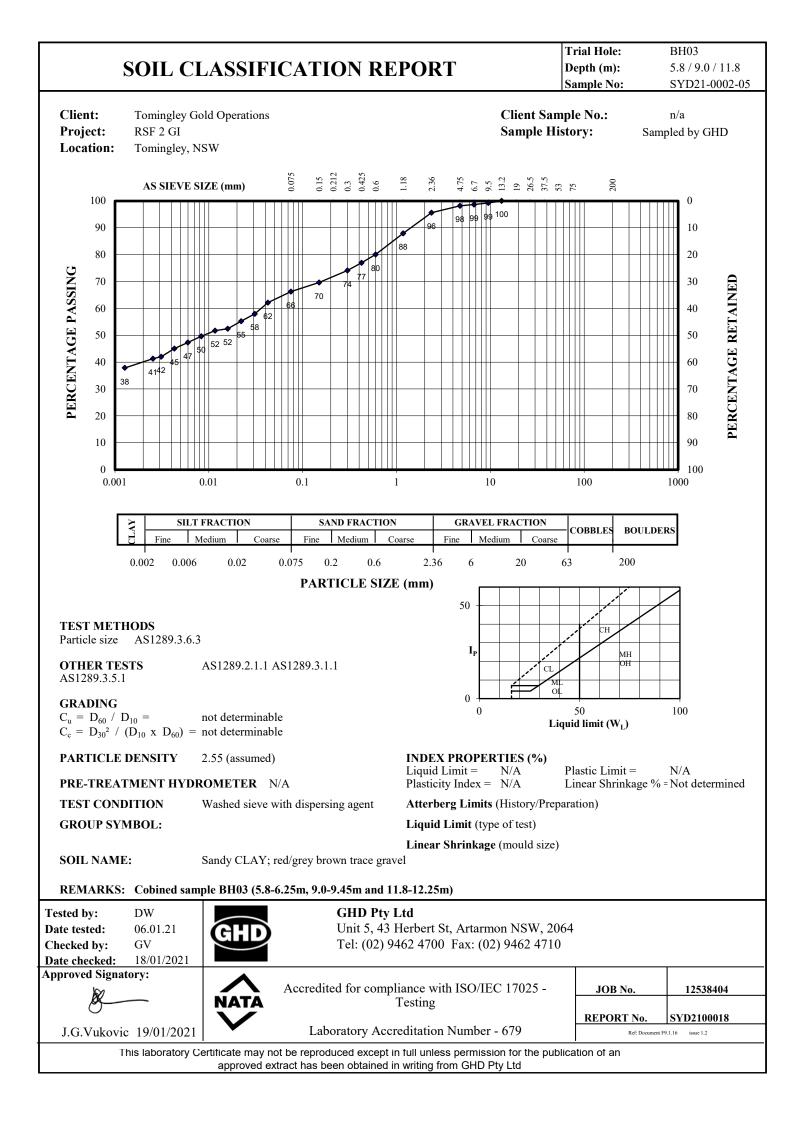
Materi	al Te	st Report	Report No: SYD2100011 Issue No: 1			
Client: Project:	RSF 2	ley NSW	Accredited for compliance with ISO / IEC 17025 - Testing NATA Accreditation Approved Signatory: Jure G Vukovic No: 679 Date of Issue: 19/01/2021 THIS DOCUMENT SHALL NOT BE REPRODUCED EXCEPT IN FULL			
Sample D)etails					
GHD Sampl Date Sampl Sampled By BH / TP No. Depth (m) Soil Descrip	ed /	SYD21-0001-11 18/11/2020 Sampled by GHD TP18 1.0 - 2.0 Sandy CLAY; red/grey brown trace gravel				

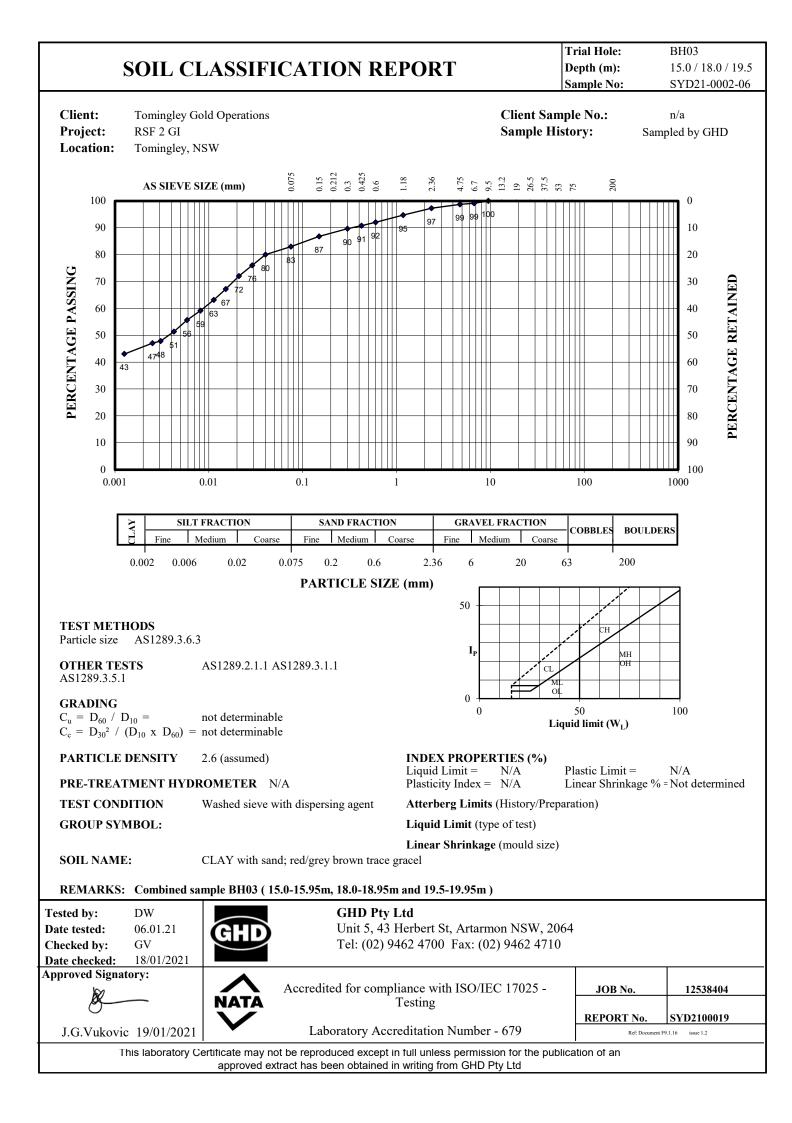
Test Results

Description	Method	Result	Limits
Date Tested		7/01/2021	
Coef of Permeability (m/s)	AS 1289.6.7.3	1 E-10	
Mean Stress Level (kPa)		30	
Permeant Used		Syd tap water	
Length (mm)		74.3	
Diameter (mm)		63.8	
Length/Diameter Ratio		1.16	
Laboratory Moisture Ratio (%)		101.0	
Laboratory Density Ratio (%)		98.0	
CompactiveEffort		Standard	
Method of Compaction		Remoulded	
Surcharge Applied (kg)		0.0	
Pressure Applied (kPa)		10	
Oversize Sieve (mm)		9.5	
Percentage Oversize (%)		0.0	
Moisture Content (%)		25.0	
Date Tested		8/01/2021	



Materi	al Te	st Report			Report No: SYD2100012 Issue No: 1
Client:	RSF 2 C Toming	ley NSW		NATA	Accredited for compliance with ISO / IEC 17025 - Testing
Project:	125384	U4		No: 679	on Approved Signatory: Jure G Vukovic Date of Issue: 19/01/2021 T SHALL NOT BE REPRODUCED EXCEPT IN FULL
Sample D GHD Sample Date Sampled By BH / TP No. Depth (m) Soil Descrip	e No ed	SYD21-0001-12 18/11/2020 Sampled by GHD TP19 1.0 - 2.0 sandy CLAY; grey/brown trace gravel		Particle Si Method: Drying by: Date Tested: Note: Sieve Size 13.2mm 9.5mm 6.7mm 4.75mm 2.36mm	ize Distribution AS 1289.3.6.1 Oven 5/01/2021 Sample Washed % Passing Limits 100 100 100 99 93
Other Tes Description Moisture Cor Date Tested Sample Histo Preparation Linear Shrink Mould Length Crumbling Curling Curling Cracking Liquid Limit (Method Plastic Limit Plasticity Ind	ntent (%) ory kage (%) h (mm) %) (%)	Method Result AS 1289.2.1.1 8.1 4/01/2021 4/01/2021 AS 1289.1.1 Oven-dried AS 1289.1.1 Dry Sieved AS 1289.3.4.1 N/A No No AS 1289.3.1.1 41 Four Point AS 1289.3.2.1 AS 1289.3.2.1 14 AS 1289.3.3.1 27	Limits	1.18mm 600μm 425μm 300μm 150μm 75μm	71 57 53 49 43 39
				Chart	
				% Pessing	







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	Result	: Limit
Method	Result	: Limit
Drying by:		
Note: Sample W Sieve Size 6.7mm 4.75mm 2.36mm 1.18mm 600μm 425μm 300μm 150μm 75μm	/ashed % Passing 100 99 96 94 93 92 89 86	Limits
	Drying by: 0 Date Tested: 4 Note: Sample W Sieve Size 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Drying by: Oven Date Tested: 5/01/2021 Note: Sample Washed Sieve Size % Passing 6.7mm 100 4.75mm 99 2.36mm 99 1.18mm 96 600µm 94 425µm 93 300µm 92 150µm 89



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ample	Details					Other	[.] Test R	esults		
ID Samp te Samp mpled B I / TP No pth (m) il Descri	led y	SYD21-(28/11/20 Samplec BH05 3.0 - 3.4 CLAY: with	020 I by GH 5 / 4.5	ID - 4.95	own/red/grey	<u>Descrip</u>	ition	Method	Result	<u>Limits</u>
rticle S	Size Dis	tributio	n					Method:	AS 1289.3.6.1	
% Pass	ing								Oven	
100 [••••				Note: Sample V	Vashed	
90 - · · · · 80 - · · · · 60 - · · · 50 - · · · 30 - · · · 20 - · · ·								Sieve Size 6.7mm 4.75mm 2.36mm 1.18mm 600μm 425μm 300μm 150μm 75μm	% Passing 100 98 92 86 83 81 76 71	Limits
	75µm	150µm	300µm 425µm	e00µm	1.18mm	2.36mm	4.75mm 6.7mm			



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ample	Details				Other [•]	Test R	esults		
HD Samp ate Samp ampled B H / TP No epth (m) oil Descri	bled By D.	SYD21-00 28/11/202 Sampled BH05 7.5, 10.5 a Sandy CLA	0 by GHD & 16.5	l, brown/red/gre	<u>Descripti</u>	on	<u>Metho</u>	d Resul	lt Limits
		tributior	•						
							Method: Drying by: Date Tested:	AS 1289.3.6.1 Oven 5/01/2021	
% Pass					· · · · · · · · · · · · · · · · · · ·		Drying by:	Oven 5/01/2021	
% Pass							Drying by: Date Tested: Note: Sample	Oven 5/01/2021 Washed	Limits
% Pass 100					· · · · · · · · · · · · · · · · · · ·		Drying by: Date Tested: Note: Sample Sieve Size 9.5mm	Oven 5/01/2021 Washed % Passing 100	Limits
% Pass 100 90					· · · · · · · · · · · · · · · · · · ·		Drying by: Date Tested: Note: Sample Sieve Size	Oven 5/01/2021 Washed % Passing 100 100 100	Limits
% Pass 100 - · · · · 90 - · · · · 80 - · · ·							Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm	Oven 5/01/2021 Washed % Passing 100 100 100 99	Limits
% Pass 100 90 80 70						· · · · · · · · · · · · · · · · · · ·	Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93	Limits
% Pass 100 90 80 60 -							Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90	Limits
% Pass 100 90 80 60 50 40							Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87 77	Limits
% Pass 100 90 60 50 40 50							Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87	Limits
% Pass 100 90 80 60 50 40 30 20							Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87 77	Limits
% Pass 100 90 60 50 40 30							Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87 77	Limits
% Pass 100 90 80 60 50 40 20 10 0	sing						Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87 77	Limits
% Pass 100 90 80 60 50 40 20 10 0			425µm 600µm	1.18mm 2.36mm	4.75mm 6.7mm	9.5mm	Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87 77	Limits
% Pass 100 90 80 60 50 40 20 10 0	sing			1.18mm 2.36mm		9.5mm	Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87 77	Limits
% Pass 100 90 80 60 50 40 20 10 0	sing		425µm 600µm	1.18mm 2.36mm		9.5mm	Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87 77	Limits
% Pass 100 90 80 60 50 40 20 10 0	sing		425µm 600µm	1.18mm 2.36mm		9.5mm	Drying by: Date Tested: Note: Sample Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	Oven 5/01/2021 Washed % Passing 100 100 100 99 97 93 90 87 77	Limits

Comments

Combined sample BH05 (7.50-7.95m, 10.5-10.95m and 16.5-16.95m)



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ample	Details					Other	Test R	esults		
HD Samp ate Samp ampled B H / TP No epth (m) oil Descri	led y	SYD21- 29/11/20 Sampled BH06 1.5 - 1.9 CLAY: with	020 d by Gl 05 / 3.0	HD	le grey/brown	<u>Descript</u>	ion	Metho	d Resul	<u>It Limits</u>
	Size Dis	tributic	on					Method: Drying by: Date Tested:	AS 1289.3.6.1 Oven 5/01/2021	
% Pass ¹⁰⁰ ⊺ · · ·	sing			••••		· · · · · · · · · · · · · · · · · · ·		Note: Sample		
90 - · · · · 80 - · · · 60 - · · · 50 - · · · 40 - · · · 20 - · · · 10 - · · ·								Sieve Size 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm 75µm	% Passing 100 100 97 93 88 86 83 76 70	Limits
0	75µm	150µm	300µm 425µm		1.18mm	2.36mm	4.75mm 6.7mm			



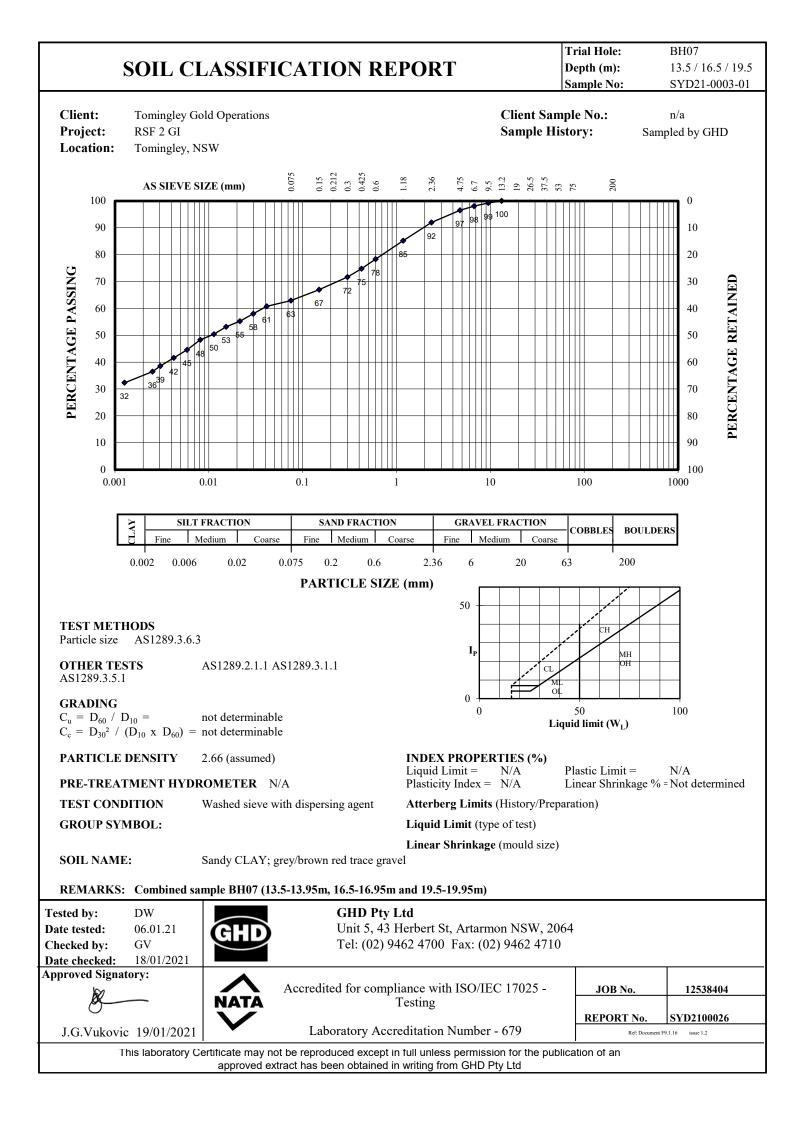
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lient:	Toming RSF 2 (-	peratio	ns Pty Ltd				Accredit Testing	ted for compliance with ISC	י / IEG 17025
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ample [Details				1	Other 1	est R	esults		
HD Sample ate Sample ampled By H / TP No. epth (m) oil Descrip	ed	SYD21-C 29/11/20 Sampled BH06 10.7, 12. CLAY: with	20 by GH 0 & 13.	D	-	Descriptio	on	Method	Result	Limits
% Passir 100 trees						+	<u> </u>			
90 + • • • •								Sieve Size	% Passing	Limits
80 - • • • •				••••	• • • • • • • • • • •	•••••	• • • • • • • •	6.7mm	100	Liiiits
70 - • • • •						•••••	••••••	4.75mm 2.36mm	100 99	
60 - • • • •			••••••	••••	• • • • • • • • • • • • •	•••••	• • • • • • • •	1.18mm 600µm	97 94	
50 + • • • •								425μm 300μm	92 90	
40 + • • • •							• • • • • • •	150µm	87	
30 - • • • •						•••••		75µm	83	
20+••••										
-										
10 - • • • •										
-		Ę	300µm 425µm	600µm	1.18mm	2.36mm	4.75mm 6.7mm			
0	75µm		25	0						
o <u>+</u>	75µm	150µm	30L	60		2.3	6.4			

Comments

Comined sample BH06 (10.7-11.15m, 12.0-12.45m and 13.5-13.95m)



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lateri	al Te	st Rep	ort					Report No: S	YD210002 Issue No:
lient:	Tomingle RSF 2 G	ey Gold Ope		y Ltd			Accre Testir	edited for compliance with I	SO / IEC 17025 -
Project:	1253840	-						of Issue: 18/01/2021	
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Sample I HD Sampl		SYD21-000	0 10		Othe Descri	r Test F	Kesults Metho	od Resu	ılt Limits
ate Sampl ampled By H / TP No. epth (m) oil Descrip	,	02/12/2020 Sampled by BH07 4.5 - 4.95 / CLAY: trace s	/ GHD 6.0 - 6.45	ottled brown/r	ed				
rticle S	ize Dist	ribution							
% Passi	ng						Method: Drying by: Date Tested:	AS 1289.3.6.1 Oven 5/01/2021	
				<u></u>					
100							Note: Sample	Washed	
100 - · · · · 90 - · · · · 80 - · · · 70 - · · · 60 - · · · 50 - · · ·			· · · · · · · · · · · · · · · · · · ·	· · · · · · · · · · · · · · · · · · ·			Note: Sample Sieve Size 2.36mm 1.18mm 600μm 425μm 300μm 150μm 75μm	Washed % Passing 100 100 99 99 98 96 91	Limits
90 - · · · 80 - · · · 70 - · · · 60 - · · ·							Sieve Size 2.36mm 1.18mm 600μm 425μm 300μm 150μm	% Passing 100 100 99 99 98 98	Limits
90 - · · · · 80 - · · · · 70 - · · · 60 - · · · 50 - · · · 40 - · · · 30 - · · ·							Sieve Size 2.36mm 1.18mm 600μm 425μm 300μm 150μm	% Passing 100 100 99 99 98 98	Limits





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ample [Details				Other T	est R	esults		
HD Sampl ate Sampl ampled By H / TP No. epth (m) bil Descrip	ed	SYD21-00 03/12/202 Sampled BH08 1.5 - 1.95 Clayey SAN	:0 by GHD / 3.0 - 3.4	:5 rel, brown/grey	<u>Descriptio</u>	n	Method	Result	Limits
% Passii		tributior							
					···/				
90 - • • • •							Sieve Size	% Passing	Limits
90 - · · · · + 80 - · · · ·							13.2mm	100	Limits
-			· · · · · · · · · · · · · · ·			•••••	13.2mm 9.5mm 6.7mm	100 99 96	Limits
80 - · · · · ·		· · · · · · · · · · · · · · · · · · ·				•••••	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm	100 99 96 93 81	Limits
80 - · · · · · 70 - · · · ·						 	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm	100 99 96 93 81 63	Limits
80 - · · · · · - 70 - · · · · · 60 - · · · ·						• • • • • • • • • • • • • • • • • • •	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm	100 99 96 93 81 63 51 45	Limits
80 - · · · · · 70 - · · · · · 60 - · · · · · 50 - · · · ·						· · · · · · · · · · · · · · · · · · ·	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	100 99 96 93 81 63 51 45 40 34	Limits
80 - · · · · · · · · · · · · · · · · · ·	· · · · · · · · · · · · · · · · · · ·					· · · · · · · · · · · · · · · · · · ·	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm	100 99 96 93 81 63 51 45 40	Limits
80 - · · · · · 60 - · · · · · · · · · · · · · · · · · ·						· · · · · · · · · · · · · · · · · · ·	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	100 99 96 93 81 63 51 45 40 34	Limits
80 - · · · · · 70 - · · · · 60 - · · · · 40 - · · · · 30 - · · · · 20 - · · · ·							13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	100 99 96 93 81 63 51 45 40 34	Limits
	150µm	300µm	425µm 600µm	L18mm	6.7mm	9.6mm +	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	100 99 96 93 81 63 51 45 40 34	Limits
	150µm	300µm	eoorm Boorm Sie	a 1.18mm 2.36mm	4.75mm 6.7mm	9.5mm +	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	100 99 96 93 81 63 51 45 40 34	Limits
	150µm	300µm			4.76mm 6.7mm	9.5mm	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	100 99 96 93 81 63 51 45 40 34	Limits
	150µm	300µm			4.76mm 6.7mm	9.5mm	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	100 99 96 93 81 63 51 45 40 34	Limits
	150µm	300µm			4.75mm 6.7mm	9.5mm + : : : : : : : : : : : : : : : : : :	13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm	100 99 96 93 81 63 51 45 40 34	Limits



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roject:	125384	04								NATA Accreditation App No: 679 Date THIS DOCUMENT SHALL	of Issue: 18/01/2021	
ample	Details					(Othe	r To	est R	esults		
HD Sampl ate Sampl ampled By H / TP No. epth (m) oil Descrij	ed /	SYD21 03/12/2 Sample BH08 7.5 - 7. Sandy C	2020 ed by G .95	iHD	red brown	_	<u>)escri</u>	ptio	<u>1</u>	Metho	od Resu	<u>It Limits</u>
rticle S	ize Dis	tributi	on							Martin a sh	A 0 4000 0 0 4	
% Passi	ing									Method: Drying by: Date Tested:	AS 1289.3.6.1 Oven 5/01/2021	
100			••••		·····			-	••••	Note: Sample	Washed	
90 - · · · · 80 - · · · 60 - · · · 50 - · · · 40 - · · ·								·····	· · · · · · · · · · · · · · · · · · ·	Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600μm 425μm 300μm 150μm 75μm	% Passing 100 100 99 97 95 91 89 86 73 62	Limits
20 - · · · · 10 - · · ·							_	6.7mm	9.5mm			



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Project:	125384	04			NATA Accreditation Appro No: 679 Date of THIS DOCUMENT SHALL	of Issue: 18/01/2021	
ample [Details			Other Test	Results		
HD Sample ate Sample ampled By H / TP No. epth (m) oil Descrip	ed	SYD21-000 04/12/2020 Sampled by BH09 1.5, 3.0, 4.5 Sandy CLAY:	GHD	<u>Description</u>	Method	d Result	Limits
rticle S	ize Dis	tribution					
% Passir	ng				Method: Drying by: Date Tested:	AS 1289.3.6.1 Oven 5/01/2021	
100					Note: Sample	Washed	
90 - · · · · 80 - · · · · 70 - · · · · 60 - · · · ·					Sieve Size 13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm	% Passing 100 99 98 97 90 77 67 61	Limits
40 - · · · · · · · · · · · · · · · · · ·		· · · · · · · · · · · · · · · · · · ·			300µm 150µm 75µm	56 48 44	
40 - · · · · · 30 - · · · ·		· · · · · · · · · · · · · · · · · · ·		· · · · · · · · · · · · · · · · · · ·	300µm 150µm	56 48	

Comments

Combined sample BH09 (1.5-1.95, 3.0-3.95, 4.5-4.95 and 6.0-6.45m)



al Te	st Rep	ort				Report No: SYI	D210003 Issue No:
RSF 2	GI	ations Pty Ltd			Accredite Testing	d for compliance with ISO	/ IEC 17025 -
-	-				No: 679 Date of	lssue: 18/01/2021	
Details			Other	Test Re	sults		
ed	04/12/2020 Sampled by BH09 15.0, 18.0 &	GHD 19.5		tion	Method	<u>Result</u>	Limits
ng					Drying by: C Date Tested: 5)ven /01/2021	
· · · · · · · · · · · · · · · · · · ·					Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600μm 425μm 300μm 150μm 75μm	% Passing 100 100 99 98 96 92 90 88 83 78	Limits
	Toming RSF 2 Toming 125384 Details e No ed	Tomingley Gold Oper RSF 2 GI Tomingley NSW 12538404 Details e No SYD21-0003 ed 04/12/2020 y Sampled by BH09 15.0, 18.0 & otion CLAY: with sand	Tomingley NSW 12538404 Details e No SYD21-0003-05 ed 04/12/2020 y Sampled by GHD BH09 15.0, 18.0 & 19.5 otion CLAY: with sand, trace gravel, brown/grey	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW 12538404 Details Other e No SYD21-0003-05 Descript ed 04/12/2020 y Sampled by GHD BH09 15.0, 18.0 & 19.5 otion CLAY: with sand, trace gravel, brown/grey/red Size Distribution	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW 12538404 Details Other Test Re Perform Office	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW 12538404 Details e No SYD21-0003-05 ed 04/12/2020 r Sampled by GHD BH09 15.0, 18.0 & 19.5 otion CLAY: with sand, trace gravel, brown/grey/red Method: A Drying by: C Date Tested: 5 Note: Sample W Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm	Tomingley Gold Operations Pty Ltd RSF 2 GI Tomingley NSW Image: Complexity of the second

Comments

Combined sample BH09 (15.0-15.95, 18.0-18.95 and 19.5-19.95m)



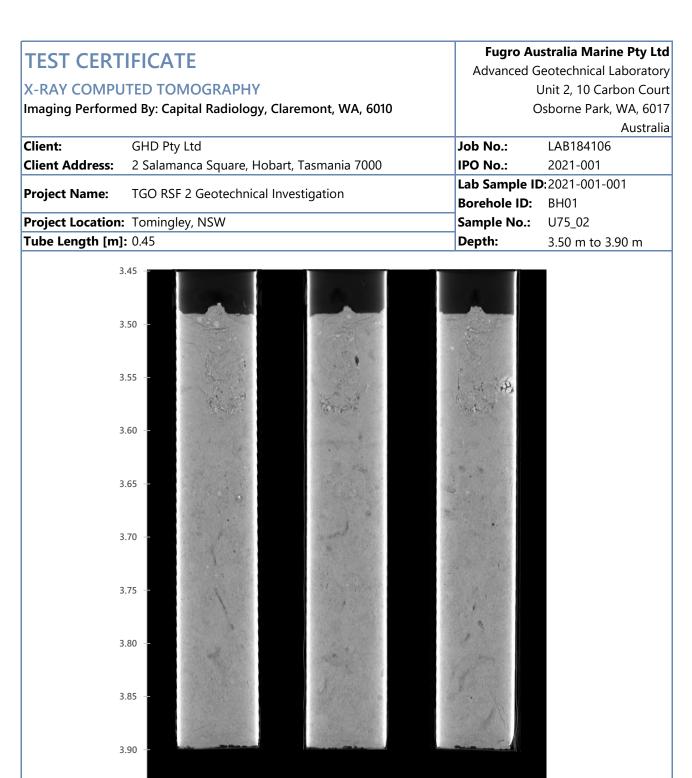
later	ial Te	est F	Repo	ort								Report No: SY	D210003 Issue No:
Client:	RSF 2		ld Operat	tions Pty	Ltd						Accredit Testing	ed for compliance with ISC) / IEC 17025 -
Project:	12538	404									NATA Accreditation Approv No: 679 Date of THIS DOCUMENT SHALL N	Issue: 18/01/2021	
Sample	Details	;					Otł	ner	Tes	st Re	esults		
HD Sam ate Sam ampled I H / TP N epth (m) oil Desc	pled By o.	04/12 Samp BH10 1.5, 3	8.0, 6.0 &	iHD 9.0	brown/red/g	-	Des	crip	<u>tion</u>		Method	Result	<u>Limits</u>
article	Size Dis	stribu	tion										
% Pa	ssing											AS 1289.3.6.1 Oven 5/01/2021	
¹⁰⁰ ↓.			•••••••••••••••••••••••••••••••••••••••	••••							Note: Sample V	/ashed	
90 - · · 80 - · 70 - · 60 - · 50 - · 30 - · 20 - · 10 - ·								· · · · · · · · · · · · · · · · · · ·	· · · · · · · · · · · · · · · · · · ·		Sieve Size 19.0mm 13.2mm 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600µm 425µm 300µm 150µm 75µm	% Passing 100 99 98 95 93 87 76 64 58 53 46 41	Limits
0	75µm 150µm	300µm	425µm 600µm	E E Sieve	2.36mm	4.75mm	6.7mm	9.5mm	13.2mm 19.0mm				



lateria	al Te	st Rep	ort						Issue No:
lient:	RSF 2	ley Gold Opera Gl ley NSW	ations Pty Ltd				NATA Accrec	lited for compliance with ISC) / IEC 17025 -
Project:	125384	04					NATA Accreditation Appro No: 679 Date of THIS DOCUMENT SHALL	of Issue: 18/01/2021	
ample [Details				Other	Test R	Results		
HD Sample ate Sample ampled By H / TP No. epth (m) oil Descrip	ed	SYD21-0003 04/12/2020 Sampled by BH10 13.5, 15.0 & CLAY: trace sand &	GHD	/pale grey/red	<u>Descrip</u>	<u>tion</u>	Methoo	d Result	Limits
% Passir 100 - · · · · ·	ng			<u></u>			Drying by: Date Tested: Note: Sample		
90 - · · · · 80 - · · · · 60 - · · · · 40 - · · · · 30 - · · · · 20 - · · · ·	150 muloc	300µm 425µm	600µm 1.18mm	2.36mm	4.76mm 6.7mm	9.5mm	Sieve Size 9.5mm 6.7mm 4.75mm 2.36mm 1.18mm 600μm 425μm 300μm 150μm 75μm	% Passing 100 100 99 98 96 95 93 91 88	Limits
0	4		-		4.0	57			

Comments

Combined sample BH10 (13.5-13.45, 15.0-15.45 and 18.0-18.45m)



Phone: +61 8 9218 2000

Comments:

Cert. No.:

3 95

0°

LAB184106_2021-001-001_XCT_XCT01(00)

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Imaged by: Capital Radiology Date Imaged: 13/01/2021

E-mail: FugroAGLab@fugro.com

120°

Reviewed by

Web: www.fugro.com

Date Reported:

Approved Signatory:

R. Calvert

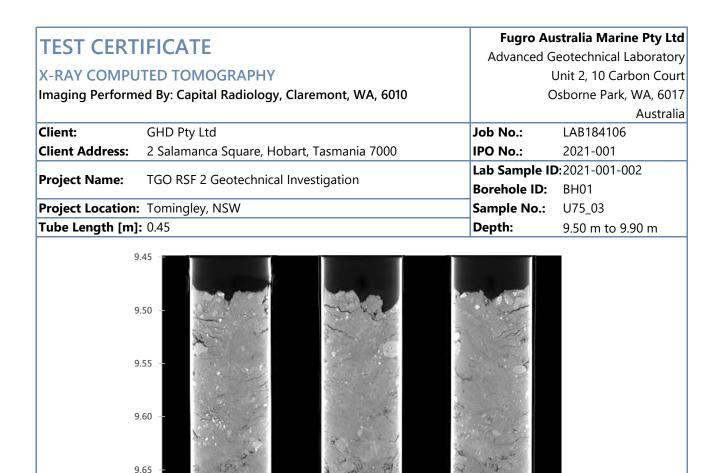
240°

RC



Figure A - 1

14/01/2021





120°

240°

FAM-18181 | Test Certificate - X-ray Computed Tomography | Rev 3BPage 2 of 5This document is uncontrolled once printed or downloaded and may not reflect the latest version.

9 70

9.75

9.80

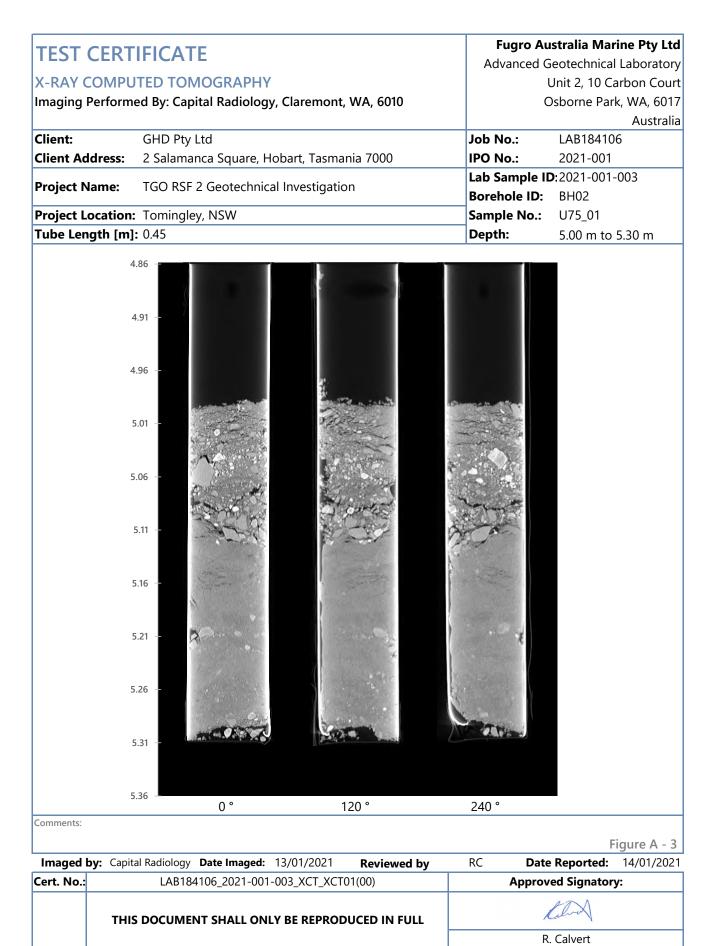
9.85

9.90

9 95

0°



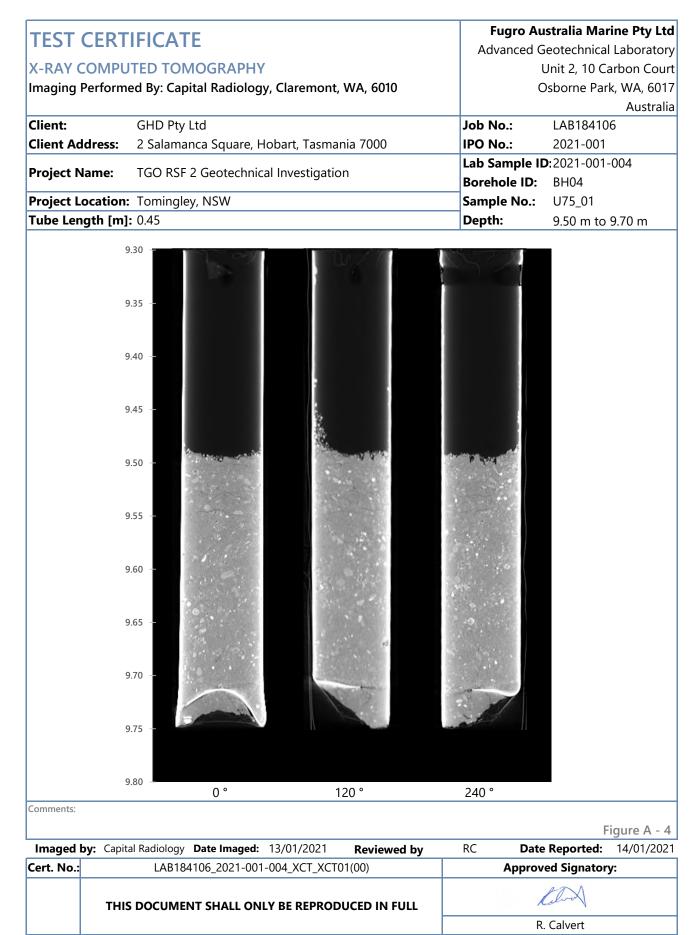


Phone: +61 8 9218 2000 **E-mail:** Fu

ail: <u>FugroAGLab@fugro.com</u>

Web: www.fugro.com



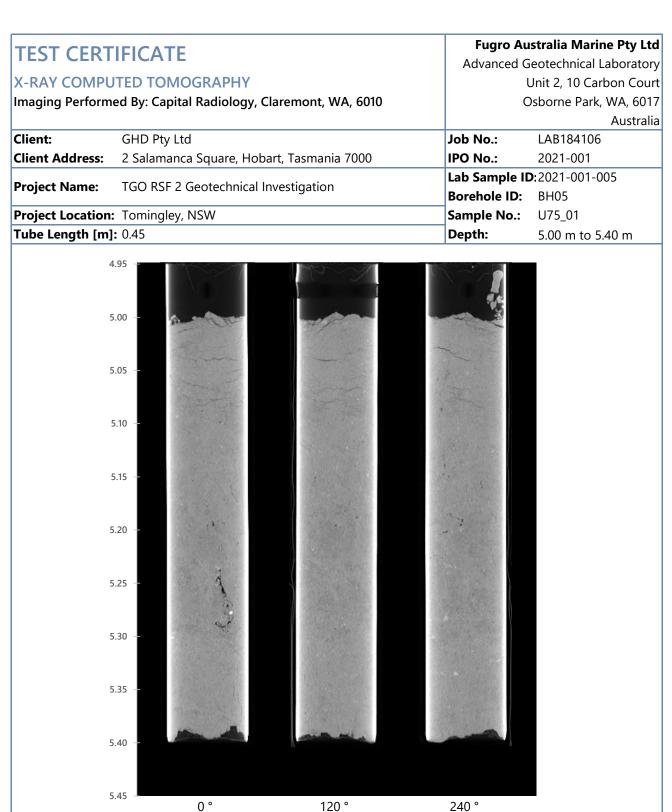


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E-mail: FugroAGLab@fugro.com

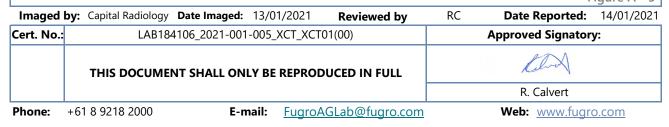
Web: www.fugro.com



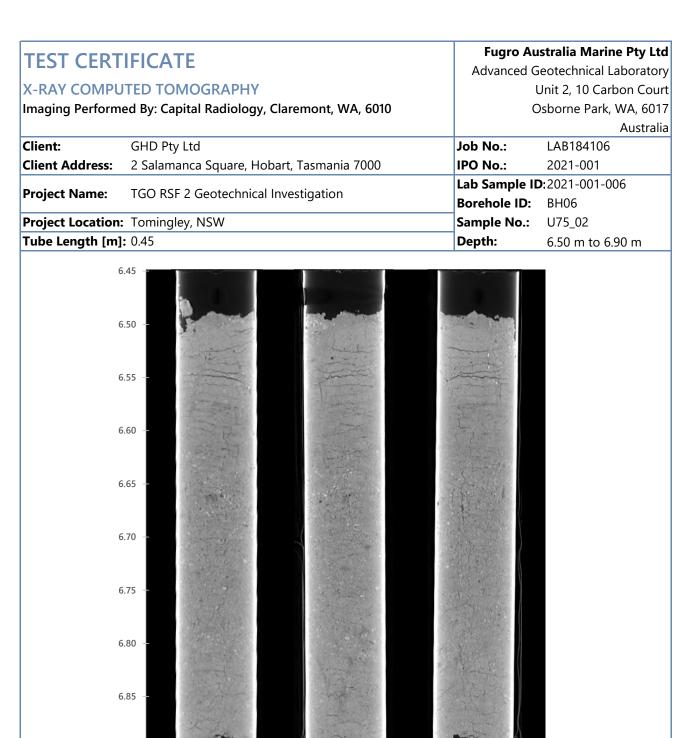


Comments:

Figure A - 5









120°

240°

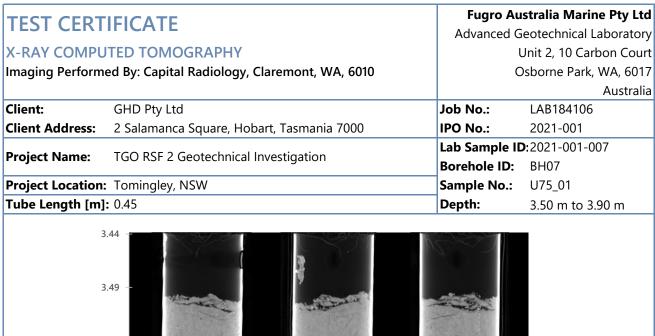
6.90

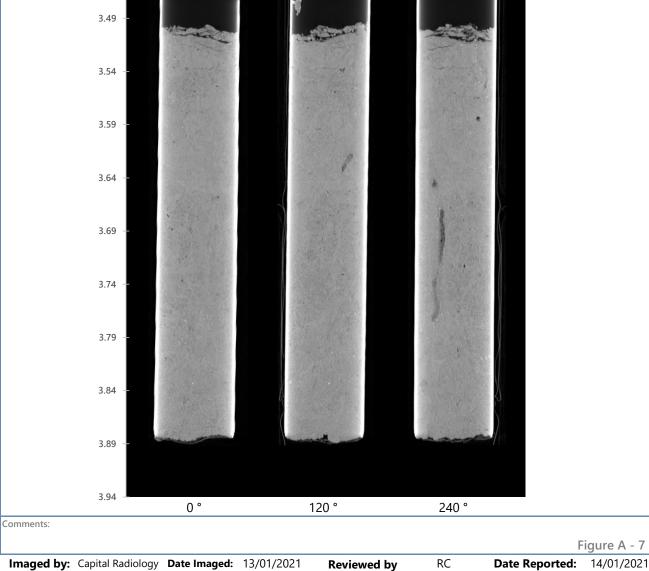
6.95

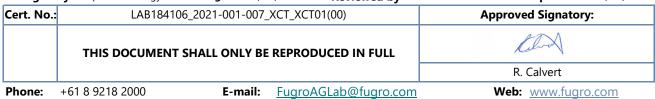
Comments:

0°



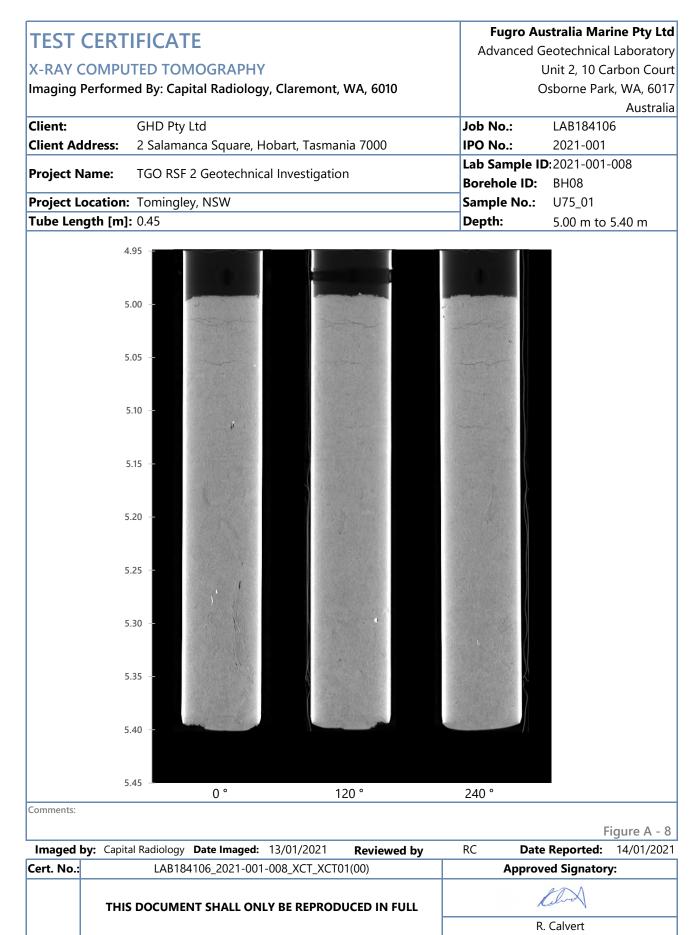






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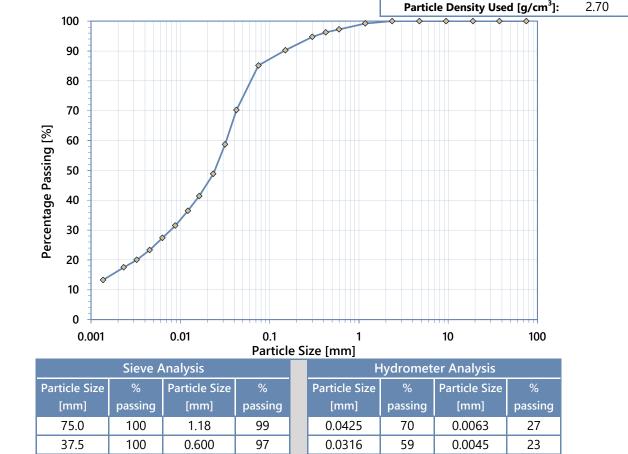
Phone: +61 8 9218 2000

E-mail: FugroAGLab@fugro.com

Web: www.fugro.com



TEST CERTIFICATE			Fugro A	Fugro Australia Marine Pty Ltd		
IESI CERI	IFICATE	Advanced (Advanced Geotechnical Laboratory			
PARTICLE SIZE	DISTRIBUT		Unit 2, 10 Carbon Court			
Test Method: AS	1289.3.6.1 & A	(Osborne Park, WA, 6017			
Sieving and Hydrometer Analysis				Australia		
Client:	GHD Pty Lto	1	Job No.:	LAB184106		
Client Address:	2 Salamanca	a Square, Hobart, Tasmania 7000	IPO No.:	2021-001		
Project Name:	TGO RSF 2 Geotechnical Investigation		Lab Sample I	Lab Sample ID:2021-001-011		
			Borehole ID:	BH01		
Project Location: Tomingley, NSW			Sample No.:	U75_02		
Method of Sample	Preparation:	As received post-triaxial testing	Depth:	3.70 m to 3.89 m		
Sample Description	: Brownish ye	llow/grey CLAY with sand	-			
			Dentiale Density II	2 - 31 = 270		



 2.36
 100
 0.075
 85
 0.0088

 Comments:
 Hydrometer used: g/L, Sample dispered by mechanical stirring

0.425

0.300

0.150

96

95

90

100

100

100

19.0

9.50

4.75

						Fi	igure B.1 - 1
Tested	by: SL	Date tested: 15	/02/2021	Reviewed by	RC	Date Reported:	10/03/2021
Cert. No.:	LAI	B184106_2021-001-0	11_PSD_CL0)3(00)	Approved Signatory:		
THIS DOCUMENT SHALL ONLY BE REPRODUCED IN FULL				China			
					R. Calvert		
Phone:	+61 8 9218 2000	E-mail	: FugroA	GLab@fugro.com		Web: www.fugro	o.com

49

41

36

32

0.0233

0.0163

0.0121

0.0032

0.0023

0.0014

20

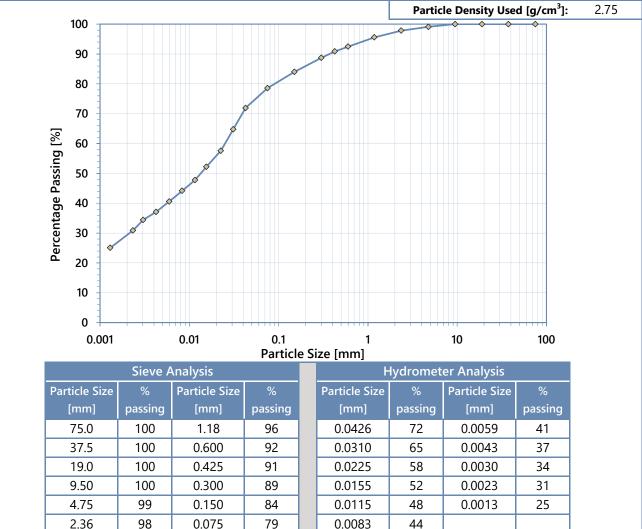
17

13





TEST CERTIFICATE			Fugro A	Fugro Australia Marine Pty Ltd		
				Advanced Geotechnical Laboratory		
PARTICLE SIZE DISTRIBUTION				Unit 2, 10 Carbon Court		
Test Method: AS1289.3.6.1 & AS1289.3.6.3				Osborne Park, WA, 6017		
Sieving and Hydrometer Analysis				Australia		
Client:	GHD Pty Ltd		Job No.:	LAB184106		
Client Address:	2 Salamanca	Square, Hobart, Tasmania 7000	IPO No.:	2021-001		
Dreigst Norma	TGO RSF 2 Geotechnical Investigation		Lab Sample ID:2021-001-015			
Project Name:	IGO KSF 2 G	eolechnical investigation	Borehole ID	: BH01		
Project Location: Tomingley, NSW			Sample No.:	U75_03		
Method of Sample I	Preparation:	As received post-triaxial testing	Depth:	9.68 m to 9.84 m		
Sample Description	Light browni	sh grey CLAY with sand				

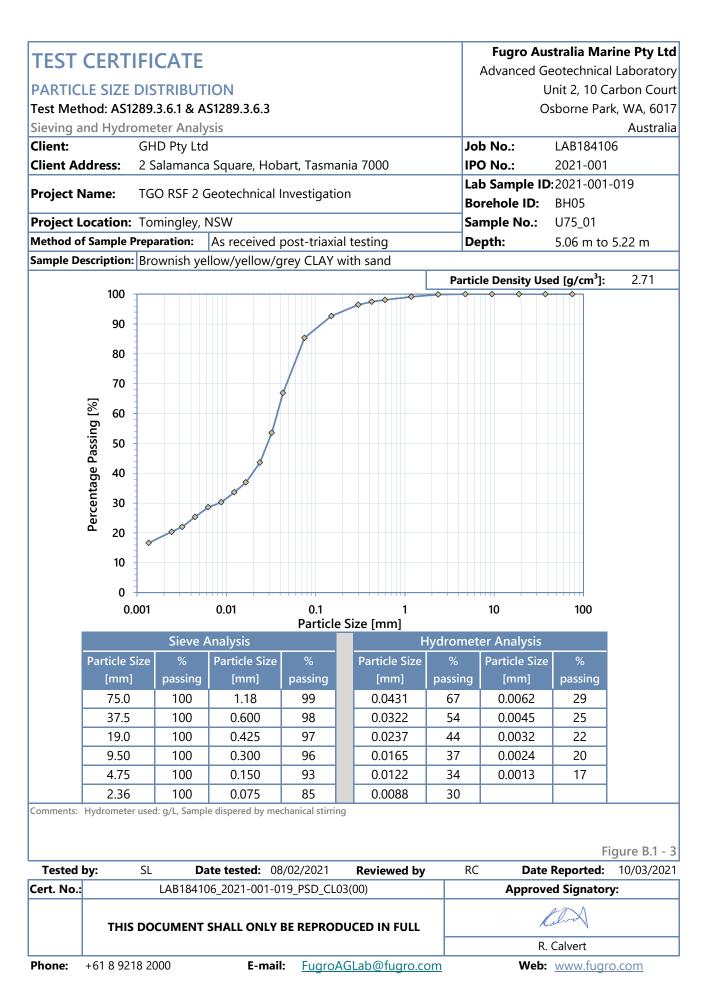


Comments: Hydrometer used: g/L, Sample dispered by mechanical stirring

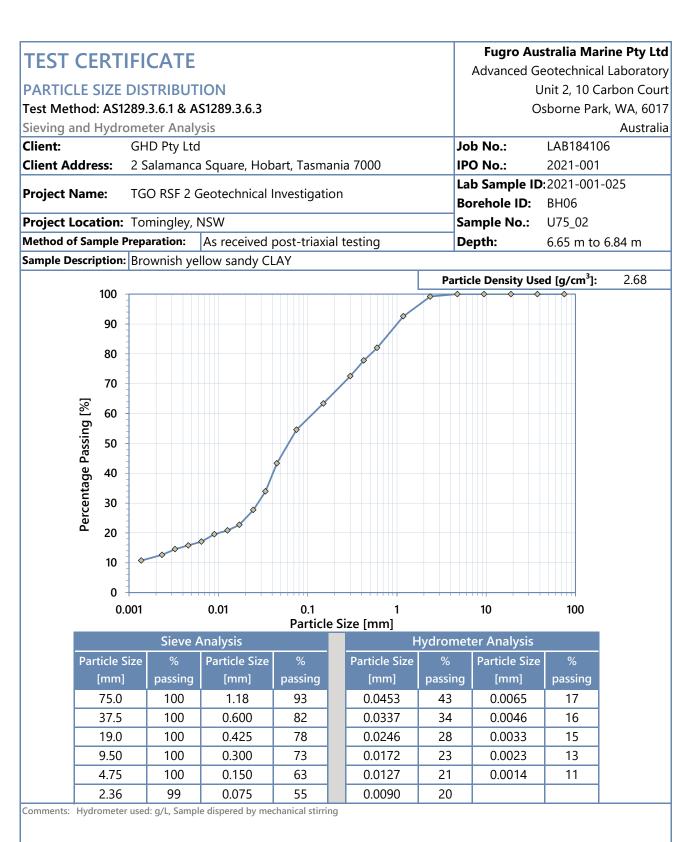
Figure B.1 - 2 **Tested by:** SL/ARC RC **Date Reported:** 10/03/2021 Date tested: 08/02/2021 **Reviewed by** Cert. No.: LAB184106_2021-001-015_PSD_CL03(00) **Approved Signatory:** THIS DOCUMENT SHALL ONLY BE REPRODUCED IN FULL R. Calvert Phone: +61 8 9218 2000 E-mail: FugroAGLab@fugro.com Web: www.fugro.com

-Jugro

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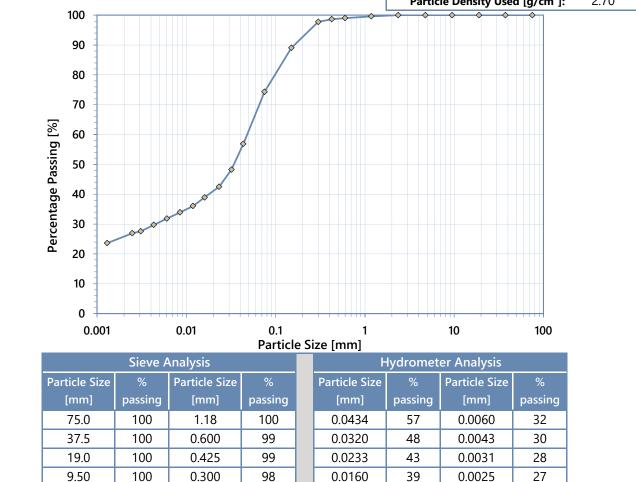
						Fi	gure B.1 - 4
Tested	by: SL	Date tested:	15/02/2021	Reviewed by	RC	Date Reported:	10/03/2021
Cert. No.:	LA	B184106_2021-001	-025_PSD_CL0)3(00)		Approved Signator	y:
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						R. Calvert	
Phone:	+61 8 9218 2000	E-m	ail: <u>Fugro</u> A	GLab@fugro.com		Web: www.fugr	o.com



FAM-18202 | Particle Size Distribution - Sieving Hydro - Test Certificate | Rev 6B

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TEST CERT		Fugro Australia Marine Pty Ltd
IESI CERI	IFICATE	Advanced Geotechnical Laboratory
PARTICLE SIZE	DISTRIBUTION	Unit 2, 10 Carbon Court
Test Method: AS	1289.3.6.1 & AS1289.3.6.3	Osborne Park, WA, 6017
Sieving and Hyd	rometer Analysis	Australia
Client:	GHD Pty Ltd	Job No.: LAB184106
Client Address:	2 Salamanca Square, Hobart, Tasmania 7000	IPO No.: 2021-001
	TCO DCC 2 Control Investigation	Lab Sample ID:2021-001-029
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID: BH07
Project Location	: Tomingley, NSW	Sample No.: U75_01
Method of Sample	Preparation: As received post-triaxial testing	Depth: 3.52 m to 3.71 m
Sample Description	n: Brownish yellow/grey CLAY with sand	
		Particle Density Used [g/cm ³]: 2.70



100 Comments: Hydrometer used: g/L, Sample dispered by mechanical stirring

100

4.75

2.36

Figure B.1 - 5 **Tested by:** SL/ARC RC **Date Reported:** 10/03/2021 Date tested: 22/02/2021 **Reviewed by** Cert. No.: LAB184106_2021-001-029_PSD_CL03(00) **Approved Signatory:** THIS DOCUMENT SHALL ONLY BE REPRODUCED IN FULL R. Calvert Phone: +61 8 9218 2000 E-mail: FugroAGLab@fugro.com Web: www.fugro.com

0.0119

0.0085

36

34

0.0013

24



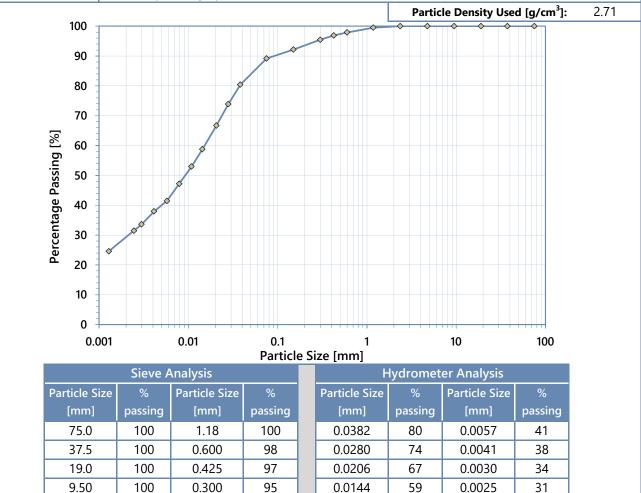
0.150

0.075

89

74

TEST CERT	ΙΕΙΟΛΤΕ		Fugro Au	Fugro Australia Marine Pty Ltd		
	IIICAIL	Advanced G	Advanced Geotechnical Laboratory			
PARTICLE SIZE	DISTRIBUT	ON		Unit 2, 10 Carbon Court		
Test Method: AS1	289.3.6.1 & A	51289.3.6.3	(Osborne Park, WA, 6017		
Sieving and Hydr	ometer Analy	sis		Australia		
Client:	GHD Pty Ltd		Job No.:	LAB184106		
Client Address:	2 Salamanca	Square, Hobart, Tasmania 7000	IPO No.:	2021-001		
Dreigst Norma		actochnical Investigation	Lab Sample II	D :2021-001-035		
Project Name:	IGU KSF 2 G	eotechnical Investigation	Borehole ID:	BH08		
Project Location:	Tomingley, N	ISW	Sample No.:	U75_01		
Method of Sample Preparation: As received post-triaxial testing		As received post-triaxial testing	Depth:	5.16 m to 5.35 m		
Sample Description						



100 Comments: Hydrometer used: g/L, Sample dispered by mechanical stirring

100

0.150

0.075

92

89

4.75

2.36

Figure B.1 - 6 **Tested by:** SL/ARC RC **Date Reported:** 10/03/2021 Date tested: 19/02/2021 **Reviewed by** Cert. No.: LAB184106_2021-001-035_PSD_CL03(00) **Approved Signatory:** THIS DOCUMENT SHALL ONLY BE REPRODUCED IN FULL R. Calvert Phone: +61 8 9218 2000 E-mail: FugroAGLab@fugro.com Web: www.fugro.com

0.0108

0.0079

53

47

0.0013

25



FAM-18202 | Particle Size Distribution - Sieving Hydro - Test Certificate | Rev 6B

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TEST CERT	TEICATE	Fugro A	Fugro Australia Marine Pty Ltd			
	IFICATE	Advanced	Advanced Geotechnical Laboratory			
ATTERBERG LI	MITS TEST		Unit 2, 10 Carbon Court			
Test Method: AS1	289.3.9.1, AS1289.3.2.1, AS1289.3.3.2 & AS1289.3.4.1		Osborne Park, WA, 6017			
Cone Penetrome	ter Method		Australia			
Client:	GHD Pty Ltd	Job No.:	LAB184106			
Client Address:	2 Salamanca Square, Hobart, Tasmania 7000	IPO No.:	2021-001			
Project Name:	TGO RSF 2 Geotechnical Investigation	Lab Sample I	D:Refer to Table			
Project Name:	IGO RSF 2 Geolecifical investigation	Borehole ID:	Refer to Table			
Project Location:	: Tomingley, NSW	Sample No.: Refer to Table				
		Depth:	Refer to Table			

		_		-	Lind		
Cert. No.:			01-012 to 030_ATT_CL	Reviewed by	Approved Si		
Sa Tested by			o contain air bubbles (gas) (ted: Refer to Table		RC Date Rep	Figure B.2 - 1	
			istory of sample : Oven drie				
	.P. = Non-p	l plastic, N.O. = Not Obtain	able, N.D. = Not Determine	d			
Type of Wat Used	ter	Potable	Potable	Potable	Potable	Potable	
Length of N [mm]	lould	250	250	250	250	250	
Mode		Curling and Cracking	Curling and Cracking	Flat with Cracking	Curling	Curling	
Linear Shrin <i>LS</i>	ıkage,	9.0	10.0	6.0	6.0	10.0	
Cone Plastic Index, I _{CP} [9		29	48	19	13	27	
Plastic Limit [%]	t, W _P	19	6	15	15	17	
Cone Liquid w _{CL} [%]	l Limit,	48	54	34	28	44	
Date Tested		24/02/2021	22/02/2021	23/02/2021	23/02/2021	25/02/2021	
Sample Des	cription	Brownish yellow/grey CLAY with sand	Light brownish grey CLAY with sand	Brownish yellow/yellow/grey CLAY with sand	Brownish yellow sandy CLAY	Brownish yellow/grey CLAY with sand	
Depth To [n	n]	3.89	9.87	5.22	6.84	3.71	
Depth From	ı [m]	3.70	9.68	5.03	6.65	3.52	
Sample No.		U75_02	U75_03	U75_01	U75_02	U75_01	
Borehole ID		BH01	BH01	BH05	BH06	BH07	
Lab Sample	ID	2021-001-012	2021-001-016	2021-001-020	2021-001-026	2021-001-030	

Phone: +61 8 9218 2000

8 2000

E-mail: <u>FugroAGLab@fugro.com</u>

Web: <u>www.fugro.com</u>

R. Calvert

TEST CERT	IEICATE	Fugro A	ustralia Marine Pty Ltd
	IFICATE	Advanced (Geotechnical Laboratory
ATTERBERG LI	MITS TEST		Unit 2, 10 Carbon Court
Test Method: AS1	289.3.9.1, AS1289.3.2.1, AS1289.3.3.2 & AS1289.3.4.1	(Osborne Park, WA, 6017
Cone Penetrome	ter Method		Australia
Client:	GHD Pty Ltd	Job No.:	LAB184106
Client Address:	2 Salamanca Square, Hobart, Tasmania 7000	IPO No.:	2021-001
Project Name:	TGO RSF 2 Geotechnical Investigation	Lab Sample I	D :Refer to Table
Project Name.	160 KSF 2 Geolechnical investigation	Borehole ID: Refe	
Project Location:	: Tomingley, NSW	Sample No.:	Refer to Table
		Depth:	Refer to Table

Lab Sample ID	2021-001-036							
Borehole ID	BH08							
Sample No.	U75_01							
Depth From [m]	5.16							
Depth To [m]	5.35							
Sample Description	Brownish yellow/grey CLAY trace sand							
Date Tested	25/02/2021							
Cone Liquid Limit, w _{CL} [%]	43							
Plastic Limit, <i>w</i> _P [%]	17							
Cone Plasticity Index, <i>I</i> _{CP} [%]	26							
Linear Shrinkage, <i>LS</i>	11.5							
Mode	Curling and Cracking							
Length of Mould [mm]	250							
Type of Water Used	Potable							
Comments: N.P. = Non-p	olastic, N.O. = Not Obtaina	ble, N.D. = Not Determine	d					
	Method of preparation : Dry sieved, History of sample : Oven dried at 50°c Figure B.2 - 2							
					5			

Tested I	oy: SL	Date tested: Refer	to Table	Reviewed by	RC	Date Reported:	10/03/2021		
Cert. No.:	LAB	3184106_2021-001-036	_ATT_CL04	ł(00)		Approved Signatory:			
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						R. Calvert			
Phone:	+61 8 9218 2000	E-mail:	FugroA(GLab@fugro.cor	<u>n</u>	Web: <u>www.fugro.com</u>			



TEST CERT	IFICATE	-	ustralia Marine Pty Ltd	
		Advanced Geotechnical Laboratory		
SOIL PARTICLE	DENSITY OF FRACTION PASSING THE 2.36 mm		Unit 2, 10 Carbon Court	
SIEVE			Osborne Park, WA, 6017	
Test Method: AS1	289.3.5.1		Australia	
Client:	GHD Pty Ltd	Job No.:	LAB184106	
Client Address:	2 Salamanca Square, Hobart, Tasmania 7000	IPO No.:	2021-001	
Project Name:	TGO RSF 2 Geotechnical Investigation	Lab Sample	ID:Refer to Table	
Project Name:	IGO KSF 2 Geolecinical investigation	Borehole ID:	Refer to Table	
Project Location:	Tomingley, NSW	Sample No.: Refer to Table		
		Depth:	Refer to Table	

Borehole ID	Sample No. Lab Sample ID	Date Tested	Dept	h [m]	Soil Particle Density, $\rho_{\rm f}$	Temperatur e of Test	
				From	То	[g/cm ³]	[°C]
BH01	U75_02	2021-001-009	09/02/2021	3.51	3.70	2.68	24.1
BH01	U75_02	2021-001-013	18/02/2021	3.70	3.89	2.70	24.1
BH01	U75_03	2021-001-017	09/02/2021	9.68	9.87	2.75	24.1
BH05	U75_01	2021-001-021	09/02/2021	5.03	5.22	2.71	24.1
BH06	U75_02	2021-001-027	18/02/2021	6.65	6.84	2.68	24.1
BH07	U75_01	2021-001-031	22/02/2021	3.52	3.71	2.70	24.1
BH08	U75_01	2021-001-037	18/02/2021	5.16	5.35	2.71	24.1
Comments: Samples 2021-00 observed to proc	1 1-031 and 2021-001-037 v luce bubbles (gas) whilst ir			e bottle follow	ving the addi	I tion of water and	were also

						Fi	gure B.3 - 1	
Tested	by: SL/ARC	Date tested: Refer	to Table	Reviewed by	RC	Date Reported:	10/03/2021	
Cert. No.:	LAB184	4106_2021-001-009 to	037_SG_C	L06(00)		Approved Signatory:		
	THIS DOCUM	IENT SHALL ONLY BE	REPROD	JCED IN FULL		that		
						R. Calvert		
Phone:	+61 8 9218 2000	E-mail:	FugroA(GLab@fugro.con	<u>n</u>	Web: www.fugro	o.com	

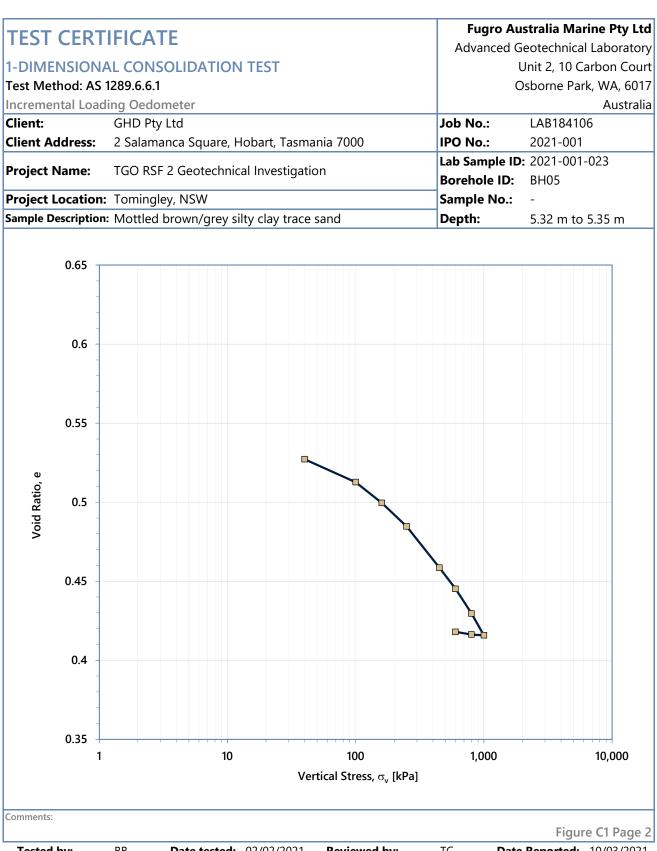


TEST CERT		ТЕ			Fugro Australia Marine Pty Ltd		
IESI CERI	IFICF				Advanced	Geotechnica	al Laboratory
1-DIMENSION	AL COM	SOLIDATION TEST				Unit 2, 10 0	Carbon Court
Test Method: AS	1289.6.6	5.1				Osborne Pa	rk, WA, 6017
Incremental Load	ling Oec	lometer					Australia
Client:	GHD P	'ty Ltd			Job No.:	LAB1841	06
Client Address:	2 Salar	manca Square, Hobart, Tasma	ania 70	000	IPO No.:	2021-001	
		CE 2 Cootoshnisel Investigati	a b		Lab Sample I	D: 2021-001	-023
Project Name:	IGO R	SF 2 Geotechnical Investigation	on		Borehole ID:	BH05	
Project Location:	: Tomin	gley, NSW			Sample No.:	-	
Sample Description	: Mottle	d brown/grey silty clay trace	sand		Depth:	5.32 m to	o 5.35 m
Test Details:				Sample Details:		Initial	Final
Test ID:		TGO-OED01		Sample Diamete	er [mm]:	60.00	60.00
Soil Particle Density	y [t/m³]:	2.71		Sample Height [mm]:	22.14	21.54
· · · · · · · · · · · · · · · · · · ·			'	Dry Density [t/m ³]: 1.73		1.79	
				Moisture Conte	nt [%]:	17.3 *	18.8
				Note: * Moisture con	tent calculated usi	ng trimmings; n	nay not be equal

Stage		σ_v	e	c v	m _v	k
Stage		kPa	e	m²/year	m²/kN	m/sec
	1	40	0.527	-	4.9E-04	-
	2	100	0.513	34.0	1.6E-04	1.7E-09
	3	160	0.500	28.2	1.4E-04	1.3E-09
Looding	4	250	0.485	27.7	1.1E-04	9.5E-10
Loading	5	450	0.459	24.7	8.8E-05	6.8E-10
	6	600	0.445	20.5	6.1E-05	3.9E-10
	7	800	0.430	10.2	5.4E-05	1.7E-10
	8	1000	0.416	8.1	4.8E-05	1.2E-10
	9	800	0.416	-	-	-
Unloading	10	600	0.418	-	-	-

Comments:							
						Figu	re C1 Page 1
Tested	by: BB	Date tested: 02/	/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021
Cert. No.:	LAB18	4106_2021-001-023_	_CONS_OED	01(00)	Approved Signatory:		
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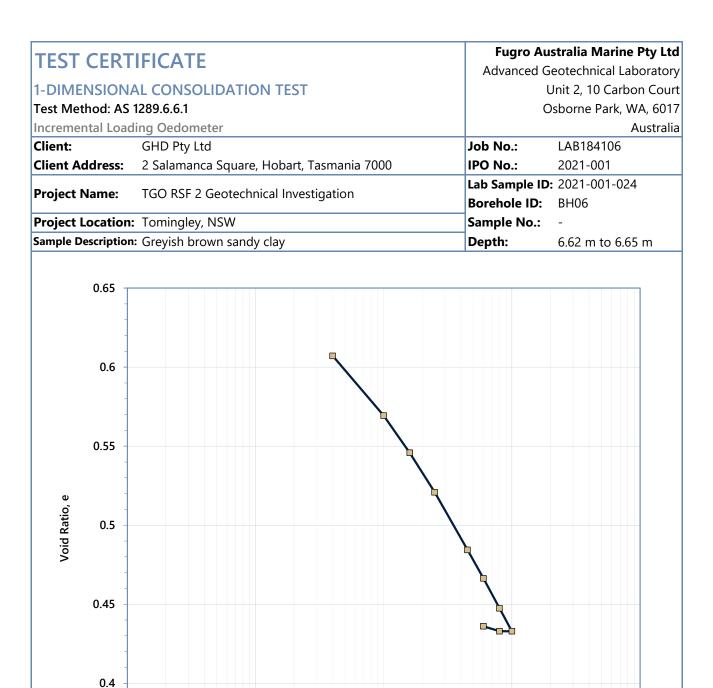


TEST CERT		ТЕ		Fugro A	ustralia Ma	rine Pty Ltd	
IESI CERI	IFICF	ATE		Advanced Geotechnical Laboratory			
1-DIMENSION/	AL CON	ISOLIDATION TEST		Unit 2, 10 Carbon Court			
Test Method: AS	1289.6.6	.1			Osborne Pa	rk, WA, 6017	
Incremental Load	ncremental Loading Oedometer					Australia	
Client:	GHD P	ty Ltd		Job No.:	LAB1841	06	
Client Address:	2 Salar	manca Square, Hobart, Tasma	nia 7000	IPO No.:	2021-001		
Dreigst Norma	reject Names TCO BSE 2 Control Investigation					-024	
Project Name:	TGO RSF 2 Geotechnical Investigation				BH06		
Project Location:	: Tomin	gley, NSW		Sample No.:	-		
Sample Description	: Greyisl	h brown sandy clay		Depth:	6.62 m to	o 6.65 m	
Test Details:			Sample Details:		Initial	Final	
Test ID:		TGO-OED02	Sample Diamete	er [mm]:	60.00	60.00	
Soil Particle Density	y [t/m³]:	2.68	Sample Height	[mm]:	22.25	20.11	
			Dry Density [t/r	Dry Density [t/m ³]: 1.60 1			
			Moisture Content [%]: 14.9 * 16.6				
			Note: * Moisture cor	tent calculated usi	ng trimmings; n	nay not be equal	

Stores		σ_{v}		C _v	m _v	k
Stage		kPa	e	m²/year	m²/kN	m/sec
	1	40	0.607	-	6.6E-04	-
	2	100	0.569	118.2	3.9E-04	1.4E-08
	3	160	0.546	93.0	2.5E-04	7.2E-09
Loading	4	250	0.521	95.5	1.8E-04	5.3E-09
Loading	5	450	0.484	56.6	1.2E-04	2.1E-09
	6	600	0.466	56.2	8.1E-05	1.4E-09
	7	800	0.447	63.2	6.4E-05	1.3E-09
	8	1000	0.433	60.3	5.0E-05	9.4E-10
Unleading	9	800	0.433	-	-	-
Unloading	10	600	0.436	-	-	-

Comments:							
						Figur	e C2 Page 1
Tested	by: BB	Date tested: 02	/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021
Cert. No.:	LAB18	84106_2021-001-024		Approved Signatory:			
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						T. Chang	
Phone:	+61 8 9218 2000	E-mail:	: FugroA	GLab@fugro.co	<u>m</u>	Web: www.fugro	o.com







100

Vertical Stress, o_v [kPa]



0.35

1

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10



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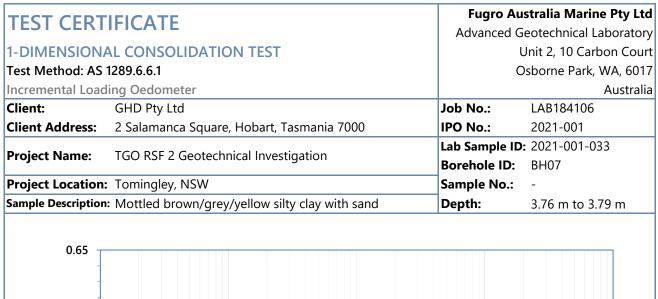
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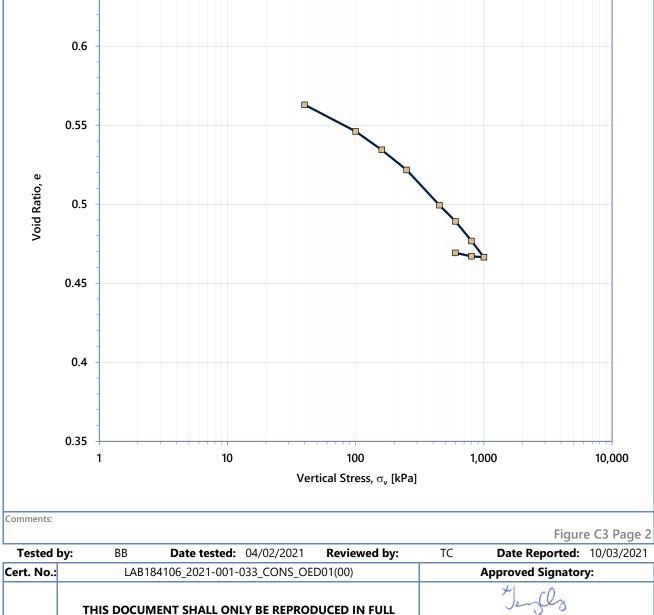
TEST CERT		\TE		Fugro A	ustralia Ma	rine Pty Ltd		
IESI CERI	IFICA	AIC		Advanced	Advanced Geotechnical Laboratory			
1-DIMENSION		NSOLIDATION TEST			Unit 2, 10 Carbon Court			
Test Method: AS	1289.6.6	5.1			Osborne Pa	rk, WA, 6017		
Incremental Load	ding Oed	dometer			Australia			
Client:	GHD F	Pty Ltd		Job No.:	LAB1841	06		
Client Address:	2 Sala	manca Square, Hobart, Tasma	ania 7000	IPO No.:	2021-007	1		
Dreiget Name	ISE 2 Contachnical Investigati	Lab Sample I	D: 2021-00 ⁻	1-033				
Project Name:	IGO R	SF 2 Geotechnical Investigati	011	Borehole ID:	: BH07			
Project Location	: Tomin	gley, NSW		Sample No.:	-			
Sample Description	n: Mottle	ed brown/grey/yellow silty cla	y with sand	Depth:	3.76 m to	o 3.79 m		
Test Details:			Sample Details	:	Initial	Final		
Test ID:		TGO-OED03	Sample Diame	ter [mm]:	60.00	60.00		
Soil Particle Densit	y [t/m³]:	2.70	Sample Height	[mm]:	22.29	21.72		
			Dry Density [t/	′m³]:	1.71	1.74		
			Moisture Content [%]: 16.0 * 18.6					
			Note: * Moisture co	Note: * Moisture content calculated using trimmings; may not be equal				

Stage		σ_v		c v	m _v	k
Stage		kPa	е	m²/year	m²/kN	m/sec
	1	40	0.563	-	5.2E-04	-
	2	100	0.546	239.8	1.8E-04	1.3E-08
	3	160	0.534	154.6	1.3E-04	6.1E-09
Looding	4	250	0.522	126.8	9.2E-05	3.6E-09
Loading	5	450	0.499	114.5	7.3E-05	2.6E-09
	6	600	0.489	45.3	4.5E-05	6.4E-10
	7	800	0.477	23.3	4.2E-05	3.0E-10
	8	1000	0.466	12.5	3.5E-05	1.3E-10
	9	800	0.467	-	-	-
Unloading	10	600	0.469	-	-	-

Comments:							
						Figur	e C3 Page 1
Tested	by: BB	Date tested: 04/0	2/2021 Review	ed by:	TC	Date Reported:	10/03/2021
Cert. No.:	LAB18	4106_2021-001-033_C	CONS_OED01(00)		Approved Signatory:		
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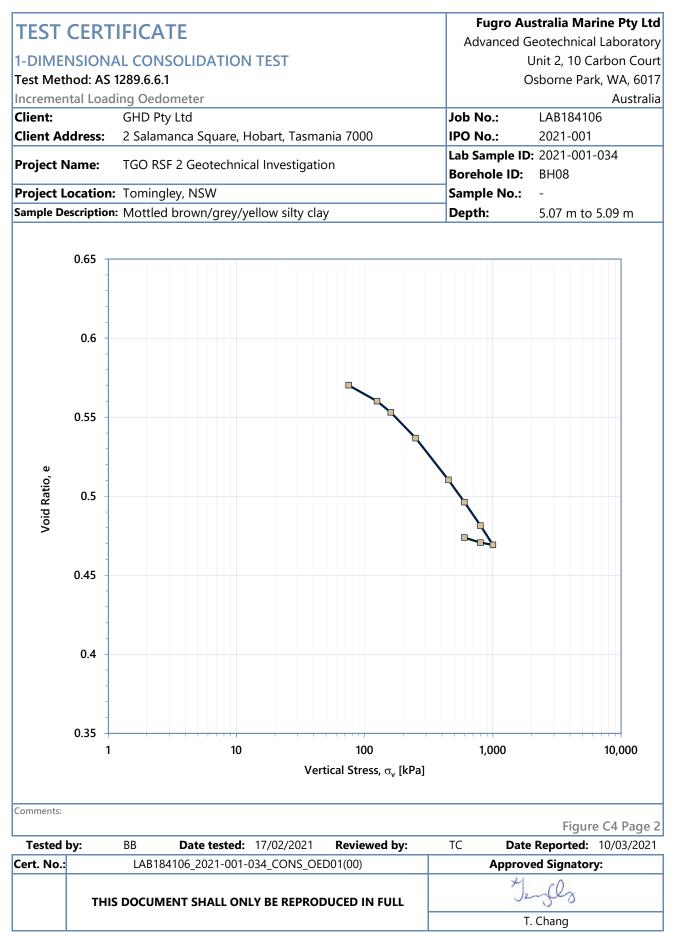
TECT CEDT		тг		Fugro A	ustralia Ma	rine Pty Ltd		
TEST CERT	IFICA	ATE		Advanced Geotechnical Laboratory				
1-DIMENSION	AL COM	ISOLIDATION TEST			Unit 2, 10 Carbon Court			
Test Method: AS	1289.6.6	5.1			Osborne Pa	rk, WA, 6017		
Incremental Load			Australia					
Client:	GHD P	'ty Ltd		Job No.:	LAB1841	06		
Client Address:	2 Salar	manca Square, Hobart, Tasma	nia 7000	IPO No.:	2021-001			
Project Name	act Name: TCO BSE 2 Contacting Investigation				D: 2021-001	-034		
Project Name:	Name: TGO RSF 2 Geotechnical Investigation				BH08			
Project Location:	: Tomin	gley, NSW		Sample No.:	-			
Sample Description	: Mottle	d brown/grey/yellow silty clay	/	Depth:	5.07 m to	o 5.09 m		
Test Details:			Sample Details:		Initial	Final		
Test ID:		TGO-OED04	Sample Diamete	er [mm]:	70.00	70.00		
Soil Particle Density	y [t/m³]:	2.71	Sample Height	[mm]:	20.18	19.51		
			Dry Density [t/n	Dry Density [t/m ³]: 1.66 1				
			Moisture Content [%]: 21.0 * 22.6					
			Note: * Moisture cor	ntent calculated usi	ng trimmings; n	nay not be equal		

Stage		σ_v	e	c v	m _v	k
Stage		kPa	e	m²/year	m²/kN	m/sec
	1	75	0.570	-	1.5E-04	-
	2	125	0.560	46.3	1.3E-04	1.8E-09
	3	160	0.553	54.3	1.3E-04	2.2E-09
Leeding	4	250	0.537	32.8	1.2E-04	1.2E-09
Loading	5	450	0.510	35.6	8.6E-05	9.5E-10
	6	600	0.496	28.6	6.2E-05	5.5E-10
	7	800	0.481	18.9	5.0E-05	2.9E-10
	8	1000	0.469	15.9	4.1E-05	2.0E-10
l la la adia a	9	800	0.471	-	-	-
Unloading	10	600	0.474	-	-	-

Comments:								
						Figur	e C4 Page 1	
Tested	by: BB	Date tested: 17,	/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021	
Cert. No.:	LAB18	4106_2021-001-034 __	_CONS_OE	D01(00)		Approved Signatory:		
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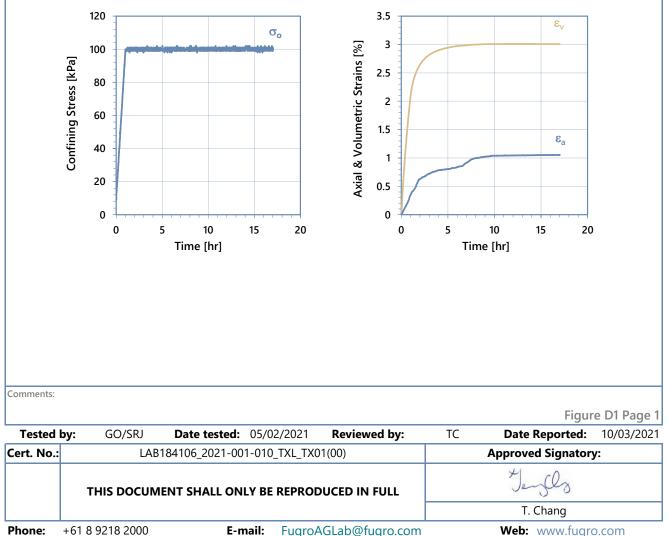
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IESI CERI	IFICAI	C			Advanced	Geotechnica	al Laboratory	
MONOTONIC ⁻	TRIAXIAL				Unit 2, 10 Carbon Court			
Test Method: AG	Lab Test Pr	ocedure FAM-17864				Osborne Pa	rk, WA, 6017	
Undrained Triaxial Compression							Australia	
Client:	GHD Pty	HD Pty Ltd				LAB1841	06	
Client Address:	2 Salama	nca Square, Hobart, Tasm	ania, 7	000	IPO No.:	2021-001	l	
Project Names TCO DSE 2 Contactorial Investigation				Lab Sample	I D: 2021-001	-010		
Project Name: TGO RSF 2 Geotechnical Investigation				Borehole ID:	BH01			
Project Location:	: Tomingle	y, NSW			Sample No.:	U75_02		
Sample Description	Brownish	yellow/grey CLAY			Depth:	3.51 m to	o 3.70 m	
Test Details:				Sample Details:		Initial	Final	
Test ID:		TGO-TX01	1	Sample Diamete	r [mm]:	72.10	82.19	
Confining Street IV	Dalı	σο	1	Sample Height [mm]:	149.81	111.82	
Confining Stress [k	Paj:	100	1	Dry Density [t/m	³]:	1.71	1.76	
Loading rate [%/hr]:	1]	Moisture Conten	ıt [%]:	19.7*	22.5	
				Note: * Moisture con	tent calculated usi	ng trimmings; m	ay not be equal	

Stage 1: Consolidation



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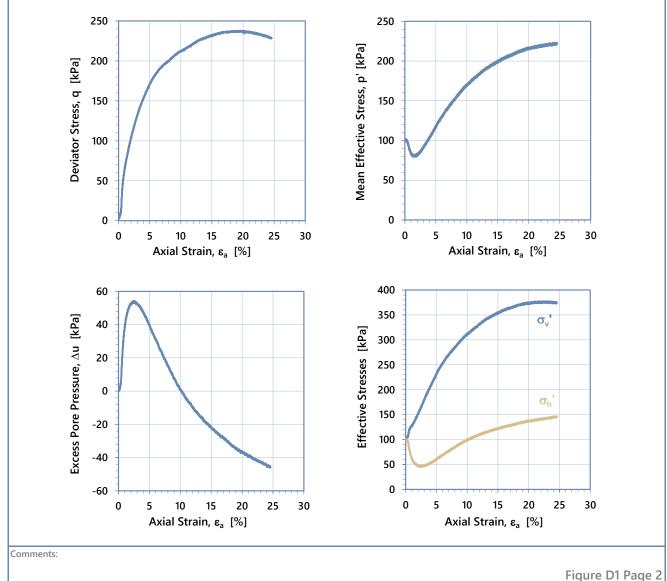
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TEST CERTIFICATE		Fugro Au	Fugro Australia Marine Pty Ltd		
IESI CERI	IFICATE	Advanced (Geotechnical Laboratory		
MONOTONIC	TRIAXIAL		Unit 2, 10 Carbon Court		
Test Method: AG	Lab Test Procedure FAM-17864	(Osborne Park, WA, 6017		
Undrained Triaxi	al Compression		Australia		
Client:	GHD Pty Ltd	Job No.:	LAB184106		
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001		
Dreiget Norma	TCO RSE 2 Controbuical Investigation	Lab Sample II): 2021-001-010		
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH01		
Project Location	: Tomingley, NSW	Sample No.:	U75_02		
Sample Description	Brownish yellow/grey CLAY	Depth:	3.51 m to 3.70 m		
Stage 2. Loadin	a				

Stage 2: Loading



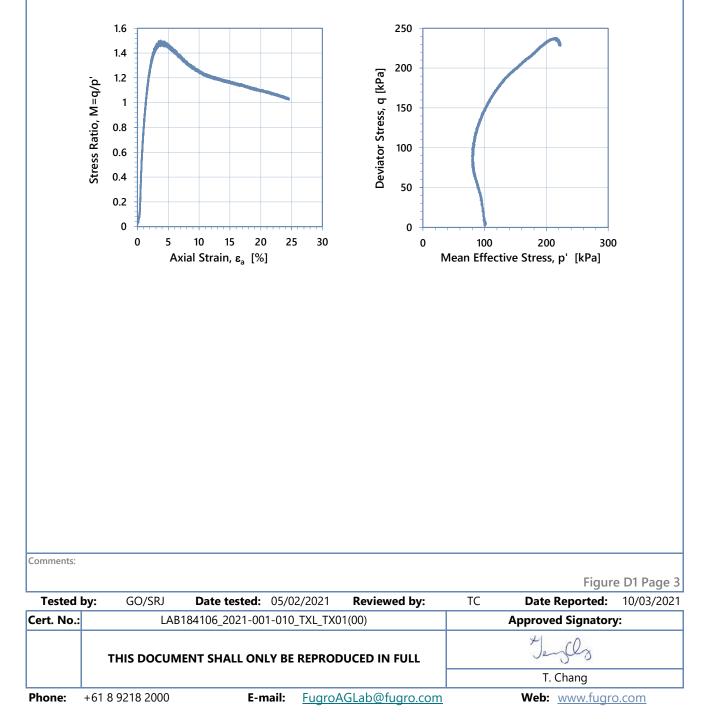
Tested	by: GO/SRJ	Date tested: 05/	/02/2021 Rev i	ewed by:	TC	Date Reported:	10/03/2021
Cert. No.:	LAB	184106_2021-001-01	10_TXL_TX01(00)			Approved Signatory	/:
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IESI CERI	IFICATE	Advanced (Geotechnical Laboratory		
MONOTONIC	TRIAXIAL		Unit 2, 10 Carbon Court		
Test Method: AG	Lab Test Procedure FAM-17864	(Osborne Park, WA, 6017		
Undrained Triaxial Compression			Australia		
Client:	GHD Pty Ltd	Job No.:	LAB184106		
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001		
Dreiget Norma	TCO RSE 2 Controbuical Investigation	Lab Sample II	D: 2021-001-010		
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH01		
Project Location	: Tomingley, NSW	Sample No.:	U75_02		
Sample Description	Brownish yellow/grey CLAY	Depth:	3.51 m to 3.70 m		
Stage 2: Loadin	g [continued]				

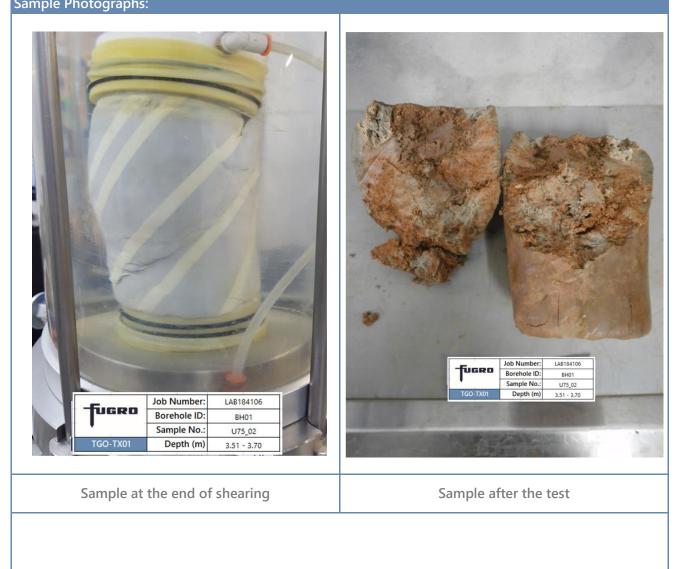


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IESI CERI	IFICATE	Advanced (Geotechnical Laboratory		
MONOTONIC	TRIAXIAL		Unit 2, 10 Carbon Court		
Test Method: AG	Lab Test Procedure FAM-17864	(Osborne Park, WA, 6017		
Undrained Triaxi	ained Triaxial Compression				
Client:	GHD Pty Ltd	Job No.:	LAB184106		
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001		
Dreiget Norma	TCO RSE 2 Controbuical Investigation	Lab Sample II): 2021-001-010		
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH01		
Project Location	: Tomingley, NSW	Sample No.:	U75_02		
Sample Description	Brownish yellow/grey CLAY	Depth:	3.51 m to 3.70 m		
Sample Photog	ranhs				



Comments:							
						Figure	e D1 Page 4
Tested	by: GO/SRJ	Date tested: 05/	/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021
Cert. No.:	t. No.: LAB184106_2021-001-010_TXL_TX01(00) Approved Signator			Approved Signatory	/:		
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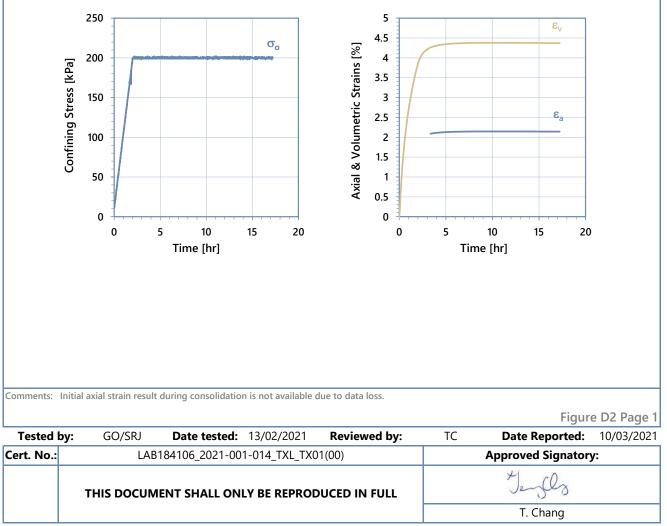
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MONOTONIC ⁻	TRIAXIAL					Unit 2, 10 0	Carbon Court
Test Method: AG	Lab Test Pr	ocedure FAM-17864				Osborne Pa	rk, WA, 6017
Undrained Triaxia	al Compres	sion					Australia
Client:	GHD Pty	Ltd			Job No.:	LAB1841	06
Client Address:	2 Salama	nca Square, Hobart, Tasr	mania, 7	7000	IPO No.:	2021-001	
				Lab Sample ID: 2021-001-014			
Project Name:	IGO KSF	2 Geotechnical Investiga	ation		Borehole ID:	BH01	
Project Location:	: Tomingle	y, NSW			Sample No.:	U75_02	
Sample Description	: Brownish	yellow/grey CLAY with s	sand		Depth:	3.70 m to	o 3.89 m
Test Details:				Sample Details:		Initial	Final
Test ID:		TGO-TX02		Sample Diamete	r [mm]:	72.10	79.85
Confining Strange II	kDeli	σο		Sample Height [nm]:	149.81	116.81
Confining Stresss [kPa]:		200		Dry Density [t/m ³]:		1.62	1.71
Loading rate [%/hr]:	1	Moisture Content		t [%]:	22.7*	23.0
		-		Note: * Moisture con	tent calculated usi	ng trimmings; m	ay not be equal

Stage 1: Consolidation



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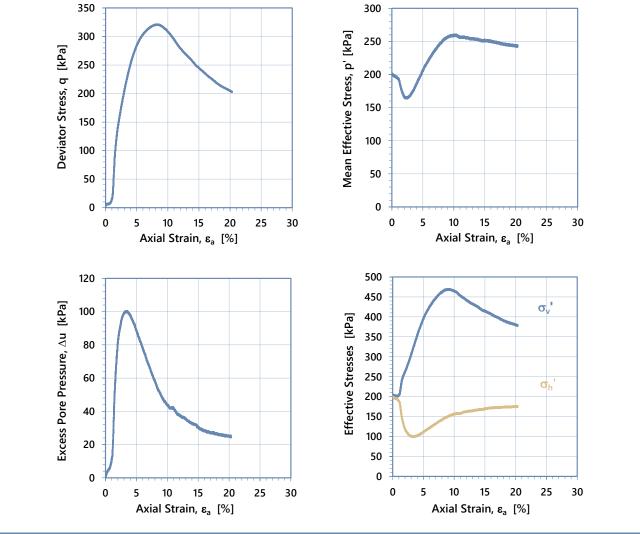
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MONOTONIC	TRIAXIAL		Unit 2, 10 Carbon Court		
Test Method: AG	Lab Test Procedure FAM-17864	(Osborne Park, WA, 6017		
Undrained Triaxial Compression			Australia		
Client:	GHD Pty Ltd	Job No.:	LAB184106		
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001		
Dreigst Norma	TCO RSE 2 Controbuical Investigation	Lab Sample II	D: 2021-001-014		
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH01		
Project Location	: Tomingley, NSW	Sample No.:	U75_02		
Sample Description	Brownish yellow/grey CLAY with sand	Depth:	3.70 m to 3.89 m		
Stage 2. Loadin	a				





Comments: Initial axial strain result during consolidation is not available due to data loss.

Figure D2 Page 2

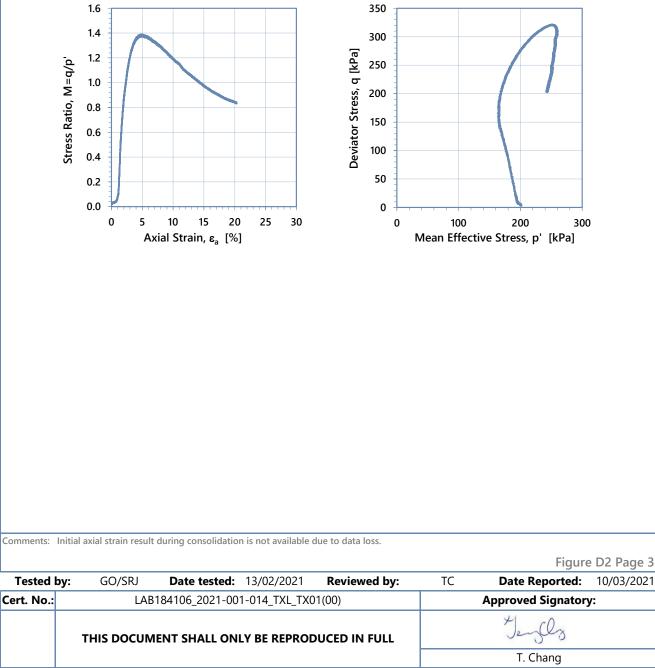
Tested	by: GO/SRJ	Date tested: 13/	/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021	
Cert. No.:	Cert. No.: LAB184106_2021-001-014_TXL_TX01(00)					Approved Signatory:		
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						T. Chang		
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MONOTONIC	TRIAXIAL		Unit 2, 10 Carbon Court		
Test Method: AG	Lab Test Procedure FAM-17864		Osborne Park, WA, 6017		
Undrained Triaxia	al Compression		Australia		
Client:	GHD Pty Ltd	Job No.:	LAB184106		
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001		
Dreiget Norma	TCO RSE 2 Controchnical Investigation	Lab Sample I	D: 2021-001-014		
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH01		
Project Location	: Tomingley, NSW	Sample No.:	U75_02		
Sample Description	Brownish yellow/grey CLAY with sand	Depth:	3.70 m to 3.89 m		
Stage 2: Loadin	g [continued]				



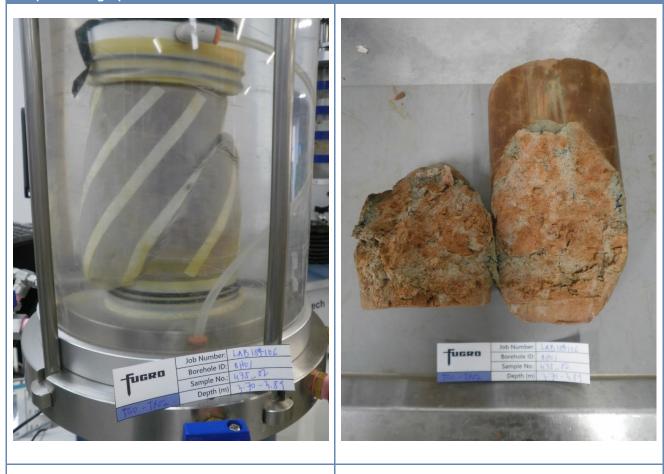
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E-mail: FugroAGLab@fugro.com

Web: <u>www.fugro.com</u>

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TEST CERTIFICATE		Fugro A	Fugro Australia Marine Pty Ltd		
IESI CERI	IFICATE	Advanced	Geotechnical Laboratory		
MONOTONIC [®]	TRIAXIAL		Unit 2, 10 Carbon Court		
Test Method: AG	Lab Test Procedure FAM-17864		Osborne Park, WA, 6017		
Undrained Triaxial Compression			Australia		
Client:	GHD Pty Ltd	Job No.:	LAB184106		
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001		
Project Name:	TCO BSE 2 Controbuical Investigation	Lab Sample I	D: 2021-001-014		
Project Name.	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH01		
Project Location	: Tomingley, NSW	Sample No.:	U75_02		
Sample Description	Brownish yellow/grey CLAY with sand	Depth:	3.70 m to 3.89 m		
Sample Photog	raphs:				



Sample at the end of shearing

Sample after the test

Comments: Initial axial strain result during consolidation is not available due to data loss.

Figure D2 Page 4 TC 10/03/2021 Tested by: GO/SRJ **Date tested:** 13/02/2021 **Reviewed by:** Date Reported: Cert. No.: LAB184106_2021-001-014_TXL_TX01(00) **Approved Signatory:** THIS DOCUMENT SHALL ONLY BE REPRODUCED IN FULL T. Chang Phone: +61 8 9218 2000 E-mail: FugroAGLab@fugro.com Web: <u>www.fugro.com</u>

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TEST CERTIFICATE			Fugro Australia Marine Pty Ltd				
		L			Advanced Geotechnical Laboratory		
MULTISTAGE T	RIAXIAL					Unit 2, 10 (Carbon Court
Test Method: AG	Lab Test Pr	ocedure FAM-17864				Osborne Pa	rk, WA, 6017
Multistage Undra	ined Triaxi	al Compression					Australia
Client:	GHD Pty	Ltd			Job No.:	LAB1841	06
Client Address:	2 Salama	nca Square, Hobart, Tasm	nania, 70	000	IPO No.:	2021-001	
Dreiget Norme		2 Contachnical Investigat	tion		Lab Sample ID: 2021-001-018		
Project Name:	IGO KSF	2 Geotechnical Investigat	lion		Borehole ID:	BH01	
Project Location:	: Tomingle	y, NSW			Sample No.:	U75_03	
Sample Description	: Light bro	wnish grey CLAY with san	nd		Depth:	9.68 m to	9.84 m
Test Details:				Sample Details:		Initial	Final
Test ID:		TGO-TX03	1	Sample Diamete	r [mm]:	72.1	79.7
Confining Street IV	Dalı	σο	1	Sample Height [I	nm]:	149.8	116.5
Confining Stress [kPa]:		130, 330 & 530	1	Dry Density [t/m ³]:		1.82	1.91
Loading rate [%/hr]:	1	Moisture Conten		t [%]:	17.0*	17.5
				Note: * Moisture con	tent calculated usi	ng trimmings; m	ay not be equal

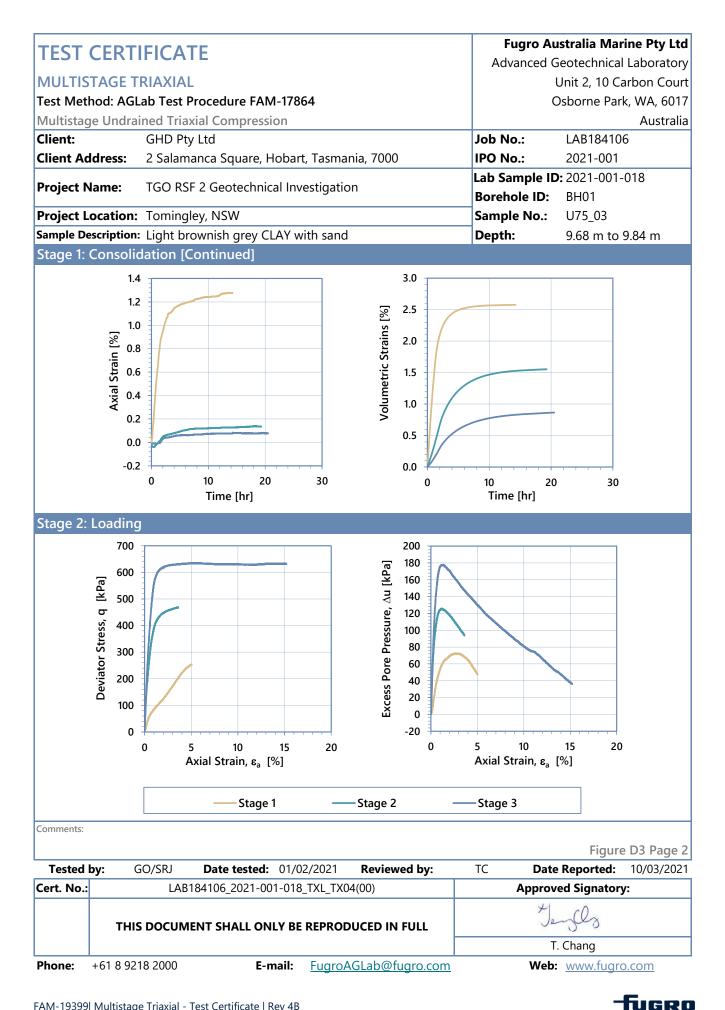
to moisture content of whole sample.

Stage 1: Consolidation

600 500 Confining Stress, م_o [kPa] 400 300 200 100 0 0 5 10 15 20 25 Time [hr] Stage 1 Stage 2 Stage 3 Comments: Figure D3 Page 1 GO/SRJ Tested by: Date tested: 01/02/2021 **Reviewed by:** TC Date Reported: 10/03/2021 Cert. No.: LAB184106_2021-001-018_TXL_TX04(00) **Approved Signatory:** THIS DOCUMENT SHALL ONLY BE REPRODUCED IN FULL T. Chang Phone: +61 8 9218 2000 E-mail: FugroAGLab@fugro.com Web: <u>www.fugro.com</u>

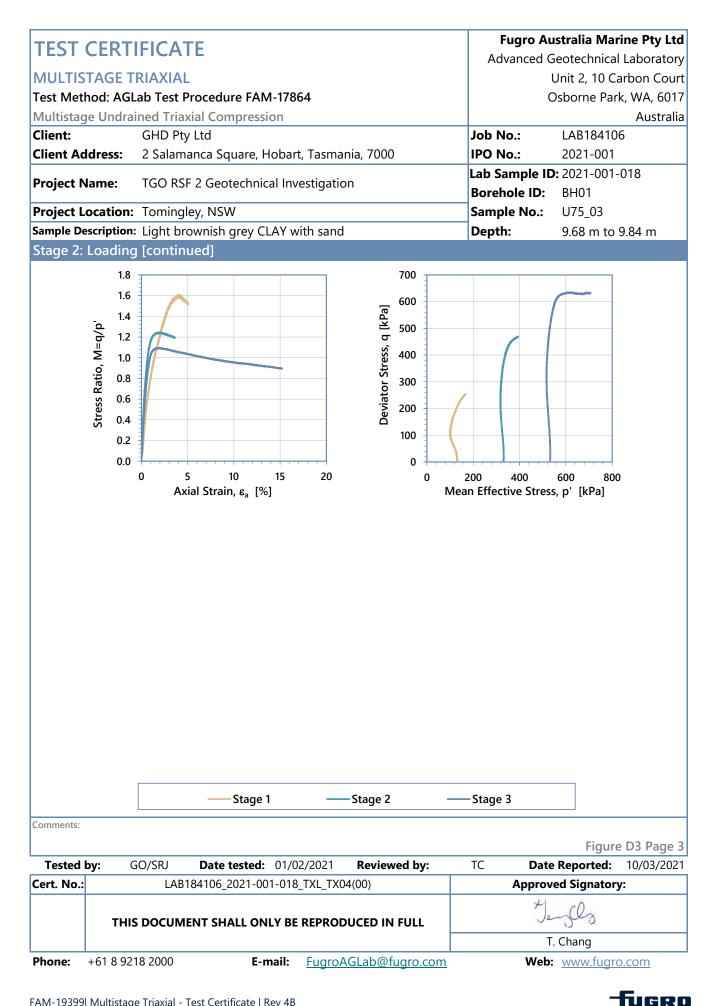
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TEST CERTIFICATE		Fugro A	ustralia Marine Pty Ltd
IESI CERI	IFICATE	Advanced	Geotechnical Laboratory
MULTISTAGE 1	RIAXIAL		Unit 2, 10 Carbon Court
Test Method: AG	Lab Test Procedure FAM-17864		Osborne Park, WA, 6017
Multistage Undrained Triaxial Compression			Australia
Client:	GHD Pty Ltd	Job No.:	LAB184106
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001
Droiget Name	TCO BSE 2 Controbuical Investigation	Lab Sample I	D: 2021-001-018
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH01
Project Location	: Tomingley, NSW	Sample No.:	U75_03
Sample Description	: Light brownish grey CLAY with sand	Depth:	9.68 m to 9.84 m
Sample Photog	raphs:		



Sample at the end of shearing

Sample after test

Comments:								
						Figure	e D3 Page 4	
Tested I	oy: GO/SRJ	Date tested: 0	1/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021	
Cert. No.:	LAB184106_2021-001-018_TXL_TX04(00)					Approved Signatory:		
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						T. Chang		
Phone:	+61 8 9218 2000	E-mai	I: FugroA	GLab@fugro.com		Web: www.fugro	D.COM	



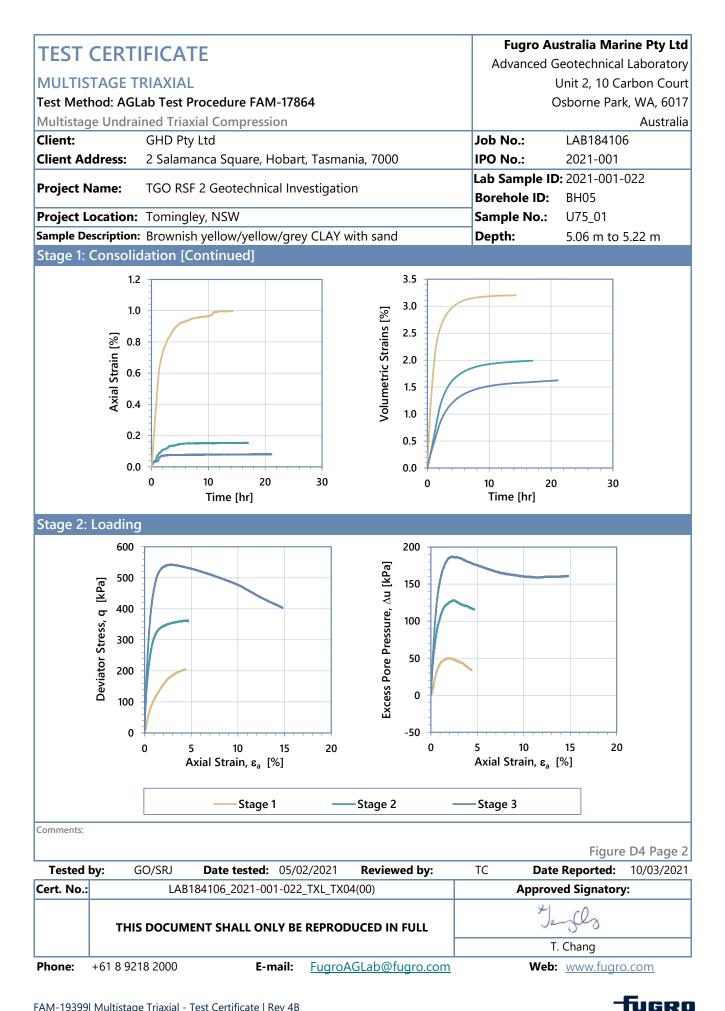
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				Advanced Geotechnical Laboratory				
MULTISTAGE T	RIAXIAL					Unit 2, 10 0	Carbon Court	
Test Method: AG	Lab Test Pr	ocedure FAM-17864				Osborne Pa	rk, WA, 6017	
Multistage Undra	ined Triaxi	al Compression					Australia	
Client:	GHD Pty	Ltd			Job No.:	LAB1841	06	
Client Address:	2 Salama	nca Square, Hobart, Tasma	ania, 7	000	IPO No.:	2021-001		
Due is at Norman					Lab Sample ID: 2021-001-022			
Project Name:	IGO KSF	2 Geotechnical Investigati	on		Borehole ID:	BH05		
Project Location:	: Tomingle	y, NSW			Sample No.:	U75_01		
Sample Description	Brownish	yellow/yellow/grey CLAY	with s	and	Depth:	5.06 m to	o 5.22 m	
Test Details:				Sample Details:		Initial	Final	
Test ID:		TGO-TX04		Sample Diamete	r [mm]:	72.1	78.8	
Confining Charles II.	D-1-	σο	1	Sample Height [mm]:		146.8	114.8	
Confining Stress [kPa]:		120, 300 & 480	1	Dry Density [t/m ³]:		1.71	1.83	
Loading rate [%/hr]: 1			Moisture Content [%]: 19.1* 19.0			19.0		
				Note: * Moisture con	tent calculated usi	ng trimmings; m	ay not be equal	

Stage 1: Consolidation

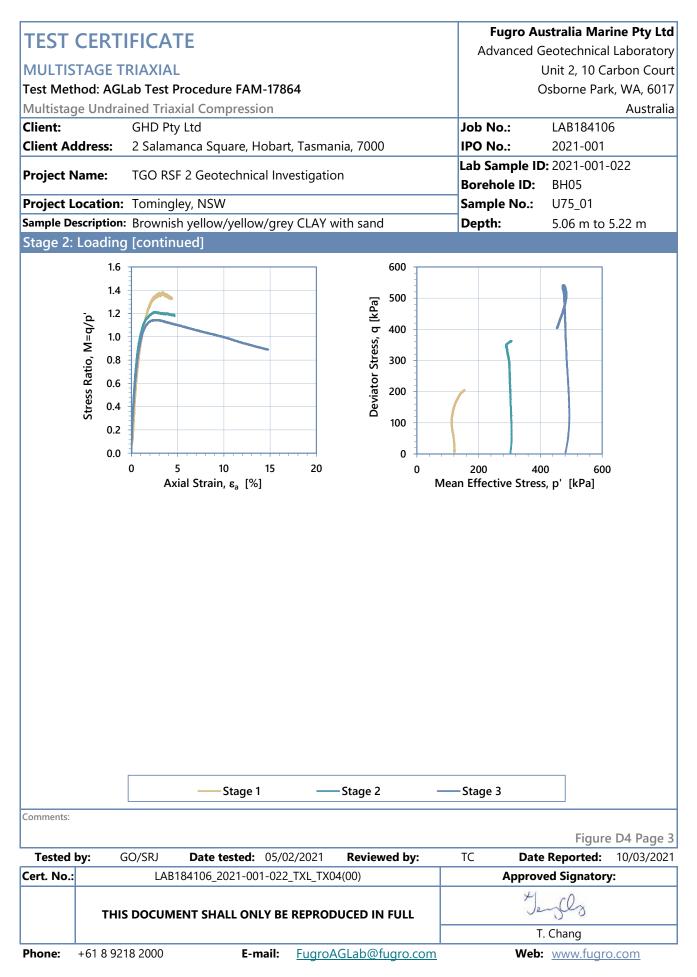
600 500 Confining Stress, م_o [kPa] 400 300 200 100 0 0 5 10 15 20 25 Time [hr] Stage 1 Stage 2 Stage 3 Comments: Figure D4 Page 1 GO/SRJ 10/03/2021 Tested by: Date tested: 05/02/2021 **Reviewed by:** TC **Date Reported:** Cert. No.: LAB184106_2021-001-022_TXL_TX04(00) **Approved Signatory:** THIS DOCUMENT SHALL ONLY BE REPRODUCED IN FULL T. Chang Phone: +61 8 9218 2000 E-mail: FugroAGLab@fugro.com Web: <u>www.fugro.com</u>





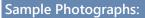
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IESI CERI	IFICATE	Advanced	Advanced Geotechnical Laboratory			
MULTISTAGE	FRIAXIAL		Unit 2, 10 Carbon Court			
Test Method: AG	Lab Test Procedure FAM-17864		Osborne Park, WA, 6017			
Multistage Undra	ained Triaxial Compression		Australia			
Client:	GHD Pty Ltd	Job No.:	LAB184106			
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001			
Dreigst Norma	TCO RSE 2 Controphysical Investigation	Lab Sample I	D: 2021-001-022			
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH05			
Project Location: Tomingley, NSW		Sample No.:	U75_01			
Sample Description	Brownish yellow/yellow/grey CLAY with sand	Depth:	5.06 m to 5.22 m			





Sample at the end of shearing

Sample after test

Comments:								
						Figure	e D4 Page 4	
Tested	by: GO/SRJ	Date tested:	05/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021	
Cert. No.:	LAB	3184106_2021-001	1-022_TXL_T	X04(00)	Approved Signatory:			
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						T. Chang		
Phone:	+61 8 9218 2000	E-m	ail: <u>Fugr</u>	oAGLab@fugro.com		Web: <u>www.fugro</u>	<u>o.com</u>	

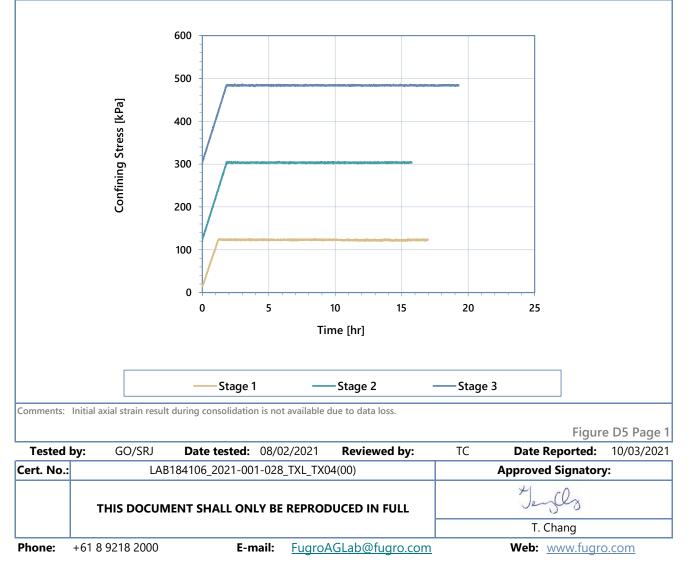


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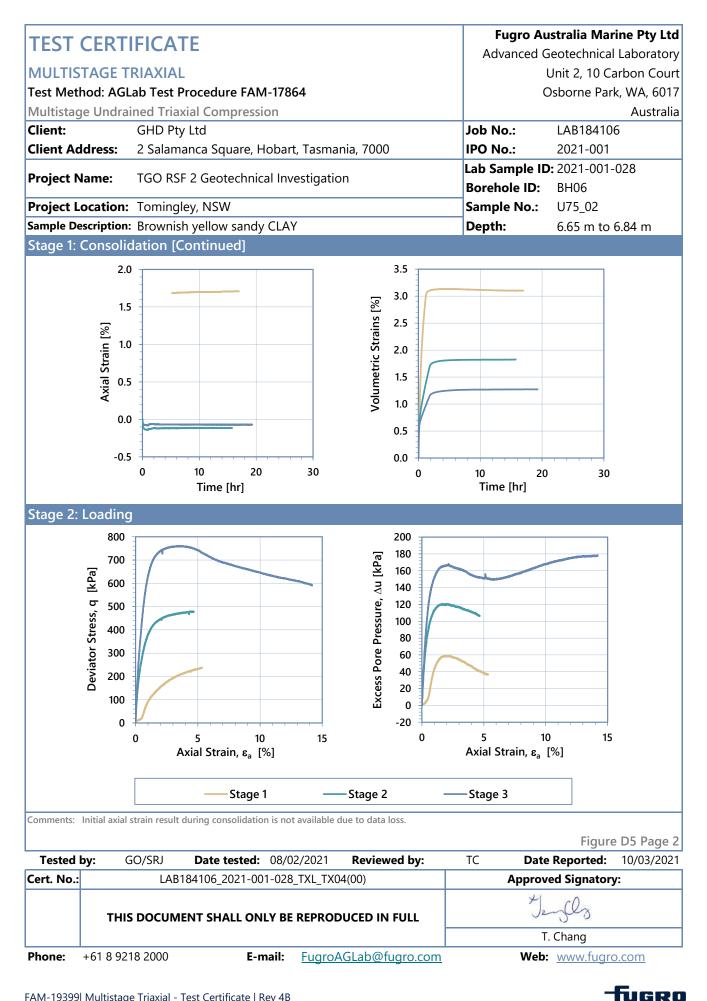


TEST CERTIFICATE					Fugro Australia Marine Pty Ltd			
					Advanced Geotechnical Laboratory			
MULTISTAGE 1	RIAXIAL					Unit 2, 10 0	Carbon Court	
Test Method: AG	Lab Test Pr	ocedure FAM-17864				Osborne Pa	rk, WA, 6017	
Multistage Undra	ined Triaxi	al Compression					Australia	
Client:	GHD Pty	Ltd			Job No.:	LAB1841	06	
Client Address:	2 Salama	nca Square, Hobart, Tasma	ania, 7	000	IPO No.:	2021-001		
		2 Cooto chaicel la vecticati			Lab Sample ID: 2021-001-028			
Project Name:	IGO KSF	2 Geotechnical Investigati	on		Borehole ID:	BH06		
Project Location	: Tomingle	y, NSW			Sample No.:	U75_02		
Sample Description	Brownish	yellow sandy CLAY			Depth:	6.65 m to	o 6.84 m	
Test Details:				Sample Details:		Initial	Final	
Test ID:		TGO-TX05]	Sample Diamete	r [mm]:	72.1	79.4	
Confining Charles II	D-1-	σο	1	Sample Height [mm]:	146.8	113.6	
Confining Stress [kPa]:		120, 300 & 480	1	Dry Density [t/m	³]:	1.86	1.95	
Loading rate [%/hr]: 1			Moisture Content [%]: 12.2* 14			14.4		
			4	Note: * Moisture con	tent calculated usi	ng trimmings; m	ay not be equal	

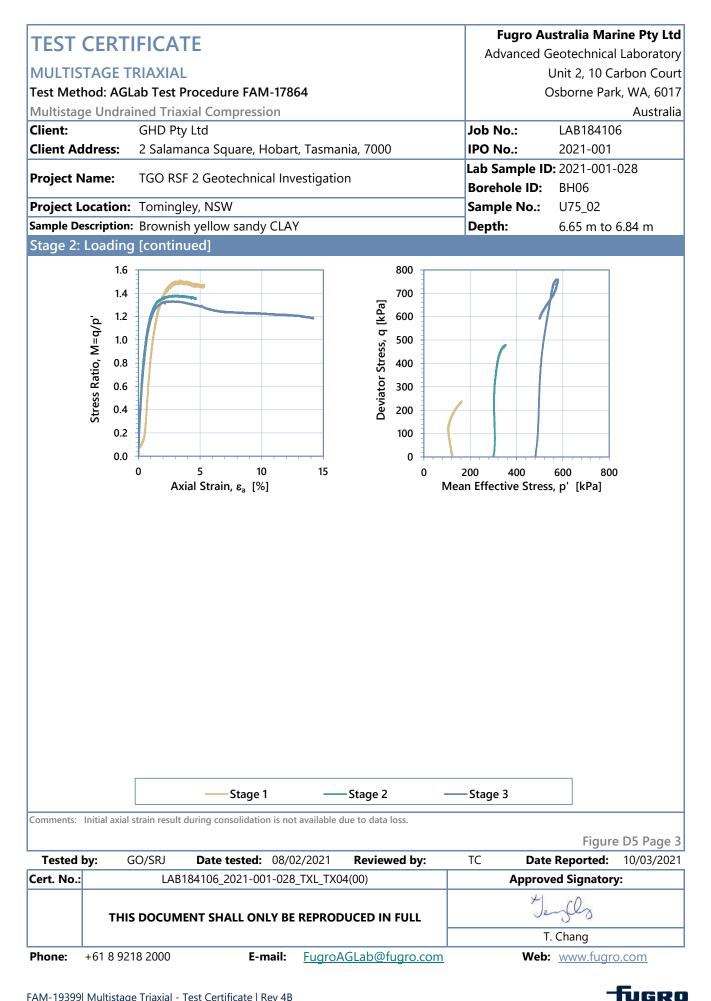




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IESI CERI	IFICATE	Advanced (Geotechnical Laboratory			
MULTISTAGE T	RIAXIAL		Unit 2, 10 Carbon Court			
Test Method: AGI	ab Test Procedure FAM-17864		Osborne Park, WA, 6017			
Multistage Undra	ined Triaxial Compression		Australia			
Client:	GHD Pty Ltd	Job No.:	LAB184106			
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001			
Dreiget Norma	TCO BSE 2 Controphysical Investigation	Lab Sample ID: 2021-001-028				
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH06			
Project Location: Tomingley, NSW		Sample No.:	U75_02			
Sample Description	Brownish yellow sandy CLAY	Depth:	6.65 m to 6.84 m			
Sample Photographs:						



Sample at the end of shearing

Sample after test

Comments: Initial axial strain result during consolidation is not available due to data loss.

						Figure	e D5 Page 4	
Tested	by: GO/SRJ	Date tested:	08/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021	
Cert. No.:	t. No.: LAB184106_2021-001-028_TXL_TX04(00)			Approved Signatory:				
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Phone:	+61 8 9218 2000	E-ma	ail: <u>Fugro</u> A	GLab@fugro.com		Web: www.fugro	<u>o.com</u>	

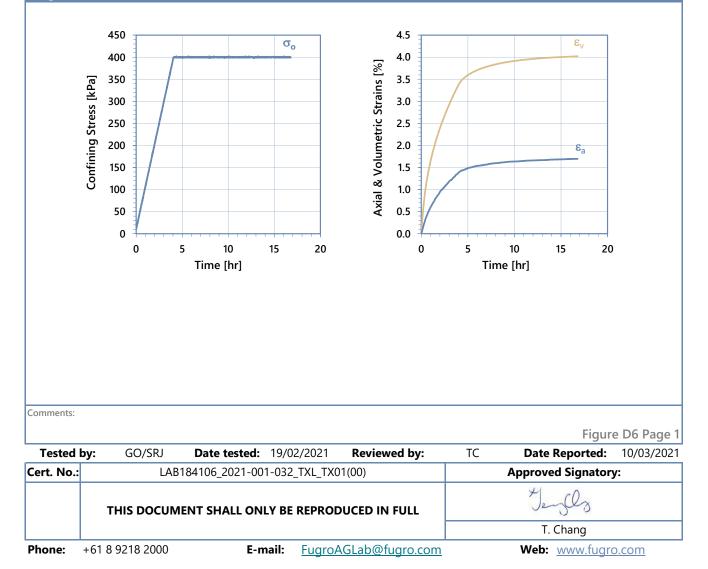
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				Advanced Geotechnical Laboratory			
MONOTONIC ⁻	TRIAXIAL					Unit 2, 10 0	Carbon Court
Test Method: AG	Lab Test Pr	ocedure FAM-17864				Osborne Pa	rk, WA, 6017
Undrained Triaxia	al Compres	sion					Australia
Client:	GHD Pty	Ltd			Job No.:	LAB1841	06
Client Address:	2 Salama	nca Square, Hobart, Tasn	nania, 7	7000	IPO No.:	2021-001	
Dreigst Norma		2 Cootochnical Investiga	tion		Lab Sample ID: 2021-001-032		
Project Name:	IGO KSF	2 Geotechnical Investiga	lion		Borehole ID:	BH07	
Project Location:	: Tomingle	y, NSW			Sample No.:	U75_01	
Sample Description	: Brownish	yellow/grey CLAY with s	and		Depth:	3.52 m to	o 3.71 m
Test Details:				Sample Details:		Initial	Final
Test ID:		TGO-TX06		Sample Diamete	r [mm]:	72.10	81.04
Confining Strees II	Dalı	σο		Sample Height [mm]:		149.81	113.82
Confining Stress [kPa]:		400	7	Dry Density [t/m ³]:		1.76	1.83
Loading rate [%/hr]: 1			7	Moisture Content [%]: 17.5* 18.9			18.9
				Note: * Moisture con	tent calculated usi	ng trimmings; m	ay not be equal

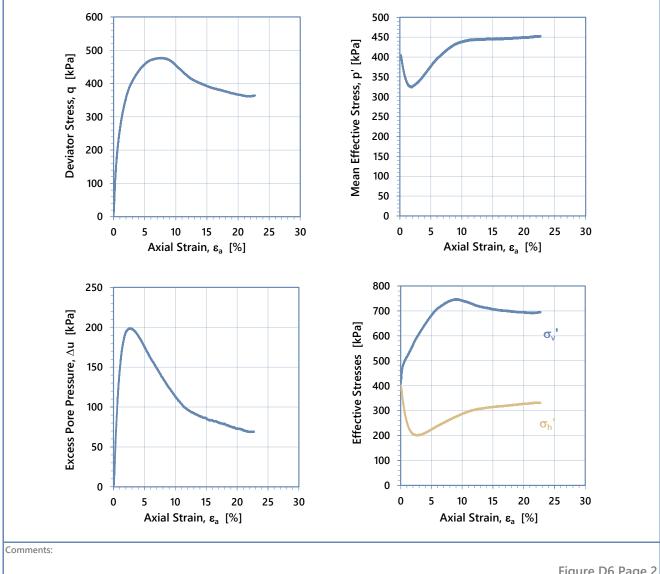
Stage 1: Consolidation





TEST CERT	TEICATE	Fugro Au	Fugro Australia Marine Pty Ltd			
IESI CERI	IFICATE	Advanced (Advanced Geotechnical Laboratory			
MONOTONIC	TRIAXIAL		Unit 2, 10 Carbon Court			
Test Method: AG	Lab Test Procedure FAM-17864	(Osborne Park, WA, 6017			
Undrained Triaxi	al Compression		Australia			
Client:	GHD Pty Ltd	Job No.:	LAB184106			
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001			
Project Name:	TCO RSE 2 Controbuical Investigation	Lab Sample II): 2021-001-032			
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH07			
Project Location	: Tomingley, NSW	Sample No.:	U75_01			
Sample Descriptior	Brownish yellow/grey CLAY with sand	Depth:	3.52 m to 3.71 m			
Stage 2. Loadin	a					

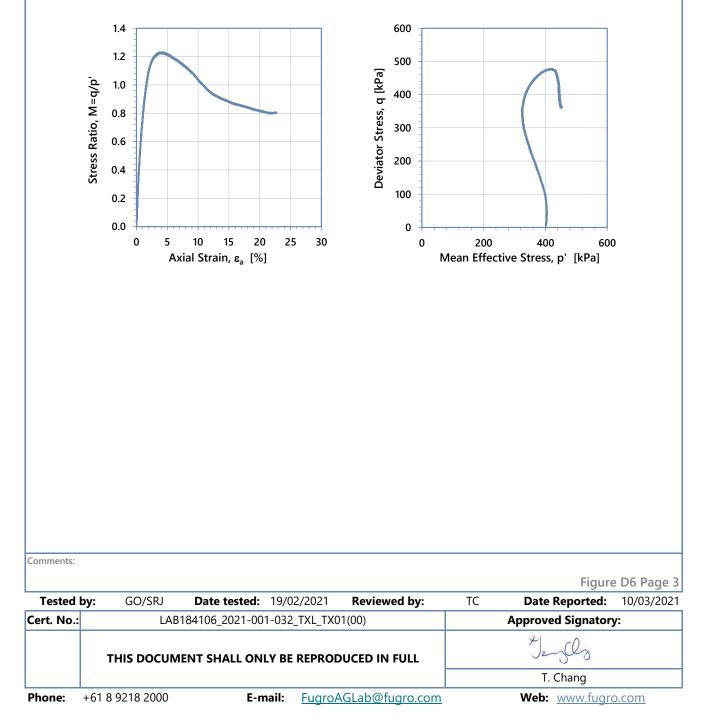
Stage 2: Loading



						rigure	DU Faye Z	
Tested	by: GO/SRJ	Date tested:	19/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021	
Cert. No.:	rt. No.: LAB184106_2021-001-032_TXL_TX01(00)					Approved Signatory:		
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IESI CERI	IFICATE	Advanced (Geotechnical Laboratory
MONOTONIC	TRIAXIAL		Unit 2, 10 Carbon Court
Test Method: AG	Lab Test Procedure FAM-17864	(Osborne Park, WA, 6017
Undrained Triaxi	al Compression		Australia
Client:	GHD Pty Ltd	Job No.:	LAB184106
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001
Ducient Norman	TCO DSE 2 Control Investigation	Lab Sample II): 2021-001-032
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH07
Project Location	: Tomingley, NSW	Sample No.:	U75_01
Sample Description	Brownish yellow/grey CLAY with sand	Depth:	3.52 m to 3.71 m
Stage 2: Loadin	g [continued]	Ċ.	



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IESI CERI	IFICATE	Advanced (Geotechnical Laboratory
MONOTONIC	TRIAXIAL		Unit 2, 10 Carbon Court
Test Method: AG	Lab Test Procedure FAM-17864	(Osborne Park, WA, 6017
Undrained Triaxia	al Compression		Australia
Client:	GHD Pty Ltd	Job No.:	LAB184106
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001
Dreigst Norma	TCO BSE 2 Contachnical Investigation	Lab Sample II): 2021-001-032
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH07
Project Location	: Tomingley, NSW	Sample No.:	U75_01
Sample Description	Brownish yellow/grey CLAY with sand	Depth:	3.52 m to 3.71 m
Sample Dhotog	ronhei		



Sample at the end of shearing

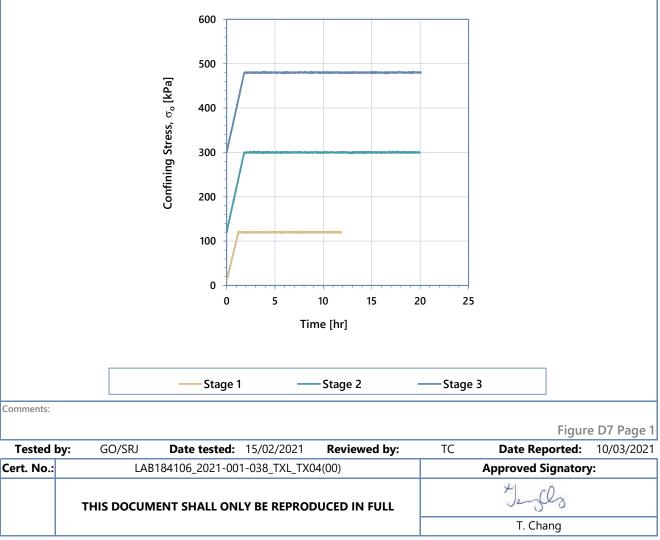
Sample after the test

Comments:									
								Figure	e D6 Page 4
Tested	by: GO/S	RJ Date tested	: 19/02	2/2021	Reviewed by:	TC	Date	Reported:	10/03/2021
Cert. No.:		LAB184106_2021-0	01-032_	TXL_TX0	1(00)		Approve	d Signatory	/:
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TEST CERT	ΙΕΙΟΛΤ	c			Fugro A	Australia Ma	rine Pty Ltd
IESI CERI	IFICAT	E			Advanced	Geotechnica	al Laboratory
MULTISTAGE T	RIAXIAL					Unit 2, 10 C	Carbon Court
Test Method: AGI	Lab Test Pr	ocedure FAM-17864				Osborne Pa	rk, WA, 6017
Multistage Undra	ined Triaxi	al Compression					Australia
Client:	GHD Pty	Ltd			Job No.:	LAB18410	06
Client Address:	2 Salama	nca Square, Hobart, Tasma	ania, 7	000	IPO No.:	2021-001	
Project Name		2 Cootochnical Investigati	on		Lab Sample	ID: 2021-001	-038
Project Name:	IGO KSF	2 Geotechnical Investigati	OII		Borehole ID	: BH08	
Project Location:	Tomingle	y, NSW			Sample No.:	U75_01	
Sample Description	: Brownish	yellow/grey CLAY trace s	and		Depth:	5.16 m to	5.35 m
Test Details:				Sample Details:		Initial	Final
Test ID:		TGO-TX07]	Sample Diamete	r [mm]:	72.1	78.8
Confining Stross [k		σο		Sample Height [nm]:	179.8	144.2
Confining Stress [k	raj.	120, 300 & 480		Dry Density [t/m	³]:	1.40	1.47
Loading rate [%/hr]:	1]	Moisture Conten	t [%]:	20.1*	21.2
			-	Note: * Moisture con		ng trimmings; m	ay not be equal
				to moisture content	of whole sample.		

Stage 1: Consolidation

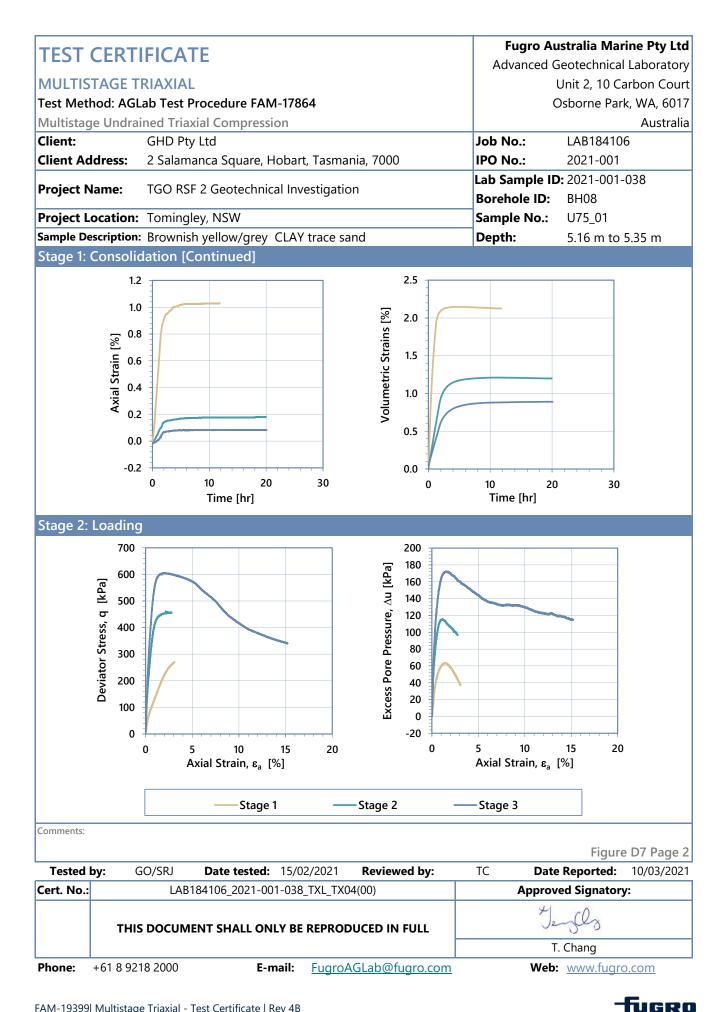


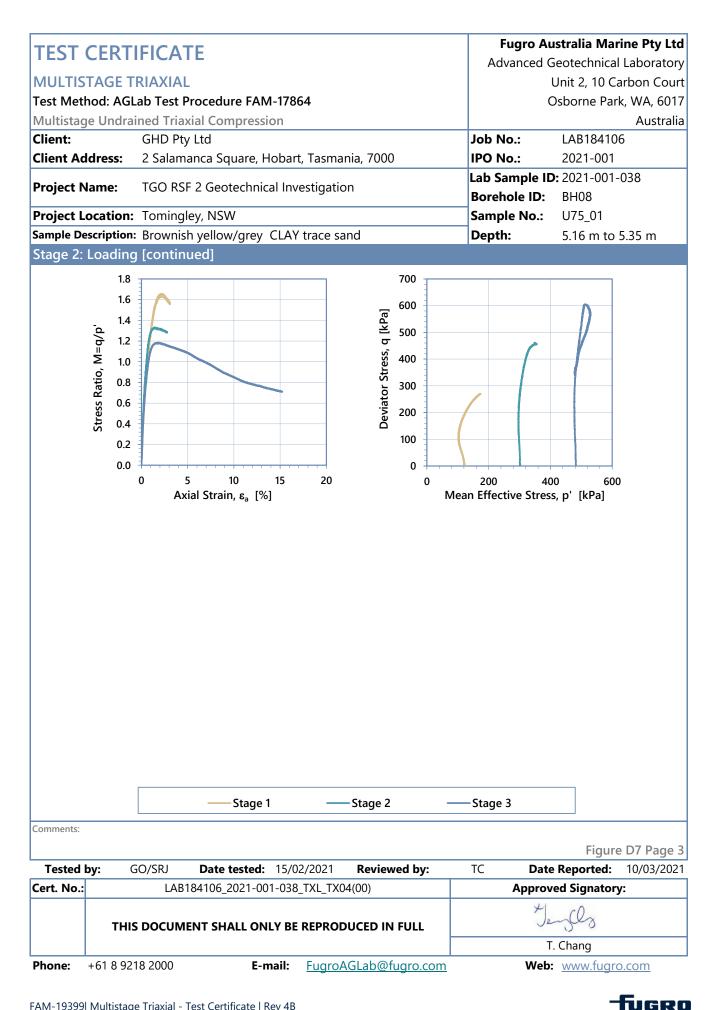
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E-mail: FugroAGLab@fugro.com

Web: www.fugro.com







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MULTISTAGE	FRIAXIAL		Unit 2, 10 Carbon Court
Test Method: AG	Lab Test Procedure FAM-17864	(Osborne Park, WA, 6017
Multistage Undra	ained Triaxial Compression		Australia
Client:	GHD Pty Ltd	Job No.:	LAB184106
Client Address:	2 Salamanca Square, Hobart, Tasmania, 7000	IPO No.:	2021-001
Dreigst Norma	TCO RSE 2 Contachnical Investigation	Lab Sample II	D: 2021-001-038
Project Name:	TGO RSF 2 Geotechnical Investigation	Borehole ID:	BH08
Project Location	: Tomingley, NSW	Sample No.:	U75_01
Sample Description	Brownish yellow/grey CLAY trace sand	Depth:	5.16 m to 5.35 m
Sample Dhotog	ve ele el		



Sample at the end of shearing

Sample after test

Comments:							
						Figure	e D7 Page 4
Tested b	oy: GO/SRJ	Date tested:	15/02/2021	Reviewed by:	TC	Date Reported:	10/03/2021
Cert. No.:	LAB	184106_2021-001	-038_TXL_TX	(04(00)		Approved Signator	y:
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Phone:	+61 8 9218 2000	E-m	ail: <u>Fugro</u>	AGLab@fugro.com		Web: www.fugro	o.com



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Document Status

Revision	Author	Reviewer		Approved for	lssue	
		Name	Signature	Name	Signature	Date
A	T.Swinoga J.Chan	J.De La Rosa	* J.De La Rosa	R. Longey	* R. Longey	14/09/2021

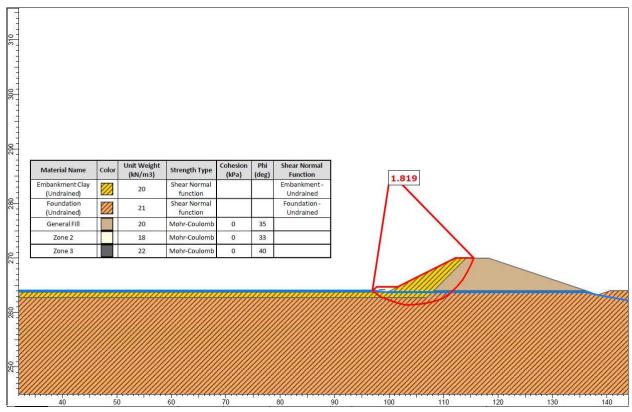
www.ghd.com



Appendix D RSF2 Slope Stability Analysis

Perimeter Embankment

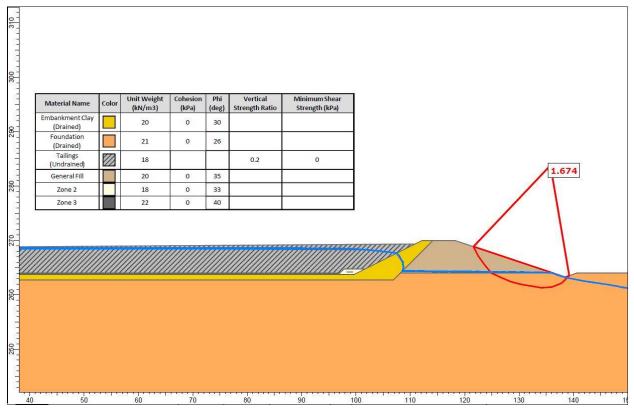
Stage 1



Perimeter Embankment Stage 1, Short Term Loading Case Upstream

Material Nan	ne Color	Unit Weight (kN/m3)	Strength Type	Cohesion (kPa)	Phi (deg)	Shear Normal Function	1					
Embankment ((Undrained		20	Shear Normal function	[kra]	(ucg)	Embankment - Undrained	-			1.8	326	
Foundation (Undrained		21	Shear Normal function			Foundation - Undrained						
General Fill		20	Mohr- Coulomb	0	35							
Zone 2		18	Mohr- Coulomb	0	33				/			
Zone 3		22	Mohr- Coulomb	0	40				_/			
						4			7			1111
50	••••••••••••••••••••••••••••••••••••••	60	70	80		90	100	110	120	130	140	1

Perimeter Embankment Stage 1, Short Term Loading Case Downstream

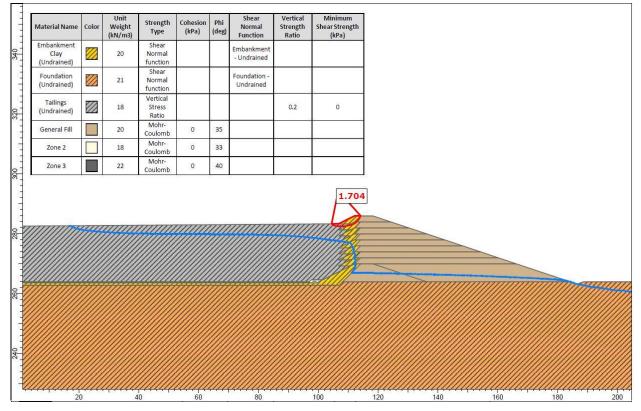


Perimeter Embankment Stage 1, Long Term Loading Case

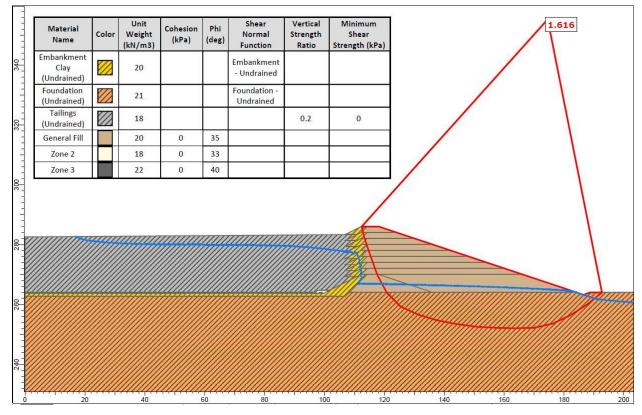
					-				•			
Material Name	Color	Unit Weight (kN/m3)	Strength Type	Cohesion (kPa)	Phi (deg)	Shear Normal Function	Vertical Strength Ratio	Minimum Shear Strength (kPa)				
Embankment Clay (Undrained)		20	Shear Normal function			Embankment - Undrained			1			
Embankment Clay (Post-Seismic)		20	Shear Normal function			Embankment - Post-Seismic]			
Foundation (Post- Seismic)		21	Shear Normal function			Foundation - Post-Seismic]			
Tailings (Undrained)		18	Vertical Stress Ratio				0.2	0				
Tailings (Post- Seismic Liq)		18	Vertical Stress Ratio				0.038	0				
General Fill		20	Mohr- Coulomb	0	35							
Zone 2		18	Mohr- Coulomb	0	33							_
Zone 3		22	Mohr- Coulomb	0	40					/	1.490)
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Perimeter Embankment Stage 1, Post Seismic Loading Case

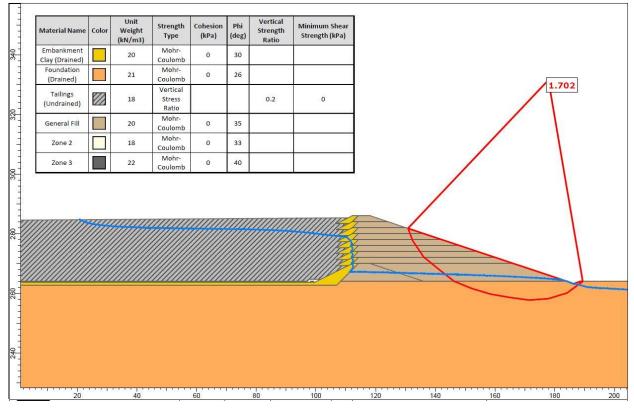
Stage 9



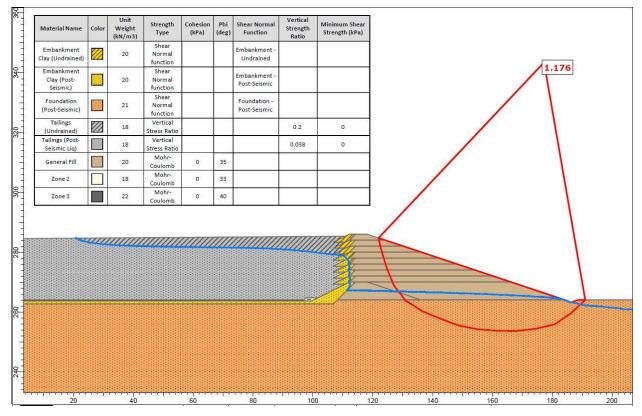
Perimeter Embankment Stage 9, Short Term Loading Case Upstream



Perimeter Embankment Stage 9, Short Term Loading Case Downstream



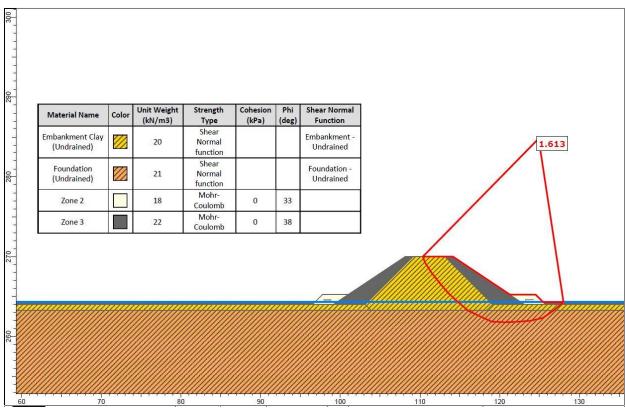
Perimeter Embankment Stage 9, Long Term Loading Case



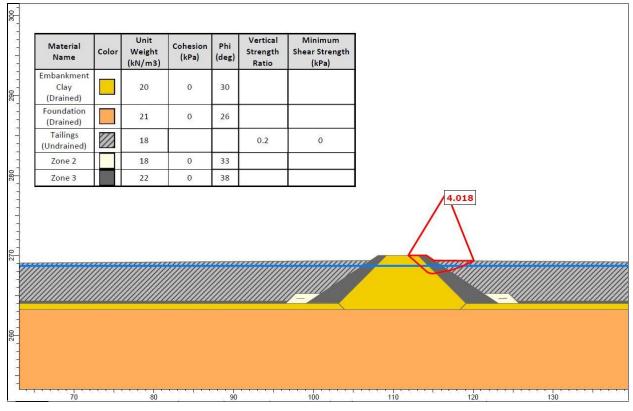
Perimeter Embankment Stage 9, Post Seismic Loading Case

Common Wall

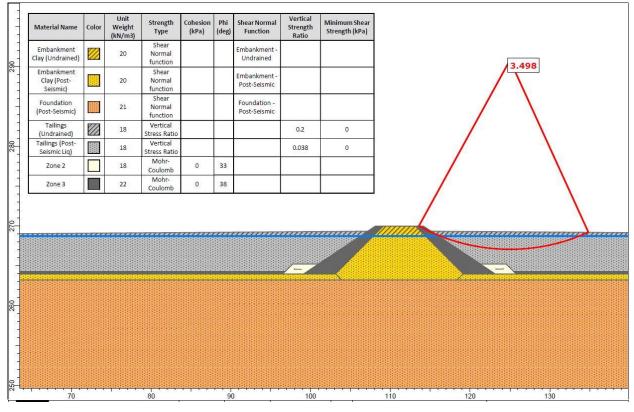
Stage 1



Common Wall Stage 1, Short Term Loading Case

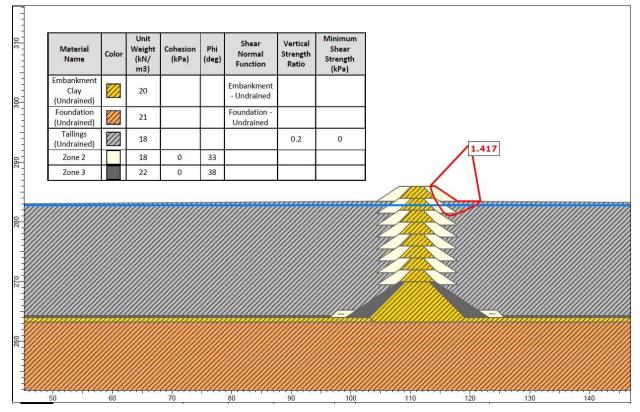


Common Wall Stage 1, Long Term Loading Case

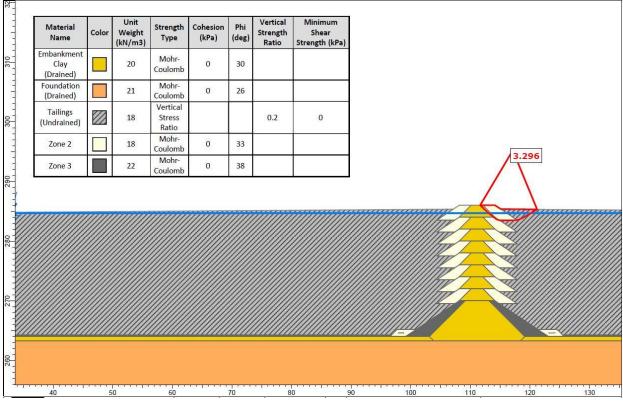


Common Wall Stage 1, Post Seismic Loading Case

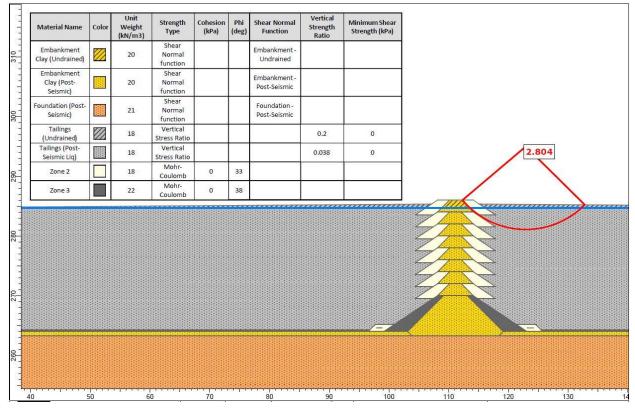
Stage 9



Common Wall Stage 9, Short Term Loading Case



Common Wall Stage 9, Long Term Loading Case

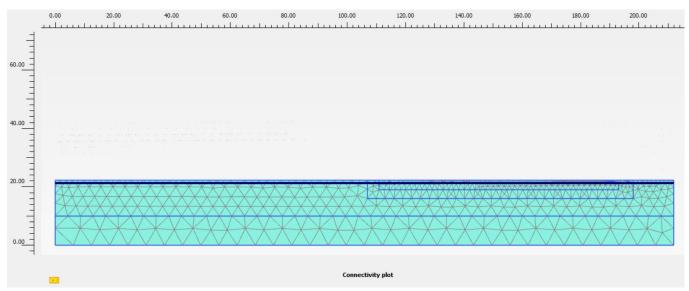


Common Wall Stage 9, Post Seismic Loading Case

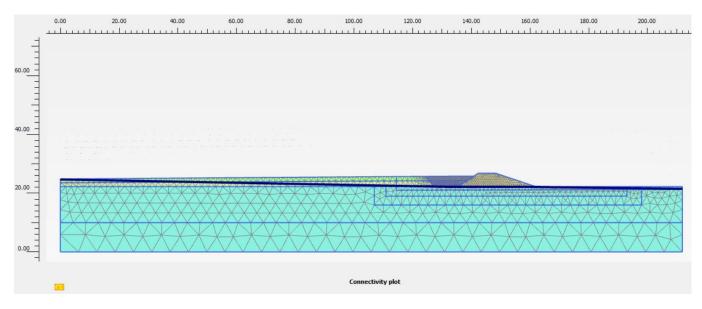
Appendix E RSF2 Finite Element Modelling

Construction Stages

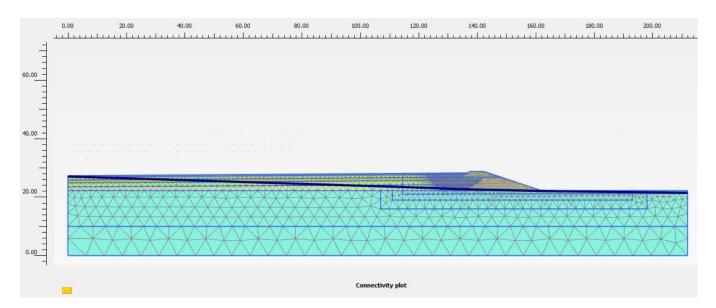
RSF 1



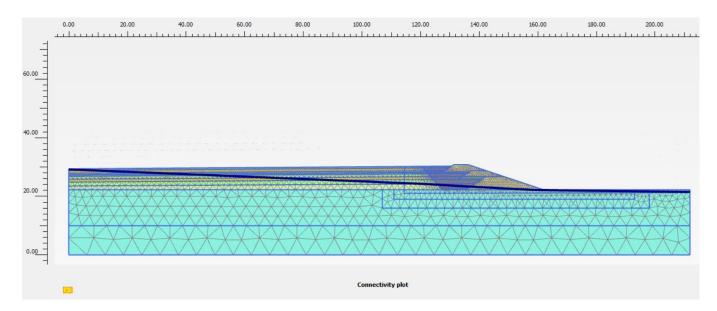




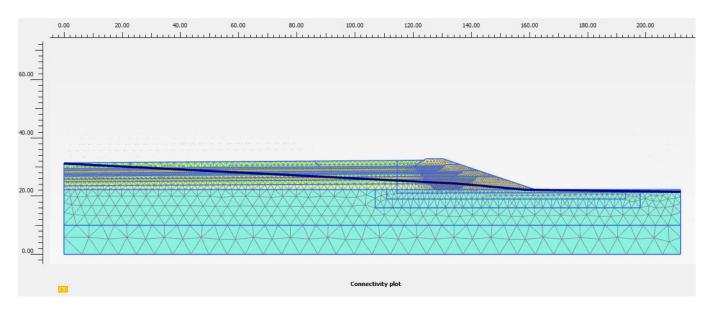
Stage 2: Construction of the Starter Dam. Tailings deposition to RL 269.85 m



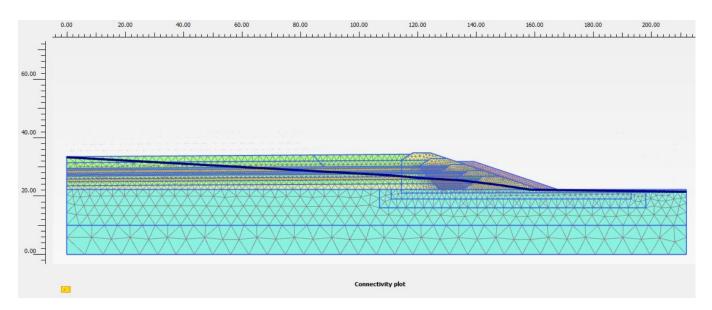
Stage 3: Construction of embankment raise 2. Tailings deposition to RL 271.85 m.



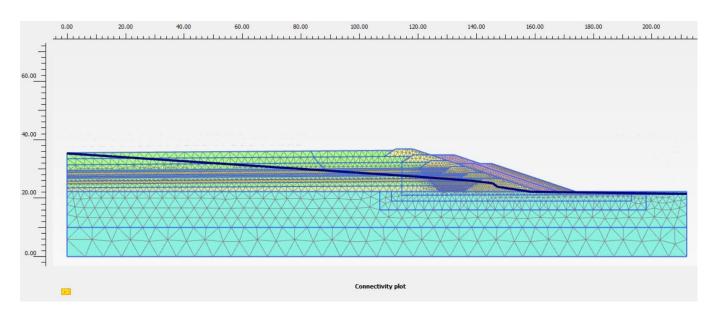
Stage 4: Construction of embankment raise 3. Tailings deposition to RL 273.85 m



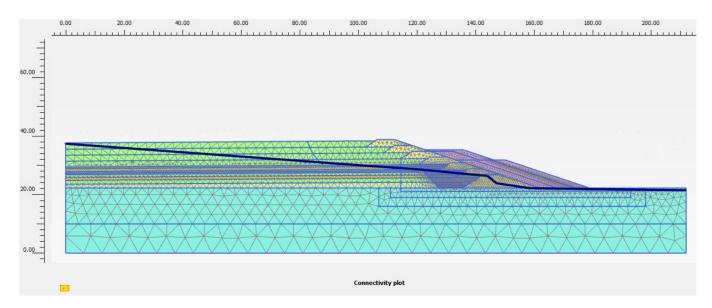
Stage 5: Construction of embankment raise 4. Tailings deposition to RL 275.85 m



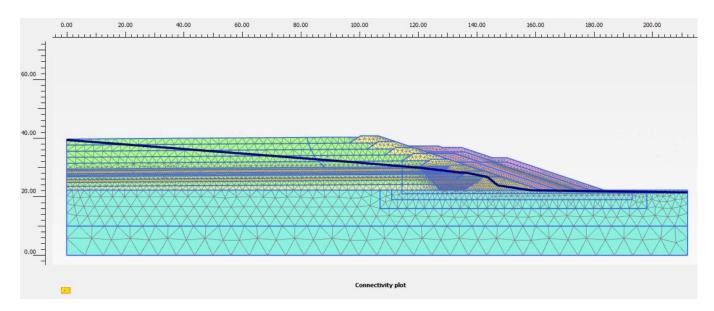
Stage 6: Construction of embankment raise 5 and buttress 1. Tailings deposition to RL 277.85 m



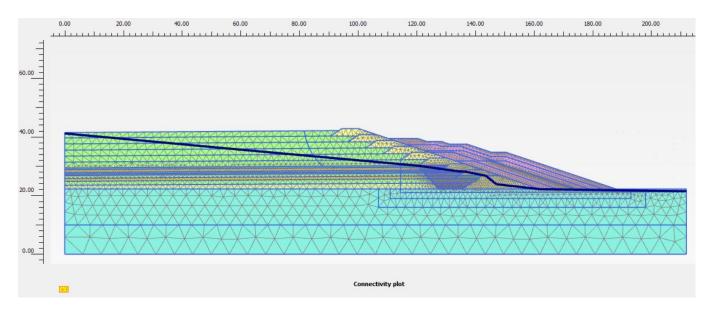
Stage 7: Construction of embankment raise 6 and buttress 2. Tailings deposition to RL 279.85 m



Stage 8: Construction of embankment raise 7 and buttress 3. Tailings deposition to RL 281.85 m

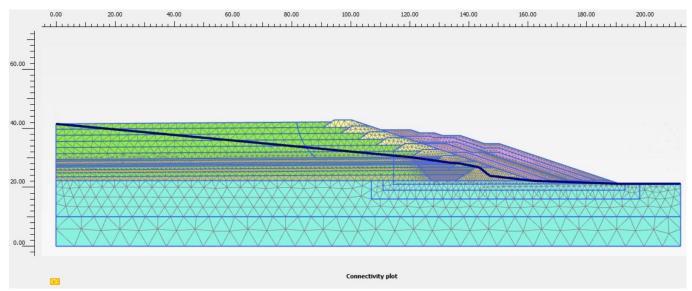


Stage 9: Construction of embankment raise 8 and buttress 4. Tailings deposition to RL 283.85 m

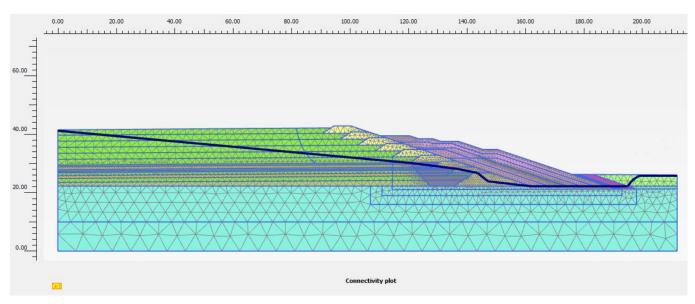


Stage 10: Construction of embankment raise 9 and buttress 5. Tailings deposition to RL 285.85 m

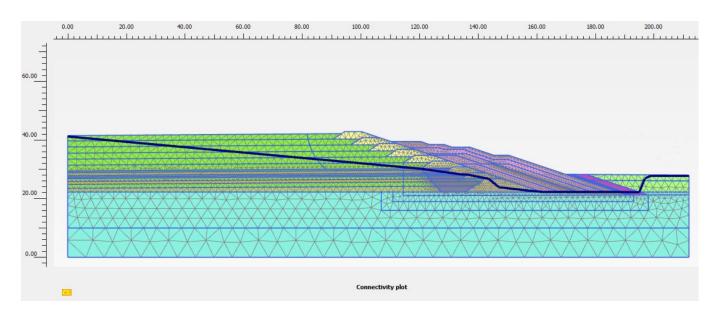




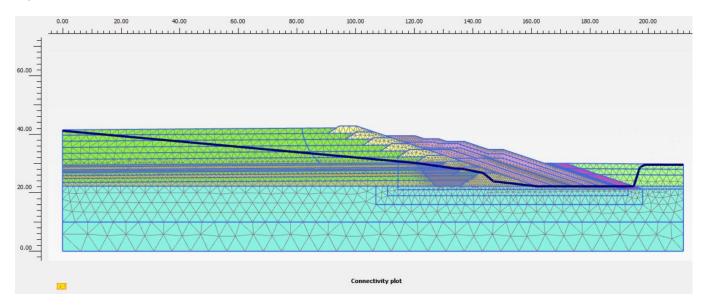
Stage 11: Blanket excavation at the RSF1 downstream toe. Consolidation phase of 10 days. Blanket construction.



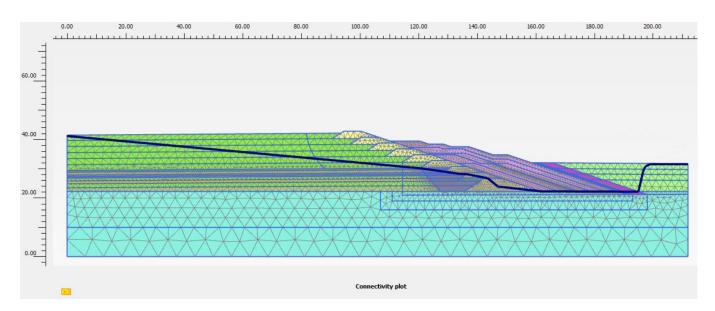
Stage 12: Drainage system construction. Tailings deposition to RL 269.85 m (RSF2 Starter Dam). Consolidation phase of 100 days.



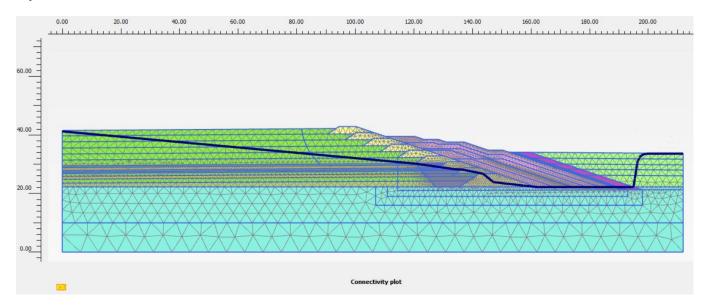
Stage 13: Drainage system construction. Tailings deposition to RL 271.85 m (RSF2 embankment raise 2). Consolidation phase of 100 days



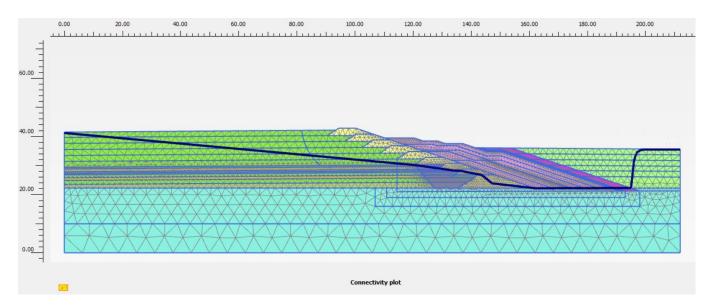
Stage 14: Drainage system construction. Tailings deposition to RL 273.85 m (RSF2 embankment raise 3). Consolidation phase of 100 days



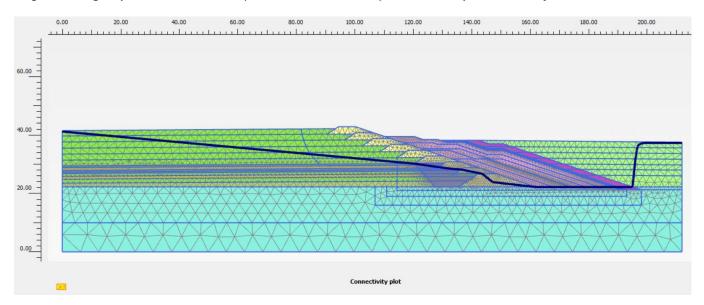
Stage 15: Drainage system construction. Tailings deposition to RL 275.85 m (RSF2 embankment raise 4). Consolidation phase of 100 days



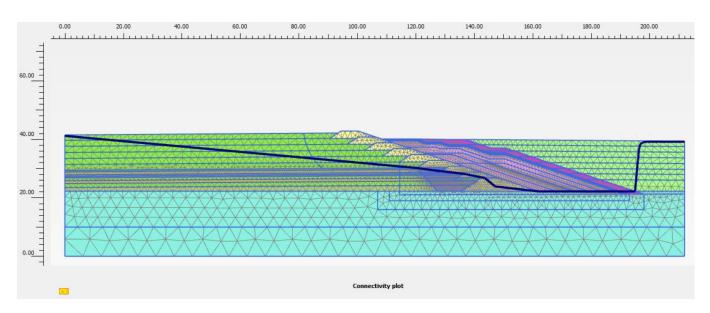
Stage 16: Tailings deposition to RL 277.85 m (RSF2 embankment raise 5). Consolidation phase of 100 days



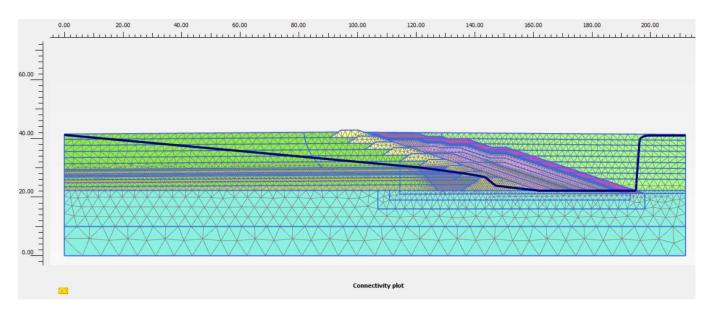
Stage 17: Tailings deposition to RL 279.85 m (RSF2 embankment raise 6). Consolidation phase of 100 days.



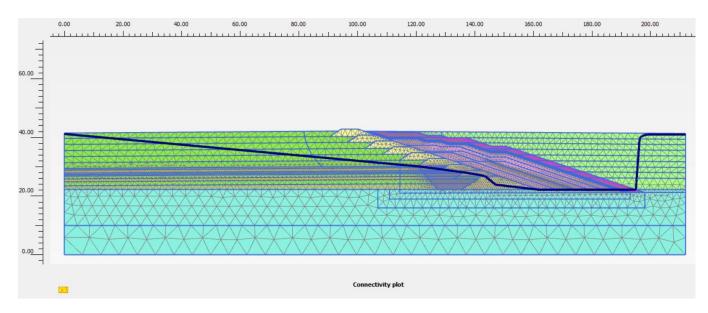
Stage 18: Tailings deposition to RL 279.85 m (RSF2 embankment raise 7). Consolidation phase of 300 days.



Stage 19: Tailings deposition to RL 279.85 m (RSF2 embankment raise 8). Consolidation phase of 300 days



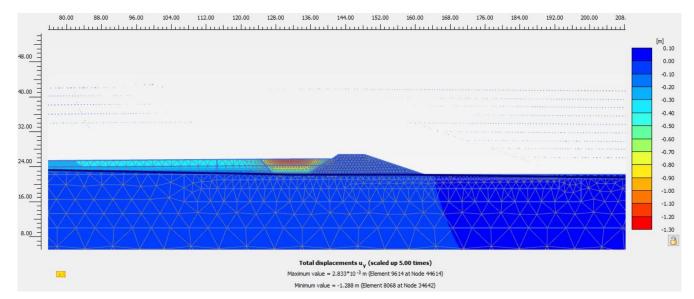
Stage 20: Tailings deposition to RL 279.85 m (RSF2 embankment raise 9). Consolidation phase of 300 days



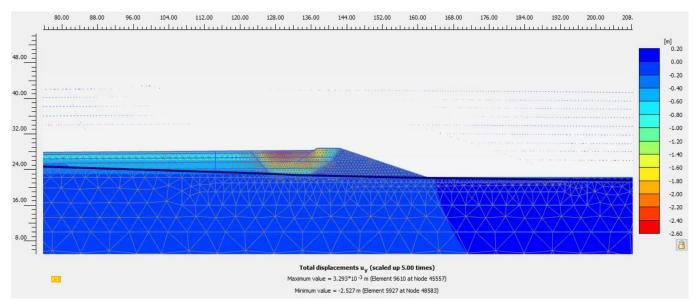
Stage 21: Long term consolidation phase until tailings pore pressure is less than 10kPa. This condition is reached after 3,584 days

Vertical Displacement

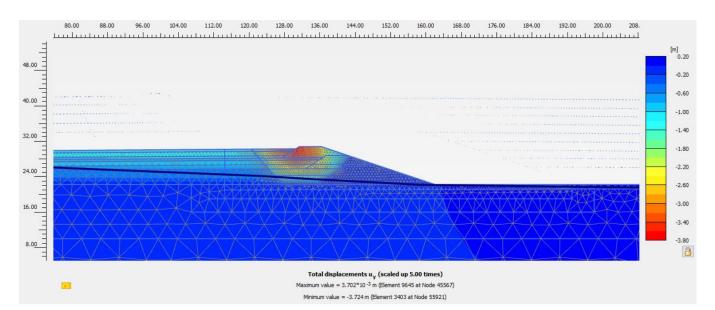
RSF 1



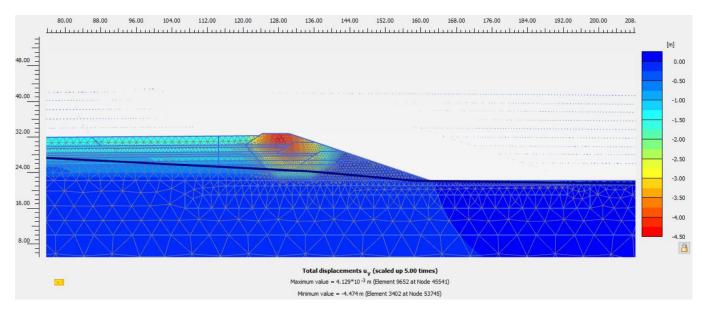
Vertical Displacements Stage 2, Construction of the Starter Dam. Tailings deposition to RL 269.85 m



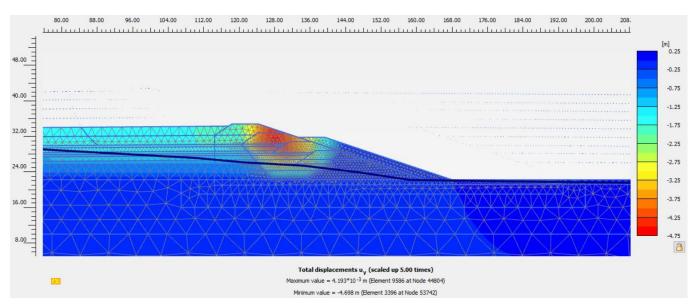
Vertical Displacements Stage 3, Construction of embankment raise 2. Tailings deposition to RL 271.85 m.



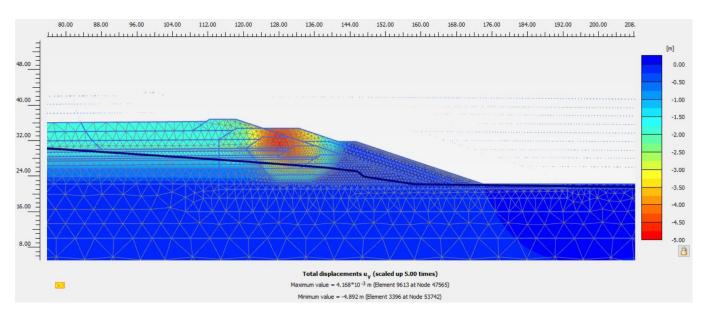
Vertical Displacements Stage 4, Construction of embankment raise 3. Tailings deposition to RL 273.85 m



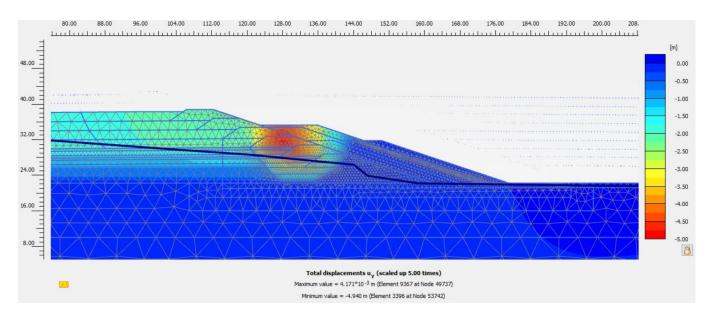
Vertical Displacements Stage 5, Construction of embankment raise 4. Tailings deposition to RL 275.85 m



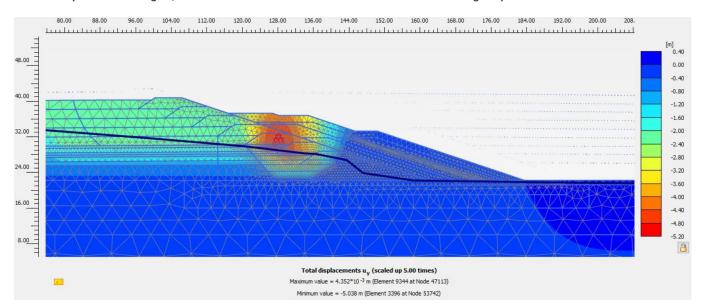
Vertical Displacements Stage 6, Construction of embankment raise 5 and buttress 1. Tailings deposition to RL 277.85 m



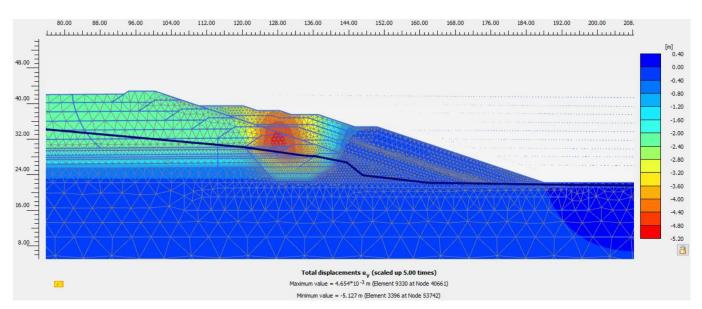
Vertical Displacements Stage 7, Construction of embankment raise 6 and buttress 2. Tailings deposition to RL 279.85 m



Vertical Displacements Stage 8, Construction of embankment raise 7 and buttress 3. Tailings deposition to RL 281.85 m

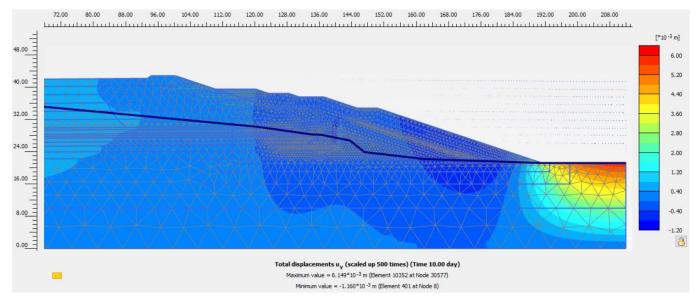


Vertical Displacements Stage 9, Construction of embankment raise 8 and buttress 4. Tailings deposition to RL 283.85 m

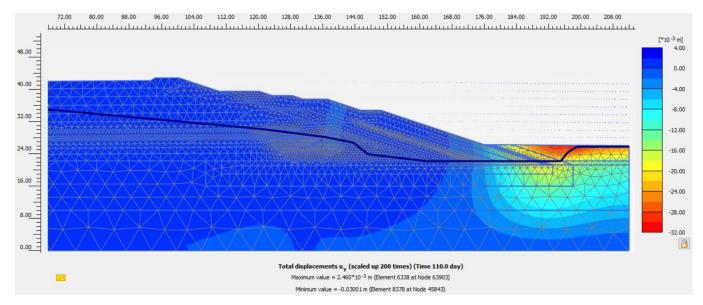


Vertical Displacements Stage 10, Construction of embankment raise 9 and buttress 5. Tailings deposition to RL 285.85 m

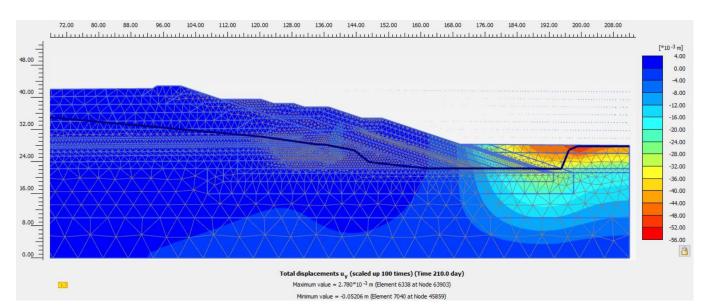
RSF 2



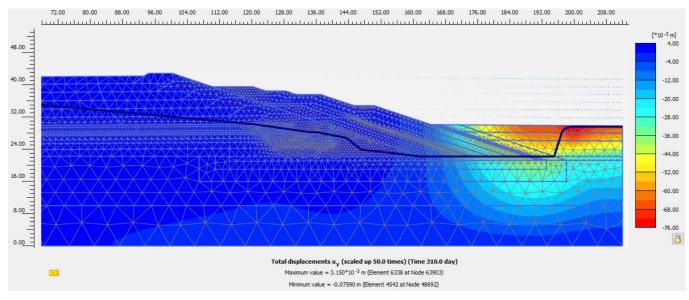
Vertical Displacements Stage 11, Blanket excavation at the RSF1 downstream toe. Consolidation phase of 10 days. Blanket construction.



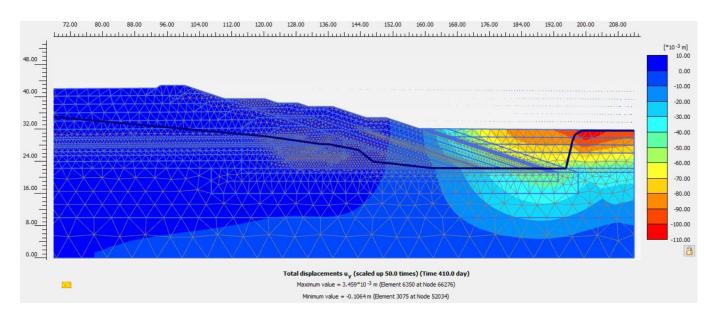
Vertical Displacements Stage 12, Drainage system construction. Tailings deposition to RL 269.85 m (RSF2 Starter Dam). Consolidation phase of 100 days.



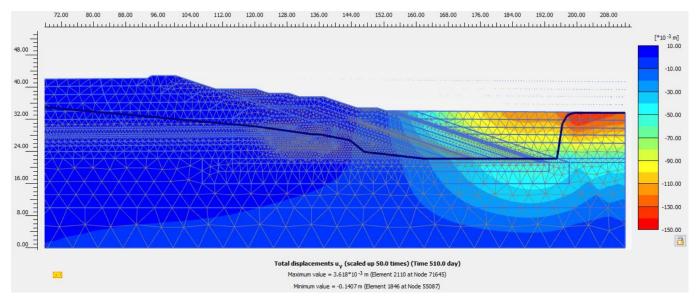
Vertical Displacements Stage 13, Drainage system construction. Tailings deposition to RL 271.85 m (RSF2 embankment raise 2). Consolidation phase of 100 days



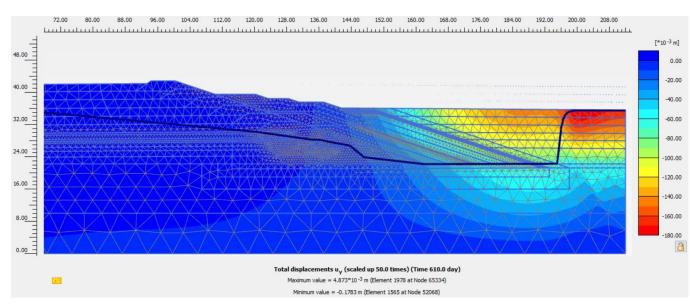
Vertical Displacements Stage 14, Drainage system construction. Tailings deposition to RL 273.85 m (RSF2 embankment raise 3). Consolidation phase of 100 days



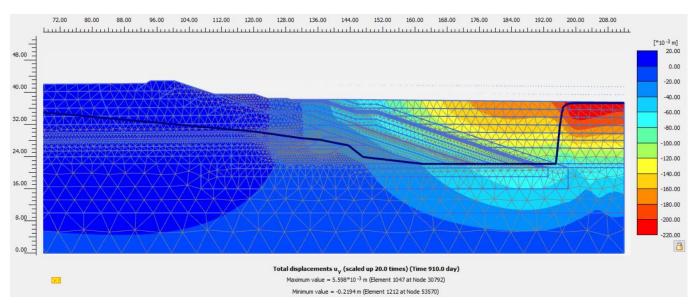
Vertical Displacements Stage 15, Drainage system construction. Tailings deposition to RL 275.85 m (RSF2 embankment raise 4). Consolidation phase of 100 days



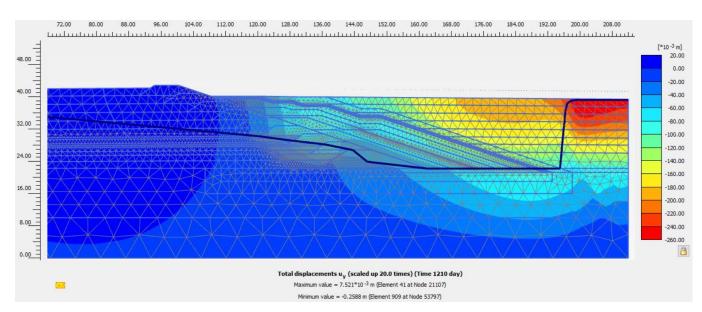
Vertical Displacements Stage 16, Tailings deposition to RL 277.85 m (RSF2 embankment raise 5). Consolidation phase of 100 days



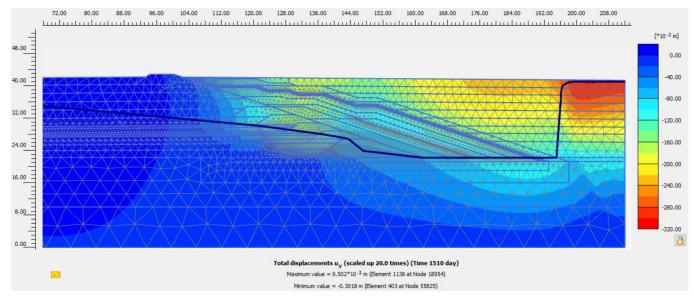
Vertical Displacements Stage 17, Tailings deposition to RL 279.85 m (RSF2 embankment raise 6). Consolidation phase of 100 days.



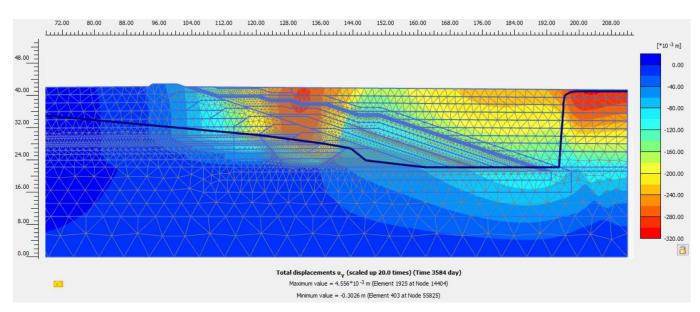
Vertical Displacements Stage 18, Tailings deposition to RL 279.85 m (RSF2 embankment raise 7). Consolidation phase of 300 days.



Vertical Displacements Stage 19, Tailings deposition to RL 279.85 m (RSF2 embankment raise 8). Consolidation phase of 300 days



Vertical Displacements Stage 20, Tailings deposition to RL 279.85 m (RSF2 embankment raise 9). Consolidation phase of 300 days



Vertical Displacements Stage 21, Long term consolidation phase until tailings pore pressure is less than 10kPa. This condition is reached after 3,584 days

Appendix F Safety in Design Risk Assessment



Risk Assessment

Notes: *Designs with significant quantities of dangerous goods may require detailed risk assessments under Dangerous Goods or Major Hazard legislation * Most industrial processes will require an industry specific assessment, e.g. HAZOP and/or Quantitative Risk Assessment for facilities that have chemical or high-pressure processes under Dangerous Goods or Major Hazard legislation.

Design Life Cycle:	Investigation and Design Setup, Construction and Commissioning Operation Maintenance Closure D		Date Prepared: 14/09/2021	Revisi	on No:		Rev A					
Job Name:	RSF 2 Stage 1 Raise	Detailed Design & Construction Services	Job No:	12545239	Client:	:	Tomingley Go	Gold Mine Pty Ltd (TGO)				
People in	volved in Risk Assessn	nent:	Rob Longey, Tom Ridgway, Gonzalo Suaz	o, Nicolas De La Maza								
Design Ref	Design Life Cycle Stage	Hazards What could cause injury or ill health, damage to property or damage to the environment	Risk What could go wrong and what might happen as a result	Existing Control Measures	consequence	ritial Ri	sk Rating ^{Guite} Buite	Potential Control Measures (Consider Hierarchy of Control - Elimination, Substitution, Isolation, Engineering Controls, Administrative Controls, PPE)	Re: Consequence	Likelihood	Risk Rating ^{Guite} Ysy X	Person Responsible
	Investigation and	Inaccurate assumptions relating to hydraulic properties of the embankment, foundations and tailings, seepage behaviour and phreatic surface	Inadequate embankment stability or dam failure as a result of higher than anticipated phreatic surface.	RSF will be constructed using sandy clay or clayey sand materials sourced from the foundation excavation. The embankment is Stage 1 and is constructed entirely on natural foundations. Tailings beach is designed to keep the decant away from the from the embankment to maintain a minimum phreatic surface through the embankment. The decant pond should be maintained as low as possible using the submersible pump. OMS of the RSF2 will be prepared prior to construction detailing minimum operating requirements. Piezometers to be installed over the storage liner to monitor pore pressure/ phreatic surface through RSF.	E	2	Significant	 Monitor piezometers to observe the phreatic surface in the dam and visual inspection to enable early detection of unusual dam behaviour. Ongoing review of design assumptions using "observational approach" during detailed design of the embankment including provision for additional stabilisation or drainage if required to maintain acceptable embankment stability. 	E	1	Moderate	TGO
	Investigation and Design	Dam Failure Consequence Category (DFCC) and Dam Spill Consequence Category (DSCC) are different compared to the RSF1 due to the location of critical infrastructure.	Incorrect design assumptions	A detailed dam break assessment has been undertaken for the consequence category assessment. Design criteria has been based on the results of the consequence category assessment.	с	3	Moderate	- Downstream development and expansions of the mining infrastructure into the identified inundation zones are to be reviewed as to their necessity as these are expected to increase the consequences of failure.	с	1	Low	TGO
	Investigation and Design	Inaccurate assumptions relating to foundation, tailings and embankment material strength properties and	Embankment failure as a result of incorrect design assumptions relating to loading conditions or material properties relating to foundation conditions (i.e. low strength foundation zone, liquefaction potential).	Material properties have been reviewed based on the geotechnical investigations on the existing RSF tailings and the RSF2 foundations. Loading conditions assessed in accordance with ANCOLD Guidelines on Tailings Dams (2019) and Guidelines for Design of Dams and Appurtenant Structures for Earthquake (2019) as appropriate for RSF design including post-seismic case using post-liquefied strength for tailings and 15% strength reduction for clay and foundation under the phreatic surface. Lower bound post liquefaction parameters adopted.	E	2	Significant	 Undertake CPT tests through the initial tailings prior to future raises to verify tailings strength. Monitor piezometer readings to confirm that phreatic surface is within or below design. 	E	1	Moderate	TGO
	Setup, Construction	TGO plans to construct Stage 1 in January 2022 but approval/form is yet to be submitted leading to time delays in construction		Stage 8 Raise will be under construction prior to this allowing for sufficient storage in RSF1 should construction be delayed.	D	2	Moderate	 Report to be submitted asap. Reduce production if required. Ongoing review of project schedule. Proactive management to ensure tasks are completed on time. Time construction to maximise production during summer weather to minimise contractor delays. Initiate RSF1 Stage 9 if necessary to achieve additional storage. Stage RSF2 construction to allow for deposition in Cell 1 while Cell 2 is finishing construction. 	с	2	Low	TGO



Design Life Cycle:	Investigation and Design	Setup, Construction and Commissioning	Operation	Maintenance		Clo	osure	Date Prepared: 14/09/2021	Revisio	on No:		Rev A
Job Name:	RSF 2 Stage 1 Raise	Detailed Design & Construction Services	Job No:	12545239	Client:		Tomingley Gold Mine Pty Ltd (TGO)					
	volved in Risk Assess	nent:	Rob Longey, Tom Ridgway, Gonzalo Sua	zo, Nicolas De La Maza								
	Design Life Cycle	Hazards	Risk		Ini	Initial Risk Rating		Potential Control Measures	Res	sidual	Risk Rating	
Design Ref	Stage	What could cause injury or ill health, damage to property or damage to the environment	What could go wrong and what might happen as a result	Existing Control Measures	Consequence	Likelihood	Risk Rating	(Consider Hierarchy of Control - Elimination, Substitution, Isolation, Engineering Controls, Administrative Controls, PPE)	Consequence	Likelihood	Risk Rating	Person Responsible
	Setup, Construction and Commissioning	Insufficient availability of suitable embankment construction materials	enable successful construction of embankment works results in compromised embankment quality,	Design and material specification has been developed to enable use of as wide a range of material as possible. Material to be sourced form the RSF2 footprint and if necessary from the same borrow area (WRE1) as previous RSF1 raises.	с	2	Low	 Future RSF2 raise construction to be a consideration in mine planning. Future RSF2 raises to use waste rock on the downstream shell. 	с	1	Low	TGO
	Setup, Construction and Commissioning	Borrow materials properties differ from those anticipated (do not meet permeability requirements)	Higher permeability may cause seepage reporting to the external environment	Materials are to be borrowed from the storage area as per the geotechnical investigations, which are expected to have suitable properties based on geotechnical investigations. Testing / compaction regime outlined in specification to accommodate permeability requirements. Tailings against embankment reduces the risk of seepage, sufficient beach freeboard maintained.	с	2	Low	- Additional testing and review of borrow materials for each construction raise.	с	1	Low	TGO
	Setup, Construction and Commissioning	Instability of excavations for the embankment and borrow areas	Slope failure of excavation results in harm to staff and/or damage to plant and equipment.	Minimum batter slopes for excavations provided in Technical Specification.	с	2	Low	 Visual assessment of the stability of batter slopes in borrow, ensure a safety distance to the slopes. 	с	1	Low	TGO
	Setup, Construction and Commissioning	Wetter or drier than anticipated conditions	RSF embankment more difficult / expensive to place or cannot be placed, results in production delay/stoppage.	Reviewed seasonal rainfall variation. Contractor experience from previous raises.	D	3	Significant	 Specification can potentially be relaxed to allow material wet of optimum following additional testing to prove required strength and permeability requirements met. Contractor to have water cart to condition material and borrow areas. 	с	2	Low	TGO
	Setup, Construction and Commissioning	Limited availability of materials eg pipes, valves and fittings	Delay to completion of cell.	Indicative construction schedule has been established. Early supplier involvement is recommended to confirm local supply.	в	2	Negligible	 Appropriate measures should be in place in the contracts and or site spares should damage occur 	в	1	Negligible	тgo
	Operation	Rising piezometric pressures	Instability of the RSF as a result of rising phreatic surface.	Vibrating wire piezometers to be installed over the clay liner to monitor pore pressures within the embankments.	D	2	Moderate	 Monitoring of the piezometers after the construction to verify pore pressures and adequate factor of safety. Undertake regular routine and intermediate surveillance inspections during operation to keep decant pond as low as practicable. 	D	1	Moderate	TGO
	Operation	Poor tailings management practices		Tailings management addressed as part of design documentation. Regular reviews of tailings management during routine operator and periodic engineering inspections. Ongoing review of design flood storage allowance.	с	3	Moderate	 Install Instrumentation (i.e. piezometers, movement monitoring, tailings beach indicators) to enable monitoring. Undertake tailings reconciliation to check tailings settled density and review filling schedule and design flood storage allowance. Undertake regular routine and intermediate surveillance inspections during operation to keep decant pond as low as practicable. 	С	2	Low	TGO

Design Life Cycle:	Investigation and Design	Setup, Construction and Commissioning	Operation	Maintenance		Clo	osure	Date Prepared: 14/09/2021	Revisi	on No:		Rev A	
Job Name:	RSF 2 Stage 1 Raise	e Detailed Design & Construction Services	Job No:	12545239	Client: Tomingley Gold Mine Pty Ltd (TGO)								
	volved in Risk Assess	ment:	Rob Longey, Tom Ridgway, Gonzalo Sua	zo, Nicolas De La Maza									
Design	Design Life Cycle	Hazards What could cause injury or ill health,	Risk		ln g	itial Ri	isk Rating ₽	Potential Control Measures	Residual Risk		Risk Rating	Person	
Ref	Stage	damage to property or damage to the environment	What could go wrong and what might happen as a result	Existing Control Measures	Consequer	Likelihoo	Risk Ratir	(Consider Hierarchy of Control - Elimination, Substitution, Isolation, Engineering Controls, Administrative Controls, PPE)	Consequer	Likelihoo	Risk Ratir	Responsible	
	Operation	RSF fills quicker than anticipated		Design based on TGO supplied production and forecasts.	с	2	Low	 Monitor RSF fill rate during operation. Tailings survey and density reconciliation. 	с	1	Low	TGO	
	Operation	High decant pond level	Potential for piping failure of RSF embankment due to increased hydraulic gradients. Reduced flood storage, increase risk of overtopping.	RSD can store 150ML of water. RSF constructed on natural foundation improving stability.	D	2	Moderate	 Undertake regular routine and intermediate surveillance inspections during operation to keep decant pond as low as practicable. Regular reviews of piezometer data to validate assumptions made in determining phreatic surface in stability models. Beach markers to check pond extents comply with design 	D	1	Moderate	тдо	
	Operation	Decant pump failure	Plant. Production must be slowed /halted.	RSF has the capacity to store 568ML (in both Cell 1 and Cell 2) of water up to embankment crest without overtopping the spillway. Water can be sourced from open pits or water dams on site. Can use decant pump from alternate cell if required.	В	3	Low	 Undertake regular routine inspection to ensure decant pond is kept as low as practicable. Operate additional pumps to drawdown pond as soon as possible post large storm event. Regular servicing of pumps. Consider backup options for pump and power supply 	в	2	Negligible	TGO	
	Operation	Severe Earthquake	Foundation liquefaction or cyclic softening. Loss of strength of embankment material. Deformation or failure of embankment leads to loss of tailings/decant water.	Embankment materials to be compacted to design specification hence loss of strength is not expected above phreatic surface. Embankment has been designed based on worst loading case which is the post-seismic case using post seismic liquefaction strength under SEE 1:10,000 AEP with PGA of 0.18g.	D	1	Moderate		D	1	Moderate	TGO	
	Operation	Extreme flood event		RSF has been designed to store PMP rainfall event within the RSF, emergency spillway capable of passing 1:1,000 AEP rainfall event to be constructed in Cell 1 to route flow safely. External excess water storage of 150ML at RSD.	E	2	Significant	 Undertake regular routine inspection to ensure decant pond is kept as low as practicable. OMS & DSEP regularly reviewed and updated if required. 	E	1	Moderate	TGO	
	Operation	Tailings delivery and decant pipelines	Failure of tailings delivery pipeline or decant pipeline results in erosion of embankment /contamination of site requiring remedial works	Tails pipe bunds. Natural foundation soils are low permeability to minimise leakage to ground. Routine shift inspection. Flow meter alarms.	с	2	Low	-	с	2	Low	TGO	
	Operation	Vehicle driving on dam crest /along tailings pipelines		Safety bund on downstream side of crest road and ramps. Suitable crest width designed for LV access. Tailings pipe on upstream side acts as partial deflector upstream batter.	D	3	Significant	 Speed / warning signage on approach to RSF crests. Temporary windrows using in construction, pending contractor methodology. 	D	2	Moderate	TGO	
	Operation	Poor operational management	Dam safety issues are not observed leading to unsafe conditions	Instrumentation has been installed for monitoring purposes. Daily monitoring by appropriately trained staff. OMS & DSEP regularly reviewed and updated if required. Annual inspections undertaken by consultant dams' engineer	с	2	Low	-	с	2	Low	TGO	
	Operation	Dam failure, or emergency situation	Inability to adequately respond to avoid risk to personnel or environment due to dam failure flood / release of tailings	Appropriate design to mitigate risk of failure. OMS and DSEP updated annually. Personnel responsible for monitoring/surveillance have been trained in RSF Surveillance.	E	2	Significant	 New RSF operators to undertake RSF Training Course. Dam break inundation mapping to be updated to account for RSF / downstream topography changes / mill changes. Run RSF Emergency Scenario simulation to test DSEP and train those responsible to act in case of emergency. 	E	1	Moderate	тgo	

	Investigation and Design	Setup, Construction and Commissioning	Operation	Maintenance		Closure D		re Date Prepared: 14/09/2021		Date Prepared: 14/09/2021 Revision No		on No:		Rev A
Job Name: RSF 2 Stage 1 Raise Detailed Design & Construction Services		Job No:		Client:	t: Tomingley Gold Mine Pty Ltd (TGO)									
People in	volved in Risk Assessn	nent:	Rob Longey, Tom Ridgway, Gonzalo Suaz	zo, Nicolas De La Maza										
Design Ref	Design Life Cycle Stage	Hazards What could cause injury or ill health, damage to property or damage to the environment	Risk What could go wrong and what might happen as a result	Existing Control Measures	Consequence	itial Ri	sk Rating ^{Guite} Buite	Potential Control Measures (Consider Hierarchy of Control - Elimination, Substitution, Isolation, Engineering Controls, Administrative Controls, PPE)	Re: Consequence	sidual F Likelihood	Risk Rating	Person Responsible		
	Closure			GHD concept developed as part of detailed design of Stage 1 of RSF2.	С	3	Moderate	- RSF2 Closure Plan to be prepared prior to the construction of RSF2 Stage 1 outlining the preliminary closure plan for the two RSF's outlining rehabilitation requirements and future investigation and trial requirements.	с	2	Low	тgo		

Appendix G RSF2 Technical Specification



RSF2 Stage 1 Technical Specification

Tomingley Gold Operations Pty Ltd

19 January 2022

→ The Power of Commitment



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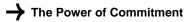
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1. Embankment Construction

1.1 Material Sourcing Requirements

The material requirements for each zone shall be as set out below, unless otherwise determined by the Principal.

Zone 1:	Low to high plasticity sandy clay (CH), or clayey sand (SC) sourced from excavation of the RSF2 footprint.
General Fill:	Highly to extremely weathered rock sourced from excavation of the RSF2 footprint with additional material from the Waste Rock Emplacement 1 (WRE1)
Zone 2:	Fine to coarse grained sand.
Zone 3 / Rip-Rap:	Rockfill. Slightly weathered to fresh rock sourced from fresh mine waste.

1.2 Tolerances

The dam embankment shall generally be constructed to the lines, grades and dimensions as shown on the drawings and derived from the set-out information. However, at any time prior to, or during construction, the Principal may vary the division lines between portions of the dam embankment and also the position of the outside faces of the dam embankment.

The upstream and downstream faces of the dam embankment shall, when completed, present an appearance of uniform texture and shall be generally planar.

The following tolerances limitations in the embankment shall be adhered to.

ltem	Towards axis of dam	Away from axis of dam			
Crest	0 mm	+ 250 mm			
Downstream slope	0 mm	+ 500 mm			
Upstream slope	0 mm	+ 500 mm			
Division between core and filter zone	0 mm	400 mm			
Width of Filter Zones	- 0 mm to + 400 mm				
Width of Rip Rap / Zone 3	- 0 mm to + 200 mm perpendicular to slope				
Tolerance on Vertical Thickness					
	Minus	Plus			
Compacted layer thickness	50 mm	50 mm			
Zone 2	0 mm	200 mm			
Transition Zone	0 mm	200 mm			
Top of core and crest elevation	0 mm	150 mm			

 Table 1
 Embankment Tolerances

1.3 Geochemical management of borrow materials

The Principal has made provision for geochemical testing of borrow materials. All borrow materials and excavated material shall be tested for Net Acid Production Potential (NAPP) and Net Acid Generation (NAG) at a frequency of 1 set of tests per 10,000 m³ of material, and results provided by the Principal. Only materials designated as non-acid forming shall be permitted for use in the dam embankment. The Contractor must make notification to the Geotechnical Engineer at quantity intervals during excavation for testing.

2. Material Definition & Requirements

2.1 Zone 1 Requirements

2.1.1 Required Properties

Zone 1 shall conform to the following soil classifications unless otherwise approved by the Principal (in accordance with AS 1726 - 2017 Geotechnical Site Investigations):

- low to high plasticity sandy CLAY or clayey SAND (with other elements whilst conforming to the table below)

The Zone 1 materials, for the construction of the embankments, shall be naturally occurring, predominantly clayey materials free of organic and other unsuitable materials. The materials shall be well graded and have a low permeability when compacted.

Property	Requirement
Maximum thickness of layer after compaction	150 mm
Size of largest particle	Not greater than 75 mm
Minimum % by weight passing 0.075 mm AS 1152 sieve	30
Minimum Plasticity Index (AS 1289.3.3.1)	10
Minimum plasticity product (PI x % by weight passing 0.075mm)	300
Maximum Plasticity Index (AS 1289.3.3.1)	60

 Table 2
 Required properties of Zone 1 after compaction

2.1.2 Moisture Content and Conditioning

Moisture conditioning shall occur in the borrow area where practical to bring the materials within the specified moisture content range. After placement within the embankment, only minor moisture adjustment is anticipated to maintain the moisture content within specification. A proposed methodology for moisture conditioning shall be submitted for approval with the borrow material approvals **Hold Point**.

The material in each layer of zone 1 material shall have a uniform moisture content, during and after compaction, with *moisture content OMC between plus 1% (Dry) and minus 3% (Wet)* (OMC = Optimum Moisture Content, Standard Compaction AS 1289). There is no maximum moisture content specified and the clay will be accepted provided the material has been compacted to the specified density. Material that fails to meet the above requirements will require corrective measures to achieve a uniform moisture content within specification or removal from the embankment.

2.1.3 Placement and Compaction Requirements

All clay material required for the core of the embankment shall be obtained from the sources designated as borrow areas or from other excavations approved by the Principal.

Each layer of Zone 1 material shall be compacted with vibrating pad foot rollers or tamping foot compactors until the field density determined in accordance with AS 1289.5.8.1 (Nuclear Densometer) meets the criterion that the **Hilf density ratio (RHD) is in excess of 98%** when tested in accordance with AS 1289.5.7.1 **using standard compactive effort** in accordance with AS 1289.5.1.1.

The Contractor's operations in handling and spreading Zone 1 material on the embankment shall result in an acceptable gradation of the materials when compacted in the fill. Should stones, cobbles or rock fragments be found in otherwise approved Zone 1 materials, the maximum dimension of the stone, cobble or rock fragment after compaction would be expected not to exceed 150 mm. Cobbles exceeding 150 mm are expected to be removed by the Contractor either at the Site of the excavation or after being transported to the dam embankment but before

the materials in Zone 1 are compacted. The Contractor will be required to develop a methodology for removal of cobbles, which may include the following removal techniques:

- Use of a Grizzly Bucket during borrow excavations to screen out stones, cobbles, and rock fragments;
- Selective borrowing as determined suitable on site; and
- Supervision during placement to remove cobbles prior to compaction.

Materials shall be placed in such a manner that the distribution and gradation of the materials throughout Zone 1 shall be free from lenses, pockets, streaks, or layers of material differing substantially in texture or gradation from the surrounding material within the zone.

Maximum layer thickness shall be and not more than 150 mm in thickness after compaction unless adequate compaction using thicker layers can be demonstrated with a roller trial to the satisfaction of the Principal.

The surfaces of any layer of Zone 1 shall be suitably textured to the Principals satisfaction to ensure bonding of the subsequent layer. Surfaces of Zone 1 which are ruptured or uneven after compaction shall be repaired and relevelled before placing the next layer of material.

2.1.4 Foundation Placement

Within 0.5 metre measured normally to the embankment abutments and foundation, the Zone 1 material shall be from the selected finer, more plastic material in the sources for Zone 1 material. The material shall be deposited in horizontal layers not more than 150 mm in thickness when compacted. The material shall be compacted by approved heavy pneumatic tyred equipment or other means in such a manner that it is forced into all the irregularities of the abutment and foundation surfaces.

2.2 General Fill

General Fill material shall be sourced from highly to completely weathered rock from necessary excavations for the RSF2 impoundment and / or external borrow areas. It is envisaged that this material will typically behave as a fine-grained cohesive soil, with gravel and some larger cobbles of less weathered rock. The largest rock size will be that which can be adequately incorporated in a lift, usually two thirds of the lift height.

Where General Fill material has greater than 80% of particles passing the 37.5 mm sieve, it shall be compacted to **95% of standard maximum dry density** at a moisture content **OMC between plus 1% (Dry) and minus 3% (Wet).** Maximum lift height after compaction shall be 300 mm.

Where General Fill material has less than 80% of particles passing the 37.5 mm sieve, the compaction requirements shall be determined by test rolling in accordance with Clause **Error! Reference source not found.**. Maximum lift height after compaction shall be 300 mm unless it can be demonstrated that adequate compaction is achieved with larger lifts.

General Fill material shall be selectively placed to ensure that more finely graded material is placed adjacent to the core and the larger materials are directed toward the outer edges of the embankment to give a gentle transition across the width of the zone. In no case will zones of rock be allowed adjacent to the Zone 1 core.

Compaction equipment shall be that most suitable for compaction of the material as directed by the Principal.

2.3 Zone 2

Durability of source rock for crushed Zone 2 coarse drainage sand material shall be tested in a manner similar to that described in AS 1141 and shall be rejected if the loss in the Los Angeles abrasion test exceeds 40 per cent by weight at 500 revolutions or if the portion retained on a 0.300 mm AS sieve, when subjected to 5 cycles of the sodium sulphate test for soundness, in accordance with AS 1141, shows a weighted loss of more than 10 per cent by mass.

The filters shall be placed to the lines and grades shown on the Drawings. The placement method shall consider the need to avoid contamination with the core or downstream zone materials, and ensuring the minimum thickness is obtained. The placement method shall be subject to approval after a filter placement trial to be carried out in the presence of the Principal (**HOLD POINT**).

The specified width of the vertical filter shall be the <u>minimum</u> thickness, where subsequent lifts are offset, the minimum thickness is the dimension of the overlap between lifts.

At the time of placement the filter materials shall be damp or moist but shall not contain excessive free water. A smooth drum vibratory roller or tamping plate shall be used for compacting the material in a manner approved by the Principal. Any contamination of the filter materials (such as roots, organics) shall be immediately removed.

If, during or after placement, any filter material has become contaminated by earthfill, filter or other objectionable material from the passage of construction machinery, or from surface run-off, or by any other means, the contaminated material shall be entirely removed from the dam embankment.

Blending of filter material on the dam embankment in order to produce the required grading will not be permitted.

A sample of Zone 2 material shall be collected from the placed filter zone for grain size analysis in accordance with Section 2.6. Compaction testing of the vertical filter shall be carried out on early placed materials to confirm the placement method is acceptable, the target density ratio shall be within the range of 60% to 80%. The Principal may also direct DCP testing of the filter zone to confirm uniform density throughout the lift height.

The methodology of sampling and testing shall be outlined in the *Contractor's Quality Plan* according to the requirements above.. The Zone 2 material grading specifications shall comply with Appendix A and will be supplied by the Principal.

2.4 Zone 3 / Rip Rap Requirements

Rip Rap shall be obtained from fresh mine waste rock complying with the Fresh to Slightly Weathered rock (F, SW) classification (in accordance with AS 1726 - 1993 Geotechnical Site Investigations) provided by the Principal.

Rip rap will generally comprise of well graded, angular and high to extremely high strength rock for durability. Unless otherwise approved, the rip rap size shall comply with the drawings and shall be well graded from the minimum to maximum rock size. Rip Rap material will be provided by the Principal to the specifications defined in this report..

Riprap for locations as described in the drawings shall have a particle size as shown on the drawings. Where only a D_{50} size is shown, the riprap shall be graded such that D_{100} is approximately 1.5 times D_{50} .

2.5 Filter rock

Filter rock provided by the Principal will be placed around the decant tower. The filter rock will consist of clean, fresh aggregate, with an average particle size of 60-150 mm. Suitability of this material shall be as directed by the Principal. Compaction will be taken with care to avoid damage to pipework and geofabric.

Filter rock may consist of run of mine material (if an appropriate material is produced) or alternatively be imported from an off-site quarry.

2.6 Embankment Construction Quality Control

2.6.1 Testing Requirements

The Principal shall engage a laboratory registered with the National Association of Testing Authorities for the particular tests specified to undertake the testing of embankment materials for the project. The testing capacity of the laboratory shall be sufficient to enable testing of all materials from the source to the fill in accordance with the Contractor's construction programme.

Test procedures and minimum testing frequency shall be in accordance with Table 3 and the requirements of this specification. Notwithstanding these minimum test requirements, the Principal may make additional tests as considered necessary to control the quality of the materials.

The results of all moisture and compaction control testing shall be reviewed and approved by the Contractor following appropriate Quality Assurance processes before work on the next layer of the embankment material can commence (**WITNESS POINT**).

The Principal may request additional random tests on the embankment to verify that quality assurance requirements are carried out adequately and consistently.

2.6.2 Lot Selection

A lot is defined as an area of work which is essentially homogeneous in: material type, moisture condition, rolling effort and compaction technique and which is to be used for the assessment of the density ratio. Where a production unit does not satisfy the definition of a lot it shall be divided into smaller units, each of which shall conform with the definition.

Maximum lot sizes are defined as follows:

Zone 1 - Maximum lot size shall comprise the lesser of:

- 2,000 m² of material placed
- 500 m³ of material placed
- One day's production
- 2 lifts.

Zone 3 – Maximum lot size shall comprise the lesser of:

- 3,000 m² of material placed
- 2,000 m³ of material placed
- One day's production
- 2 lifts.

Table 3 Quality Control Testing Schedule

Zone	Process	Test Procedure for Quality Compliance	Minimum Testing Frequency
	Particle Size Distribution with Hydrometer	AS 1289.3.6.3	1 test per 5,000 m ³
	Plasticity Index	AS 1289.3.3.1	1 test per 5,000 m ³
	Liquid Limit	AS 1289.3.1.1	1 test per 5,000 m ³
	Field wet density	AS 1289.5.3.1, 1289.5.3.2 or 1289.5.3.5	Every 300 mm, 1 test every 1,000 m ³
	Field moisture content	AS 1289.2.1.1	Every 300 mm, 1 test every 1,000 m ³
Zone 1	Moisture variation	AS 1289.5.7.1 using standard compactive effort in accordance with AS 1289.5.1.1	Every 300 mm, 1 test every 1,000 m ³
	Hilf Density Ratio	AS 1289.5.7.1 using standard compactive effort in accordance with AS 1289.5.1.1 and field wet density in accordance with AS 1289.5.8.1	Every 300 mm, 1 test every 1,000 m ³
	Emerson Class Testing	AS 1289.3.8.1 using Process Water & Fresh Water	1 test per 5,000 m ³
	Triaxial Testing	AS 1289.6.4.2	1 test per 10,000 m ³
	Permeability	AS 1289.6.7.1	1 test per 5,000 m ³
General	Particle Size Distribution	AS 1289.3.6.3	1 test per 10,000 m ³
Earth-fill	Plasticity Index	AS 1289.3.3.1	1 test per 10,000 m ³
	Liquid Limit	AS 1289.3.1.1	1 test per 10,000 m ³
	Field wet density	AS 1289.5.3.1, 1289.5.3.2 or 1289.5.3.5	1 test every 5,000 m ³

Zone	Process	Test Procedure for Quality Compliance	Minimum Testing Frequency	
	Field moisture content	AS 1289.2.1.1	1 test every 5,000 m ³	
	Moisture variation	AS 1289.5.7.1 using standard compactive effort in accordance with AS 1289.5.1.1	1 test every 5,000 m ³	
Hilf Density Ratio		AS 1289.5.7.1 using standard compactive effort in accordance with AS 1289.5.1.1 and field wet density in accordance with AS 1289.5.8.1	1 test every 5,000 m ³	
	Emerson Class Testing	AS 1289.3.8.1 using Process Water & Fresh Water	1 test per 10,000 m ³	
	Triaxial Testing	AS 1289.6.4.2	1 test per 20,000 m ³	
	Permeability	AS 1289.6.7.1	1 test per 10,000 m ³	
Engineered Fill	Hilf Density Ratio or Equivalent Compaction	AS 1289.5.7.1 using standard compactive effort in accordance with AS 1289.5.1.1 and field wet density in accordance with AS 1289.5.8.1	Every 300 mm, 1 test every 1,000 m ³	
	Particle Size Distribution	AS 1289.3.6.3	1 test per 2,000 m ³	
	Triaxial Testing	AS 1289.6.4.2	1 test per 5,000 m ³	
Tests to be ca	arried out from stockpile at qu	arry or commercial supplier:		
Filter Zone 2	Particle Size Distribution (wet sieve method)	AS 1289.3.6.1 Note maximum 3% fines at supplier	3 tests prior to approval then 1 test per 500 m ³	
	Los Angeles Abrasion	AS 1141	1 tests prior to approval	
	Sodium Sulphate Test	AS 1141	1 tests prior to approval	
	Organic Content	AS 1141.34	1 tests prior to approval	
	Tests to be carried out on samples taken after placement:			
	Particle Size Distribution (wet sieve method)	AS 1289.3.6.1	1 test per 250 m ³	
	Density Index	AS 1289.5.6.1	2-3 tests as directed by the Principal	

2.6.3 Testing Results

All NATA certified test results will be supplied to the Contractor in a timely manner, whereas the Contractor will manifest these into a control report.

In addition to satisfying the specific requirements detailed in other parts of this specification, the report shall include;

- Location of all tests in terms of chainage, transverse location, elevation RL and easting and northing of the test.
- Individual test values of moisture content (both nuclear and oven check) and dry density at each test location.
- Reference density for each test location
- Hilf density index and moisture variation.

2.6.4 Acceptance Criteria

All lots shall be compacted to a dry density ratio or density index equal to or greater than the specified value and moisture content within the specified range. In the event that a test result for a lot is below the specified value, remedial work shall be conducted over the whole lot.

REMEDIAL WORK: Any unsatisfactory filling is to be made good and may include removal or reworking of subsequently placed or overlying material which conforms with the Specification requirements.

Typical remedial action required for unacceptable fill placement will include but may not be limited to the following:

 Table 4
 Remedial actions for compacted fill

Category	Hilf Density Ratio Result	Moisture Result	Remedial Action 1
A	Fail by less than 1%	Pass	Re-compact (number of passes to be specified by the Principal)
В	Fail by 1% or more	Pass	Rip, re-compact and re-test
С	Pass	Fail, but no more than 2.0% dry of OMC	Rip, re-water, re-compact and re-test
D	Fail	Fail	Remove fill, replace, compact and re-test

Note: ¹ Should the Principal deem the depth of insufficiently compacted material to be greater than can be effectively compacted from the surface, material shall be removed to a depth at which compaction is satisfactory and replaced and compacted in layers.

2.6.5 Further Potential Actions

The primary strategy will be to utilise borrow materials that can be moisture conditioned to meet the compaction specification, and not be placed "over-wet".

If, however, there are repeated occurrences of moisture content above the specification limits, and remedial works to address the repeated failures become overly cumbersome and restrict construction progress, additional supporting strategies may be required. The Contractor will be required to demonstrate that all other practical strategies to source and treat materials to be compacted within specification have been exhausted.

In this event, the Design Engineer will review the circumstances and conditions and present a course of action for the Contractor and the Principal to review and agree. This may include insitu testing (such as shear vane) to confirm achievement of the minimum undrained shear strengths adopted in the stability analyses, and a programme for monitoring the piezometers during construction (including trigger limits).

2.6.6 Control Charts

The Contractor shall maintain and keep up to date charts on site as defined below:

- A longitudinal section (schematic) of the works identifying the precise location and extent of all lots tested, the lot identification, the dates of all compaction tests and the date at which compliance was achieved.
- A control chart presenting all test results (including re-tests), pertinent test details and confirmation of compliance with the acceptance criteria.
- For every nuclear meter used on site, control charts of daily standard counts (for applicable channels) and secondary block checks shall be made available to the Principal for inspection when required.

2.6.7 Reduction in Testing Frequency

Where the Contractor can demonstrate consistent conformance with the Specification requirements for a given test or group of tests, to the satisfaction of the Principal, then the Contractor may apply to the Principal to reduce the testing frequency to no less than half that defined in the Specification for the relevant test or group of tests.

A minimum of four (4) consecutive lots or batches of work tested in accordance with the Specification shall be taken to assess consistency.

If, when the testing frequency has been reduced, a subsequent test by the Principal shows a departure from the consistent conformance, then the Contractor shall return to full testing frequency as defined in the Specification. The Principal shall likewise return to a full testing frequency if the materials, methods of work or conditions of application alter.

Notwithstanding the provision for reduced frequency of testing, the Principal may elect to test a particular lot.

2.7 Geotextile

The Principal shall supply and the Contractor will install a layer of geotextile filter fabric (labelled interchangeably as geotextile or geofabric in the Drawings) at the locations shown on the Drawings or as directed.

Geotextile shall be suitable and durable for the intended application as satisfactorily demonstrated by similar and prior applications. Geotextile shall be non-woven 100% polyester or polypropylene (with the exception of inhibitors and/or carbon black added for UV resistance). Polypropylene materials shall be UV stabilised. The geotextile filament shall be rot proof, chemically stable and shall have low water absorbency. They shall resist delamination and maintain their relative position.

All geotextile material shall be free of flaws that may have an adverse effect on the physical and mechanical properties of the geotextile.

All geotextile material shall comply with AS 3706 and have the properties listed in the table below when tested in accordance with AS 3706.

Pore Size	AS 3706.1	90 micron
Pore Size	AS 3706.1	90 micron
Mean wide strip tensile strength	AS 3706.2	20 kN/m in both machine and cross- machine direction
Mean trapezoidal tear strength	AS 3706.3	500 N in both machine and cross- machine direction
Geotextile Strength Rating G ⁽¹⁾	Austroads	2,800
Permittivity	AS 3706.9	1.0 per second

Table 5 Geotextile required properties

(1) $G = (L X H_{50})^{0.5}$

Where L = CBR Burst Strength (AS 3706.4)

H₅₀ = Puncture Resistance by Drop Cone (AS 3706.5)

A proprietary product and grade of geotextile is shown on the Drawings. If, where an alternative is provided by the Principal it shall meet or exceed the grab tear strength, permittivity and be less than or equal to the apparent opening size of the specified product and the following requirements.

Prior to delivery, all individual roll manufacturer certifications required by the Specification shall be received and approved by the Principal. Delivery of any unapproved roll shall not be allowed and unapproved rolls shall be transported off-site.

Each delivery shall be subjected to a visual inspection prior to being allowed into the permanent Works stocks. For each delivery a quality control certificate shall be provided. The product name shall clearly be identified on each roll. Quality control certificates shall be retained by the Principal for future management of independent quality assurance testing.

After delivery of geotextile to site, the Contractor shall ensure that all geotextile surfaces are kept clean and undamaged and stored away from direct sunlight until covered. Any damaged or improperly stored geotextile materials shall be replaced by the Contractor at the Contractor's expense.

Geotextile shall be laid evenly with no kinks or folds. Geotextile shall be placed such that adjacent lengths of fabric overlap on longitudinal and transverse joints by a width not less than 500 mm. Joints shall be overlapped such that the upslope geotextile is lapped over the downslope geotextile. Geotextile placed on slopes of 1V:5H or steeper shall be laid upslope/downslope.

Any tear or puncture in the fabric caused by the Contractor's operations shall be repaired by the Contractor at its expense by patching using geotextile patches fastened over the tear or puncture. Patches shall overlap the tear or puncture by a minimum of 500 mm in all directions.

3. Concrete

3.1 Scope

The work specified under this Section includes the supply, mixing, placing, compacting and finishing of concrete to the lines, levels and dimensions shown on the Drawings, and it represents the minimum requirements with regard to concrete durability.

3.2 Standards

3.2.1 General

All concrete work shall conform to the requirements of the latest editions, including amendments, of the following Australian Standards, unless varied otherwise by this Specification.

3.2.2 Concrete Generally

AS 1478-1Chemical Admixtures for Concrete, Mortar and GroutAS 3582Supplementary Cementitious Materials for use with Portland and Blended Cement.AS 3600Concrete StructuresAS 3972Portland and Blended CementsAS 3850Prefabricated Concrete Elements Code

3.2.3 Sampling and Testing

- AS 1012 Methods of Testing Concrete
- AS 1141 Methods for Sampling and Testing Aggregates
- AS 3583 Methods of Tests for Supplementary Cementitious Materials for use with Portland Cement

3.2.4 Concrete Supply and Placing

- AS 1379 The Specification and Manufacture of Concrete
- AS 2349 Methods of Sampling Portland and Blended Cements
- AS 2350 Methods of Testing Portland and Blended Cements
- AS 2758 Aggregates and Rock Engineering Purposes

3.2.5 Formwork

AS 3610 Formwork for Concrete

3.3 Materials

3.3.1 General

All concrete shall be dense and well graded.

All materials used in the manufacture of concrete shall, as a minimum, conform to the requirements of this Specification, and shall be subject to approval by the Principal before concreting is commenced. Details of the proposed materials and concrete mix proportions shall be submitted to the Principal at least 28 days prior to their intended use.

3.3.2 Portland Cement

All cement used in the Works shall be Portland Cement of Australian manufacture and shall comply with AS 3972. The type of cement shall be ASCE Specification Type GP General Purpose Portland Cement unless otherwise designated. The maximum alkali content of the cement (measured as Na2O equivalent) shall be 0.6% by weight.

Cement shall not be more than six months old when used in the Works and any cement that has become lumpy or partially set prior to mixing will be condemned and shall be removed from the site immediately.

All bagged cement shall be supplied in unopened and undamaged bags, bearing the Manufacturer's brand and date of manufacture.

3.3.3 Aggregate

Fine aggregate shall consist of clean natural sand.

Coarse aggregate shall consist of crushed stone or gravels composed of clean, sound, hard and durable particles free from dust, clay, organic materials or other deleterious substances. Unless otherwise noted on the Drawings or in this Specification, the maximum size of coarse aggregate to be used in each class of concrete shall be 20 mm.

All fine and coarse aggregates shall comply with the requirements of AS 2758.1 and shall be tested in accordance with AS 1141 Section 38.

All aggregates shall be tested by the Contractor for alkaline reactivity in particular and the results submitted to the Principal. Aggregates with high alkaline reactivity will be rejected altogether.

3.3.4 Water

Clean water of potable quality shall be used in all concrete mixes and shall be free from oil and injurious amounts of alkalis, salts, organic materials or other deleterious substances harmful to concrete or reinforcement.

3.4 Quality and Mix

3.4.1 Quality of Concrete

Concrete shall be specified by required properties as defined in AS 3600. The properties required for each element of the Works are as shown on the Drawings and herein.

The total cementitious content shall be between 320 and 360 kg/m³ of concrete where the concrete grade is N32 or higher.

The concrete to be used shall attain a characteristic compressive strength at 28 days as indicated on the Drawings.

The maximum water / cementitious material ratio for all concrete shall be 0.5 by mass. Liquid additives shall be included in the water volume for the purpose of this calculation.

The consistency of concrete shall be determined by a slump test. Concrete shall be of such consistency that it can be readily placed and compacted without segregation of the materials and without excess water collecting on the surface. Slump measured is to be no greater than 80 mm within the tolerances given in AS 1379 clause 5.2.3, unless a superplasticiser, approved by the Principal, is used.

The Contractor shall note that the strength requirements are a minimum and that other criteria may control the approved mix.

3.4.2 Concrete Mix Design

The Contractor shall be responsible for the design and production of all concrete used in the Works. The use of ready-mixed concrete shall in no way lessen or remove this responsibility.

Not less than 2 weeks prior to the commencement of concreting operations, the Contractor shall submit details of the mix design for each specified class of concrete for the approval of the Principal. The submission shall include the following information:

- Class of Concrete;
- Supplier of Concrete (if ready mixed);
- Proportion (by weight) of various materials comprising the mix (including any proposed admixtures);
- Certificates of Compliance for various materials used in the mix;
- Target Strength;
- Target Slump;
- Degree of quality control over the mix;
- Method of Placing.

The Contractor shall state the proposed degree of control when details of the proposed mix design are submitted. If during the course of the work this degree of control is not maintained (as evidenced by either the batching and mixing and placing methods employed, or by the strength of test cylinders taken on the work), the Principal may withdraw approval, pending either redesign of the mix, or establishment of improved quality control, or revision to the method of placing. Where a new mix design is prepared, testing of the new mix shall be carried out entirely at the Contractor's expense.

Approval of the Concrete Mix Design shall be a HOLD POINT.

3.5 Grades of Concrete

Concrete Grades

The following table presents the required concrete grades for various structures.

Structure	Required Grade (MPa)	
Spillway Control Structure	N32	
Outlet Tower Concrete	N32	
Survey Monuments and Points	N25	

Table 6

3.6 Testing

3.6.1 Sampling and Testing

All concrete with strength grade equal to or greater than 32 MPa.

The Contractor shall take samples from separate batches of concrete selected at random during the placing operation. Each specimen cylinder shall be identified with the batch or as directed by the Principal. The minimum rate of sampling shall be as follows:

- Slump testing on at least the first two trucks where concrete testing is to be carried out
- 4 cylinders from each lot of 25 cubic metres (or part thereof);
- For each set of specimen cylinders one shall be tested at 7 days and two at 28 days or as directed by the Principal;
- At least one specimen shall be tested at 3 days to obtain an indication of the strength development rate on major pours, as directed by the Principal.

Specimen cylinders taken from the batches shall be cured in accordance with AS 1012, except that they shall be stored at the site of the Works under moist conditions, sheltered from the influence of sun and wind and protected from extremes of temperature.

The compressive strength of a batch of concrete shall be determined as the mean 28-day strength of the specimen cylinders cast from that batch.

All concrete shall be supplied by a quality assured supplier who shall provide test certificates to confirm that the concrete complies with maximum drying shrinkage and density requirements.

All test results shall be submitted to the Principal, and in particular, notification of any failures on the test specimens shall immediately be given to the Principal.

Should the Contractor seek to vary any requirement of this Specification and should the Principal's approval be subject to additional sampling and testing at other than 28 days, the cost of such additional sampling and testing shall be borne entirely by the Contractor.

The Contractor shall keep on the site, a log book in which is recorded the following information:

- Specimen cylinder number or identifying mark;
- Date of sampling and name of operator;
- Slump of batch from which sample is taken;
- Portion of structure represented by the sample;
- Test results.

This logbook shall be available for inspection by the Principal at all times.

Batch Certificates

All concrete supplied shall at arrival on site be provided with batch plant certificates for the respective batches. Batch certificates shall contain all information about the concrete supplied as required by AS 1379.

3.6.2 Acceptance of Testing

Where the concrete already placed is classed as defective and liable to rejection, the criteria for rejection shall be as set out in AS 3600.

Should any concrete be liable to rejection, the Contractor may submit specimens cut from the completed work for testing. The number, form and dimensions of the specimens, and the location from which they are cut shall be subject to the approval of the Principal. The Contractor shall arrange for an approved Testing Authority to carry out the required tests and submit the results directly to the Principal, with a copy to the Contractor.

The Principal will then consider the test results and other information, and may, with absolute discretion, determine whether the strength of the specimens cut from the work and adjusted for the age of the specimens, is to be taken as the actual strength of the concrete for acceptance purposes. The entire cost of cutting specimens from the completed work, testing and restoration shall be borne by the Contractor.

Nothing in the foregoing shall in any way relieve the Contractor of any contractual responsibilities, and the Principal shall be indemnified against any damage from the quality of the concrete being below that specified.

Concrete classed as defective and rejected by the Principal shall be removed from the Works and, together with any other work subsequently erected thereon, shall be replaced, at the Contractor's entire expense, with concrete complying with the Specification.

3.7 Formwork and Falsework

3.7.1 Formwork

Formwork shall be designed and constructed in accordance with AS 3610 and AS 3600. All forms shall be built surface smooth and mortar tight and have sufficient rigidity to prevent distortion due to the pressure of fresh

concrete and other construction loads. Particular care shall be taken in the preparation of the joint around piles penetrating the soffit.

It should be noted that welding of formwork or its supports to exposed parts of permanent piles will not be permitted.

Forms for plane exposed surfaces shall consist of plastic-coated plywood, waterproof plywood, timber forms lined with tempered hard board or close-fitting unwarped metal forms. Joints in the form sheeting for plane exposed concrete surfaces shall be either vertical or horizontal unless otherwise specified.

Forms for surfaces not exposed to general view may consist of standard timber or metal panels.

Timber forms shall be constructed and maintained in such a manner as to prevent warping and opening of joints due to shrinkage of the timber. The timber shall be free of any defects which will affect the structure.

Unless otherwise shown on the Drawings all corners shall be provided with 20 mm x 20 mm chamfers or fillets.

All forms shall be set and maintained to line and level such that the finished concrete will conform to the proper levels, dimensions and contours, as shown on the Drawings, within the tolerances specified.

All forms shall be cleaned and coated with the lightest practical coating of release agent prior to erection. The Contractor shall ensure that reinforcing steel and construction joints are not contaminated with release agent.

Pockets formed in the concrete for subsequent casting of anchor bolts, or the like shall be carefully constructed to allow for the full reinforcement to project through the forms.

Metal form-ties shall be of an approved type. If the form-ties are cast in, they shall be of a type which permits removal of the end fittings to a depth at least 50 mm below the finished surface of the concrete. Ordinary wire ties shall not be used.

Form-ties shall be located in a uniform symmetrical pattern relative to the finished surface. The cavities left when the end fittings of embedded ties are removed shall be as small as possible and shall be filled with an approved epoxy cement mortar at the earliest possible time. The surface of such filled cavities shall be left smooth and uniform in colour.

When forms are re used, their original shape, strength, rigidity, mortar tightness and surface smoothness shall be maintained. Forms which, in the opinion of the Principal, become unsatisfactory in any respect shall not be used.

The Contractor shall check the formwork prior to placing the concrete. All dimensions, particularly those affecting the construction of subsequent portions of the Works, shall be carefully checked.

All formwork shall be subject to inspection and approval by the Principal or a nominated representative, and shall not be used until approval has been given. Notwithstanding any inspection or approval given by the Principal, the responsibility for the structural sufficiency of the whole of the formwork shall remain entirely with the Contractor.

Should any formwork be displaced during concreting, or within the periods specified for the retention of formwork, the concrete shall be removed between such limits as the Principal may determine, construction joints shall be formed, and the section of work reconstructed, after the formwork has been strengthened and adjusted.

3.7.2 Falsework

The design and erection of falsework, the method of founding or supporting the falsework, and the time, order and manner of its release shall all be the responsibility of the Contractor, but will be subject to the approval of the Principal before commencement of construction. The Contractor shall supply the Principal with detailed drawings of such falsework at least four (4) weeks prior to the commencement of erection.

It should be noted that welding of falsework or its supports will not be permitted to exposed parts of permanent piles.

All falsework shall include adequate bracing to prevent movement of piles or formwork during concreting.

The provisions of such drawings and their approval by the Principal shall in no way relieve the Contractor of any responsibility for the satisfactory performance of the falsework.

Subject to the Principal's approval, falsework may be supported on completed sections of the Works provided that the construction loads subsequently applied do not result in over stressing or instability within the sections, and that due allowance is made for any deflection of the sections.

If any structural strengthening or modification of such sections of the Works is necessary as a condition for approval of their use as support structures, the cost of such strengthening or modification shall be borne entirely by the Contractor.

The Principal's approval of the use of completed sections of the Works as support structures for falsework, shall in no way relieve the Contractor of any responsibility for the cost of restoration or repair of any damage occasioned by or resulting from such use.

3.7.3 Finishes

Formwork shall be in accordance with the AS 3610 Class 2. Aluminium alloy sheeting shall not be used as formwork.

The number of re uses, and the conditions of faces and edges of forms, shall be consistent with the formwork surface class specified. Forms shall not be disturbed until the concrete in contact with them has hardened sufficiently to withstand such action without damage.

3.7.4 Stripping

Forms shall be so designed and constructed that they may be removed without injuring the concrete surfaces and without excessive jarring or hammering.

Forms and falsework beneath the soffit of self-supporting concrete elements shall not be removed until a minimum of 75% of the Characteristic Compressive Strength at 28 days has been achieved and in accordance with AS 3610 Table 4.2, unless noted otherwise. Special care shall be exercised when removing formwork from the vicinity of piles to ensure that the protective treatment is not damaged.

Precast reinforced and prestressed concrete elements shall not be lifted before a minimum of 75% of the Characteristic Compressive Strength at 28 days has been achieved, or any larger limit as specified on the Drawings, and approval must be obtained from the Principal before commencement of lifting.

3.8 Ready Mix Concrete

The Contractor shall submit the name of the proposed Supplier to the Principal at least four (4) weeks prior to the commencement of concreting operations.

Only those Suppliers approved by the Principal will be allowed to supply ready mixed concrete. The Principal shall reserve the right to withdraw approval of any Supplier at any time.

The production and delivery of ready mixed concrete shall be in accordance with the requirements of AS 1379 except as otherwise specified hereunder.

Mixing speeds shall range from 12 - 20 rpm and agitating speeds shall range from 2 - 3 rpm.

Notwithstanding any requirement to the contrary contained in AS 1379, the sampling, testing and acceptance of ready-mixed concrete shall conform to the requirements of this Specification.

Ready-mixed concrete shall be delivered to the site in trucks of the revolving drum type and the use of nonagitating equipment for this purpose will not be permitted.

The quantity of concrete delivered in any truck shall not exceed the rated capacity of its agitator drum. The timing of deliveries shall be such as to ensure an essentially continuous placing operation.

3.9 Placing

3.9.1 General

Before commencing placement of concrete in any section of the Works, the formwork shall be checked for accuracy and tightness and all joints shall be effectively stopped.

No concrete shall be placed in the Works until the formwork and reinforcing steel have been inspected and approved by the Principal and any foreign material has been completely removed from the forms and reinforcing.

Concrete shall be liable to rejection if the slump is in excess of the maximum slump specified.

Except for underwater concrete, all concrete shall be placed under dry conditions.

Concreting operation shall be carried out in a continuous manner between approved construction joints. Should a delay occur which allows the concrete in the forms to take its initial set as determined by the Principal, a construction joint shall be formed as specified elsewhere herein and fresh concrete shall not be placed against the concrete already in the forms at that time.

Chutes, if used, shall be arranged in a manner which avoids segregation of the concrete. Apart from an initial flushing immediately prior to commencement of concreting, the use of water in chutes to assist movement of concrete will not be permitted.

Buckets shall be capable of discharging portions of their load at a time in a controlled manner.

Concrete shall not generally be dropped from a height exceeding 2 m, nor shall it be placed in any other manner which results in segregation or loss of concrete and damage to formwork and reinforcement.

If placing operations necessitate a drop of more than 2 m, the concrete shall be placed using a method approved by the Principal.

Fresh concrete shall be deposited within the forms as near as possible to its final location. Excessive use of vibrators and tamping rods to move the concrete along the forms will not be permitted.

Formwork shall not be disturbed or adjusted during the concreting operation, and shall remain undisturbed up to the minimum removal time specified.

No strain shall be placed on any projecting reinforcing steel for a period of at least 24 hours following completion of concreting.

The full thickness of all structural members shall be concreted in the one operation, unless approved otherwise by the Principal.

If a cessation of work becomes unavoidable, a construction joint shall be formed as specified elsewhere herein, at the point of stopping. When work is resumed, the treatment of the concrete at the cold joints shall be as specified under Section 3.9.5. "Construction Joints".

3.9.2 Pumping

Approval to use concrete pumps to place concrete may be given by the Principal, subject to the Contractor submitting full details of the type and capacity of the pumps, pipework, and procedures to be adopted, and satisfactory evidence that any modifications to the concrete proportions and consistency to suit pumping procedures shall comply with the provisions and intent of this Specification.

The pump shall have a variable speed control and shall be capable of pumping concrete containing 20 mm aggregate through delivery lines not less than 75mm diameter.

Delivery lines shall be of an approved metallic type. Under no circumstance shall aluminium alloy pipes or fittings be used nor shall concrete be permitted to come into contact with aluminium during its manufacture, transport or placing.

Where delivery lines are exposed to the direct sunlight, they shall be protected by covering with bags, wet hessian or other approved means.

Direct, efficient communication shall be maintained at all times between the pump operator and the concrete placing crew.

During delays in delivery of concrete to the pump, concrete in the lines shall be pumped at approximately 10 minute intervals (5 minutes where the ambient temperature is greater than 32°C) to ensure that the concrete is "live". For piston type pumps at least two strokes of the piston shall be made at each pumping interval. In the event of any delay or breakdown in the equipment not exceeding 20 minutes, during which time concrete cannot be placed, the concrete already in place shall have the "wet edges" maintained in an approved manner.

If the delay exceeds twenty minutes, an emergency construction joint shall be formed where directed, the concrete completed to it, and all concrete in the pipeline discarded.

Should the atmospheric (shade) temperature on the site not exceed 18°C, the concrete in the receiving hopper and lines may be placed in the work on the resumption of pumping, when approved by the Principal.

Should the temperature exceed 30°C, the receiving hopper and lines shall be cleaned out and the concrete contained therein discarded.

Where the temperature lies between 18°C and 30°C, the using or discarding of the concrete shall be determined by the Principal.

In any case where initial setting of the concrete has begun in the hopper or discharge lines, the concrete shall be discarded.

3.9.3 Placing in Hot Weather

The Contractor shall take all practical precautions to maintain the concrete at a temperature not exceeding 32°C. Such precautions shall be taken to avoid premature stiffening of the fresh mix, reduce water absorption and evaporation loss, and prevent shrinkage, settlement, and pre setting cracks.

Such precautions shall be taken if the ambient air temperature at the time of placing is likely to be greater than 32 °C and may include:

General	(a)	All concreting shall be placed during the early morning and commencement of work shall be timed to ensure completion of concreting before 10.00 am to the satisfaction and approval of the Principal.
At the mixer	(a)	Using crushed ice in the mixing water.
	(b)	Liquid nitrogen cooling of the mix.
	(c)	Shading and watering aggregate stockpiles.
	(d)	Insulating containers and delivery lines.
	(e)	Any combination of the above methods.
At the site	(a)	Cooling formwork by dampening with water sprays.
	(b)	Shading work areas.
	(c)	Erecting wind breaks.
	(d)	Minimising the time for placing and finishing

No concrete shall be placed with temperature greater than 32°C at the time of placing in the forms.

Unless a concrete mix with a retarder has been approved, concrete shall be placed and compacted within the time limits specified below.

 Table 7
 Maximum Elapsed Time from Time of Charging Mixture

Concrete Temperature	Ready-mixed (min)	Site-mixed (min)
Less than 24C	120	60
24C - 27C	90	45
25C - 30C	45	30
30C - 32C	30	20
Over 32C	Not acceptable	Not acceptable

After placing and compacting, the concrete shall be covered with an impervious membrane or hessian kept wet until cured to the satisfaction of the Principal.

The Contractor's proposed hot weather placing techniques shall be subject to the Principal's approval, and full details shall be submitted to the Principal at least 28 days before the first concrete pour at a time when hot weather could be expected.

3.9.4 Cold Weather Concreting

In cold weather the temperature of freshly mixed concrete shall be maintained within the limits shown below. "Outdoor" air temperature is air temperature at time of mixing, or predicted or likely air temperature during the next 48 hours. Before and while placing concrete maintain the temperature of formwork and reinforcement at more than 5°C. Do not use calcium chloride, salts, chemicals or other material in the mix to lower the freezing point of concrete. Do not allow frozen materials to enter the mixer. Keep forms, materials, equipment in contact with concrete free of frost and ice. Heat concrete materials (other than cement) to the minimum temperature necessary to ensure temperature of placed concrete is within the limits specified. Maximum water temperature: 60°C when placed in the mixer.

Outdoor Air Temperature	Temperature of Concrete	
	Minimum	Maximum
>50C	10 °C	32 °C
<50C	18 °C	32 °C

3.9.5 Construction Joints

A construction joint is defined as the junction between two successive concreting operations in an element, intended to act structurally.

Generally, construction joints shall be located away from the mid span of the slab and not within 1000mm of any supports, openings or inserts in the deck. Pile caps, and similar sections shall be cast in one operation without construction joints.

Whenever the work of placing concrete is delayed so that the initial set has taken place as determined by the Principal, the point of stopping shall be deemed a construction joint and formed accordingly. The initial set shall be deemed to occur if a standard vibrator will not penetrate the surface under its own weight, and will not re work the concrete mix once penetration has been achieved.

Where such a point of stopping occurs at a location considered unsuitable by the Principal, the Contractor shall remove such concrete already placed as will enable a satisfactory construction joint to be formed where directed by the Principal.

Before depositing new concrete on or against concrete which has hardened, the forms shall be re tightened. The surface of the hardened concrete shall be roughened in a manner that will expose sound concrete at the surface. It shall be thoroughly cleaned of foreign matter and laitance and saturated with potable quality water.

Immediately before the next pour, the joint surface shall be saturated with potable quality water.

3.10 Compaction and Curing

3.10.1 Compaction

Concrete shall be thoroughly compacted during and immediately after depositing. Concrete other than no-fines concrete shall be compacted with high frequency internal vibrators in the manner described below. Hand compaction in lieu of mechanical vibration will be allowed only as an emergency measure when approved by the Principal.

The vibration shall be internal except for the use of form vibrators.

Vibrators shall be of an approved type, capable of transmitting vibration to the concrete at frequencies of not less than 12,000 impulses per minute at an intensity to visibly affect a zero slump concrete at a radius of 300mm.

For concrete of 28 day strengths of 20MPa or less, the vibrators shall be capable of transmitting vibration to the concrete at frequencies of not less than 8,000 impulses per minute at such an intensity to visibly affect a 25mm slump concrete at a radius of 300mm.

The Contractor shall provide a sufficient number of vibrators to properly compact each batch immediately after it is placed. The minimum number of vibrators to be provided will depend on the rate of placing concrete but in no case shall be less than 1 vibrator for each 5 m³ of concrete or part thereof placed per hour, with a minimum of 2 vibrators.

At least one vibrator in working order shall be held in reserve at all times.

A vibrator shall be inserted into the concrete at successive positions not more than 500mm apart and vibration shall continue at each position until air bubbles cease to emerge. It shall then be withdrawn slowly.

Vibrators shall be inserted so as to thoroughly compact the concrete around the reinforcement and embedded fixtures into the corners and angles of the forms. Vibration shall be applied at the point of deposit and in the area of freshly deposited concrete. Where more than one layer is being placed in a continuous operation the vibrator shall be inserted through the layer into the layer below.

The vibrators shall be inserted into and withdrawn from the concrete slowly. The vibration shall be of sufficient duration to thoroughly compact the concrete, but shall not be continued so as to cause segregation.

Vibration shall not be applied, directly to or through the reinforcement, or to sections or layers of concrete which have hardened to the degree that the concrete ceases to be plastic under vibration. It shall not be used to make

concrete flow in the forms over distances so great as to cause segregation, and vibrators shall not be used to transport concrete in the forms.

3.10.2 Form Vibrators

The provision of this section shall also apply to precast members except that if approved by the Principal, the Manufacturer's method of vibration may be used. For precast slab units, internal vibration shall be used in conjunction with external mould vibration.

Except where authorised by the Principal and in thin web sections, concrete shall be placed in horizontal layers not more than 300mm thick. Each layer shall be placed and compacted before the preceding layer has taken its initial set.

Immediately following the discontinuance of placing concrete, all accumulations of mortar splashed upon exposed reinforcing steel and the surfaces of the forms shall be removed.

3.10.3 Protection and Curing

The Contractor shall ensure that all concrete is properly and adequately cured. The Principal shall approve the curing methods before any concrete is placed. The Contractor shall modify or change the curing procedures as directed by the Principal at any time during the work of the Contract, if the Principal is not satisfied that the concrete is properly cured.

Initial curing of exposed concrete surfaces shall be commenced as soon as the surface of the concrete has hardened sufficiently and shall continue for not less than 72 hours. The concrete surface shall be kept continuously moist, preferably by ponding, but where this method is considered impractical, an absorbent cover that is kept continuously wet shall be used.

Final curing shall commence immediately following the initial curing period and shall continue for not less than seven days or such longer period as the Principal may direct. During this period the curing shall be carried out either by continuing the method used during the initial curing period or by the following method, subject to the provisions of the hot weather placement in Clause 3.9.3.

a) Waterproof Sheeting Covers

The Concrete shall be covered with an approved waterproof sheeting maintained in close contact with the surface of the concrete.

The edges of the sheeting shall be taped or shall overlap by at least 200mm, and the whole sheeting shall be securely held in position. Any damage to, or displacement of the sheeting during building operations, shall immediately be made good, and should the concrete surface show signs of drying out during the final curing period, the sheeting shall be temporarily removed, the surface of the concrete wetted, and the sheeting replaced as before.

b) Membrane Curing Compound (note that not permitted under hot weather conditions)

Proprietary membrane curing compounds shall be to the specific approval of the Principal and shall be applied in accordance with the Manufacturer's instructions and recommended rate, half to be applied back and forth in one direction, and the remainder at right angles. The compound shall be thoroughly mixed before application. It shall provide a continuous flexible coating without visible breaks or pinholes which remains unbroken for not less than the curing period after application.

Curing compounds shall not be used on surfaces which are to receive other finishes or further concrete.

The curing compound shall be either a wax emulsion or chlorinated rubber based composition containing a fugitive white pigment. The Contractor shall submit the name, type and Manufacturer of the proposed compound for the approval of the Principal, at least two (2) weeks prior to commencing concrete operations.

Steel forms exposed to direct sunlight, and all wood forms in contact with the concrete during the final curing period, shall be kept wet. If forms are removed during the curing period, one of the previously described curing methods shall be employed and continued for the remainder of the period.

The Contractor shall ensure that a uniform colour is achieved on adjacent surfaces by applying uniform curing methods, the details of which must be submitted for the approval of the Principal at least 28 days before concreting commences.

3.10.4 Concrete Cracking

The Contractor shall treat all cracks at his expense as directed by the Principal using the method described as follows:

- Cracks of width between 0.2 and 1.0 mm shall be sealed using an approved low viscosity epoxy.
- Cracks of width greater than 1.0 mm shall be grooved out, cleaned and sealed with a special modified bentonite/cement grout.
- Cracks which have developed along construction joints shall be sealed over their full depths with a low viscosity epoxy, except that where cracking of width greater than 1.0 mm occurs in these joints, they shall be grooved out and sealed with an approved polyurethane sealant in addition to sealing with a low viscosity epoxy.

Polyurethane sealants and low viscosity epoxies shall be used in accordance with the manufacturer's written instructions and allowance made for the time to complete full cure.

Where a low viscosity epoxy is used, the surface of the concrete in the vicinity of the construction joints and cracks shall be wire brushed, dried and all dust removed by air blasting or other approved methods before application of the epoxy.

Where grooving out followed by sealing is required, the grooves shall be cut 10 mm wide x 10 mm deep, by means of a suitable cutting disc over the construction joint or located crack. Such grooves shall be wire brushed and kept clean, dried and blown free of dust immediately before application of either the grout, epoxy or sealant primer, whichever is applicable.

3.11 Surface Finish

The finish of formed concrete surfaces shall be in accordance with and as defined in AS3610, except that surface tolerances shall be in accordance with the Specification. Unless shown otherwise on the Drawings, surfaces which will remain permanently exposed shall be formed using Class 2 formwork and surfaces which will be permanently concealed shall be formed using Class 3 formwork

Unformed surfaces shall be of finish Class U1.

3.12 Tolerances and Building-In

3.12.1 Dimensional Tolerances

Where tolerances are not stated in the Specification or the Drawings for any individual structure or feature thereof, they shall not be greater than the deviations listed below.

These tolerances shall be applied to the completed work, and forms shall be set and maintained so as to ensure the completed work is within the tolerance limits. Should a specified tolerance be exceeded, the work shall be liable to rejection or remedy at the Principal's discretion, and the nature of remedial work shall be at the Principal's discretion. The cost of any remedial work shall be borne entirely by the Contractor.

3.12.2 Pre-Cast Concrete

Item	Tolerance Requirements
Size of members of thickness of slab	+2 mm
	- 0 mm
Cover of concrete over reinforcement	+2 mm
	- 1 mm
Departures from plane in exposed surfaces.	

Gradual	-2 mm in 3 m	
Sudden	-1 mm in 3 m	
Departure from line on exposed edges	-2 mm in 3 m	
Departure from Contract dimension where other	0 – 20 m	2 mm
tolerances do not apply.	over 20 m	5 mm

3.12.3 Cast Insitu Concrete

Item	Tolerance Requirements
Size of members of thickness of slab	-0 mm +6 mm
Cover of concrete over reinforcement	+6 mm -0 mm
Departures from plane in exposed surfaces.	
Gradual Sudden	-6 mm in 3 m -3 mm in 3 m
Departure from line on exposed edges	-3 mm in 3 m
Departure from Contract dimension where other tolerances do not apply.	0 – 20 m 5 mm over 20 m 10 mm

Where in the opinion of the Principal the application of the above tolerances would adversely affect the serviceability of the structure, the tolerances may be amended at the Principal's discretion. Where appearance or serviceability of the structure will not be impaired (e.g., in concrete to be subsequently buried below ground level), the Principal may relax the tolerances above.

3.13 Handling of Pre-Cast Units

Precast concrete shall only be used where shown on the drawings or approved by the Principal.

All precast units shall be lifted only by double slinging from twin lifting devices or a suitable spreader bar and single hoist from attachments near each end. No other lifting points shall be permitted without the prior approval of the Principal.

Precast units shall be lifted by both ends simultaneously during general transportation, but this may be altered to a single vertical lift from the top during installation.

All precast units shall be stacked by placing soft timber packing under both end lifting points, whether during transportation or when stockpiled for subsequent use. Under no circumstances shall any other means of stacking be adopted.

Units may be placed on top of each other, spaced by timbers at support points mentioned above, provided that the net bearing stress so produced on the lowermost unit is less than 7MPa, and provided furthermore that the Principal is satisfied as to the suitability of the foundation material and the stability of the stacked units.

During erection, units shall be handled by slinging from lifting points only, and all movements shall be gradual and procedures used shall be such as to minimise shock loading. Precast units shall always be handled with care to avoid damage.

All damaged precast units (cracked or spalled) are to be set aside for inspection by the Principal repair or recasting to be undertaken at the Principals discretion.

3.14 Grout, Mortars and Other Components

3.14.1 Materials in Sand/Cement Mortars and Grouts

Cement and water shall be as specified in this document.

Sand shall consist of sharp, coarse, clean siliceous sand, free from dust, clay, organic matter or other deleterious substances.

Grading and fineness shall be such that the mortar produced from the sand shall be impervious to moisture.

The sand, cement and water shall be mixed in the preparation specified in the relevant clauses of this Specification. They shall be efficiently mixed and no segregation shall be allowed. The water cement ratio for the grout or mortar shall be the minimum required to allow placement as specified herein.

3.14.2 Miscellaneous Compounds

Where proprietary sealants, epoxy compounds, bond breaking compounds, glues and grouts are used, details shall be supplied for the approval of the Principal prior to use. Such items shall always be applied strictly in accordance with the Manufacturer's Specification.

3.15 Pre-Cast Workshop Drawings

Prior to the proposed commencement of fabrication of precast portions of the Works, the Contractor shall submit Workshop Drawings for the approval of the Principal. Approval by the Principal shall not relieve the Contractor of any responsibility in regard to the proper execution of the Work under this Contract.

4. Reinforcement

4.1 General

4.1.1 Responsibilities

Requirement: Provide concrete reinforcement, as documented.

4.1.2 Standards

Reinforced concrete construction: To AS 3600.

Reinforcement production: To AS4671

4.1.3 Tolerances

Fabrication and fixing: To AS 3600 clause 17.2.

Reinforcement position: To AS 3600 clause 17.5.3.

4.1.4 Submissions

Submit the following:

- Certification: Valid ACRS certification of compliance with AS 4671 for manufacturers, processors, suppliers and mill, product tags and supporting documentation for all reinforcement to be included.
- General: Details of any proposed changes to documented reinforcement.
- Damaged galvanizing: Details of proposed repair to AS/NZS 4680 Section 8.
- Mechanical bar splices: Details and test certificates for each size and type of bar to be spliced.
- Provision for concrete placement: Details of spacing or cover to reinforcement that does not conform to AS 3600.
- Splicing: Details of any proposed changes to documented requirements.
- Welding: Details of any proposed welding of reinforcement.

4.1.4.1 Materials

Reinforcement strength and ductility: Submit type-test reports to verify conformance to AS 3600 Table 3.2.1 for each reinforcement type.

The reinforcement must have unique marks to identify the supplier and these must correspond to the ACRS certification provided with the reinforcement.

4.1.5 Inspection

Inspection: Give notice so that inspection may be made of the following:

- Cores and embedment's fixed in place.
- Reinforcement fixed in place, with formwork completed.
- Inspection of reinforcement shall constitute a HOLD POINT.

4.2 Products

Steel reinforcement

Standard: To AS/NZS 4671.

Surface condition: Free of loose mill scale, rust, oil, grease, mud or other material which would reduce the bond between the reinforcement and concrete.

Bars are to be supplied to site pre-bent to required shapes, reinforcement cages to be prefabricated off site as far as practicable. Bend diameters are to be to AS3600, do not bend after galvanising or application of other coating.

4.2.1.1 Tie wire

General: Annealed steel 1.25 mm diameter (minimum).

External and corrosive applications: Galvanized.

4.3 Execution

4.3.1.1 Dowels

Fixing: If a dowel has an unpainted half, embed in the concrete placed first.

Tolerances:

Alignment: 1:150.

Location: ± half the diameter of the dowel.

Grade: 250 N.

4.3.1.2 Cover

Concrete cover shall be minimum:

Generally - 30 mm cover

Cast against the ground - 45 mm

4.3.1.3 Supports

Proprietary concrete, metal or plastic supports: To AS/NZS 2425 and as follows:

Able to withstand construction and traffic loads.

With a protective coating if they are ferrous metal, located within the concrete cover zone, or are used with galvanized or zinc-coated reinforcement.

Spacing:

Bars: ≤ 60 diameters.

Mesh: ≤ 800 mm.

Supports over membranes: Prevent damage to waterproofing membranes or vapour barriers. If appropriate, place a metal or plastic plate under each support.

4.3.1.4 Projecting Reinforcement

Protection: If starter or other bars extend beyond reinforcement mats or cages, through formwork or from cast concrete, provide a plastic protective cap to each bar until it is cast into later work.

4.3.1.5 Tying

General: Secure the reinforcement against displacement at intersections with either wire ties, or clips. Bend the ends of wire ties away from nearby faces of formwork or unformed faces to prevent the ties projecting into the concrete cover. Ensure all bars are restrained before stopping work to prevent bars rolling underfoot

Beams: Tie stirrups to bars in each corner of each stirrup. Fix other longitudinal bars to stirrups at 1 m maximum intervals.

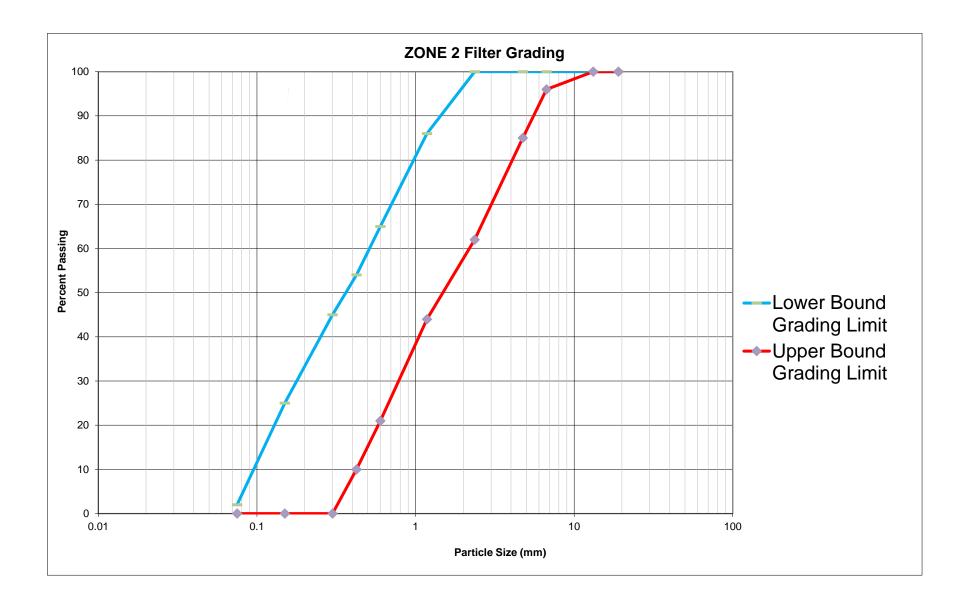
Bundled bars: Tie bundled bars in closest possible contact. Provide tie wire at least 2.5 mm diameter and spaced not more than 24 times the diameter of the smallest bar in the bundle.

Columns: Secure longitudinal column reinforcement to all ties at every intersection.

Mats: For bar reinforcement in the form of a mat, secure each bar at alternate intersections.

Appendices

Appendix A Zone 2 Filter Grading Requirements





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→ The Power of Commitment

TOMINGLEY GOLD MINE RSF 2 STAGE 1 DETAILED DESIGN 12545239



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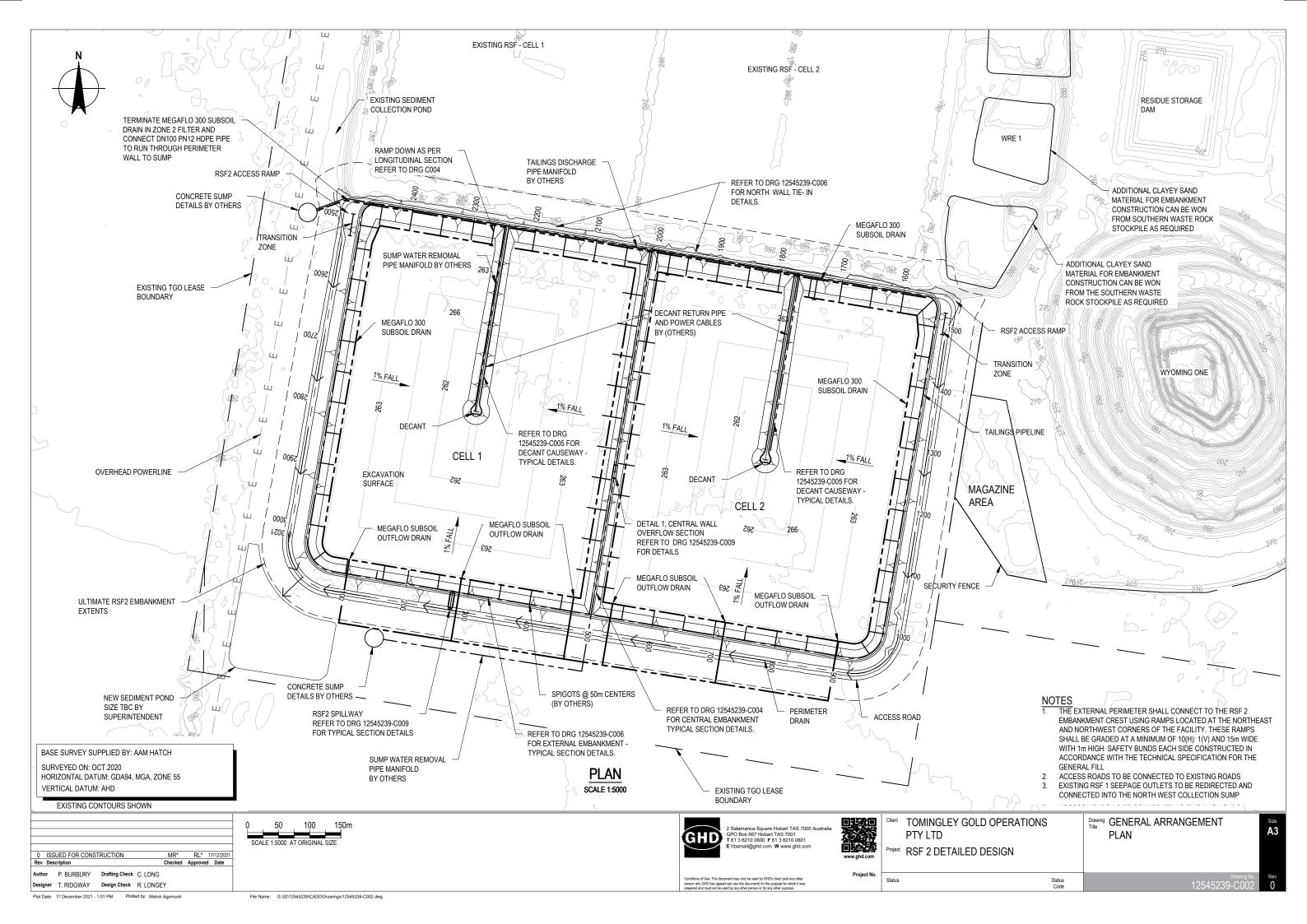
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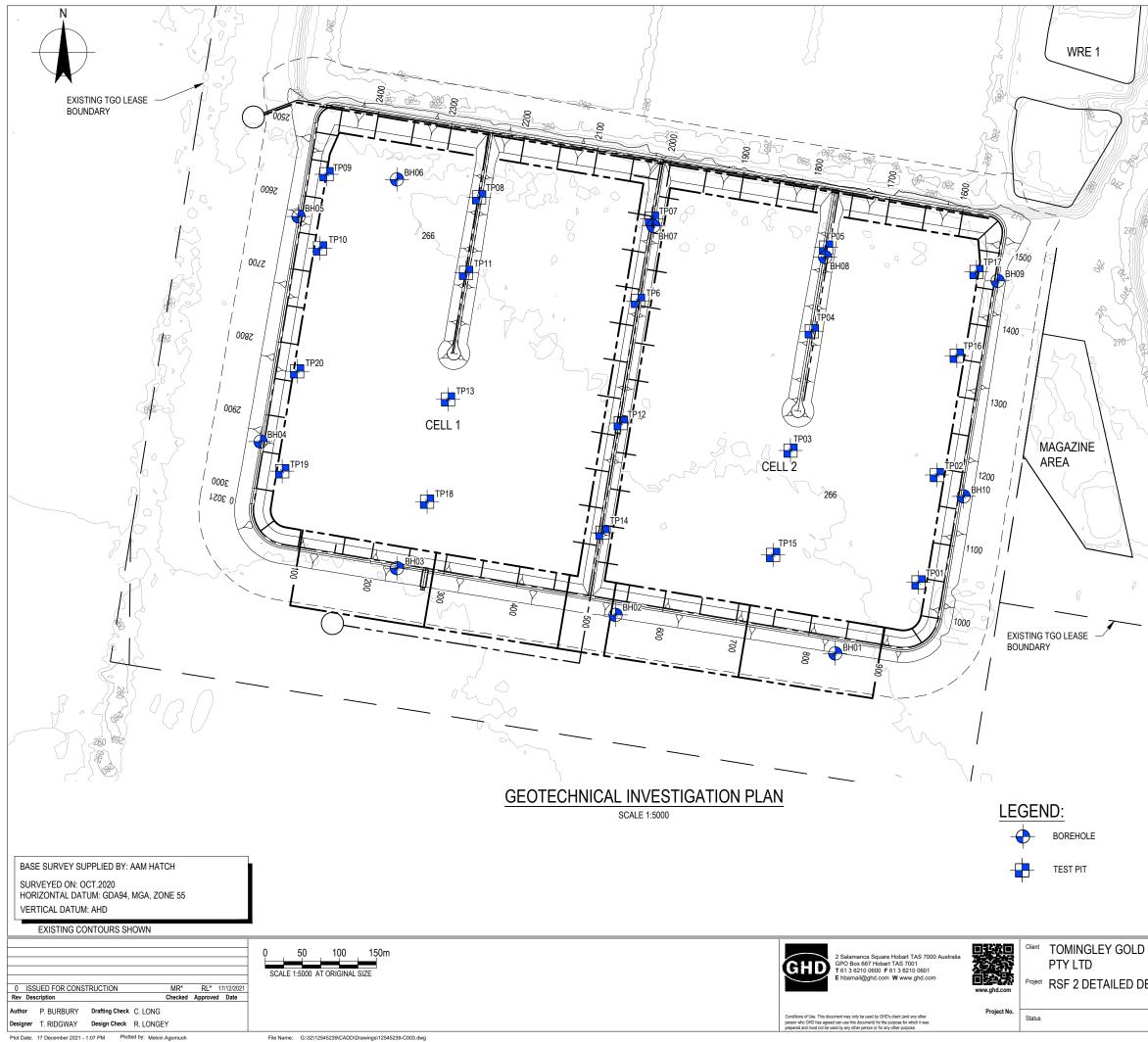


DRAWING TITLE

- COVER SHEET AND DRAWING LIST
- GENERAL ARRANGEMENT PLAN
- GEOTECHNICAL INVESTIGATION PLAN
- TYPICAL EMBANKMENT SECTIONS AND DETAILS SHEET 1 OF 4
- TYPICAL EMBANKMENT SECTIONS AND DETAILS SHEET 2 OF 4
- TYPICAL SECTIONS SHEET 3 OF 4
- TYPICAL SECTIONS SHEET 4 OF 4
- MATERIALS SPECIFICATION
- TYPICAL SPILLWAY SECTION AND DETAIL
- INSTRUMENTATION PLAN
- GPS INSTRUMENTATION TYPICAL SECTIONS
- GNSS GPS UNIT DETAILS
- SURVEY REFERENCE APRIL TAG MOUNT
- STAND ALONE APRIL TAG MOUNT DETAILS
- LONGITUDINAL SECTIONS SHEET 1 OF 5
- LONGITUDINAL SECTIONS SHEET 2 OF 5
- LONGITUDINAL SECTIONS SHEET 3 OF 5
- LONGITUDINAL SECTIONS SHEET 4 OF 5
- LONGITUDINAL SECTIONS SHEET 5 OF 5
- ULTIMATE TYPICAL SECTIONS
- CONCEPTUAL CLOSURE PLAN
- CONCEPTUAL CLOSURE SECTION AND DETAILS

D OPERATIONS	Drawing COVER SHEET AND DRAWING LIST	Size A3
DESIGN		
Status Code	Drawing No. 12545239-C001	Rev 0



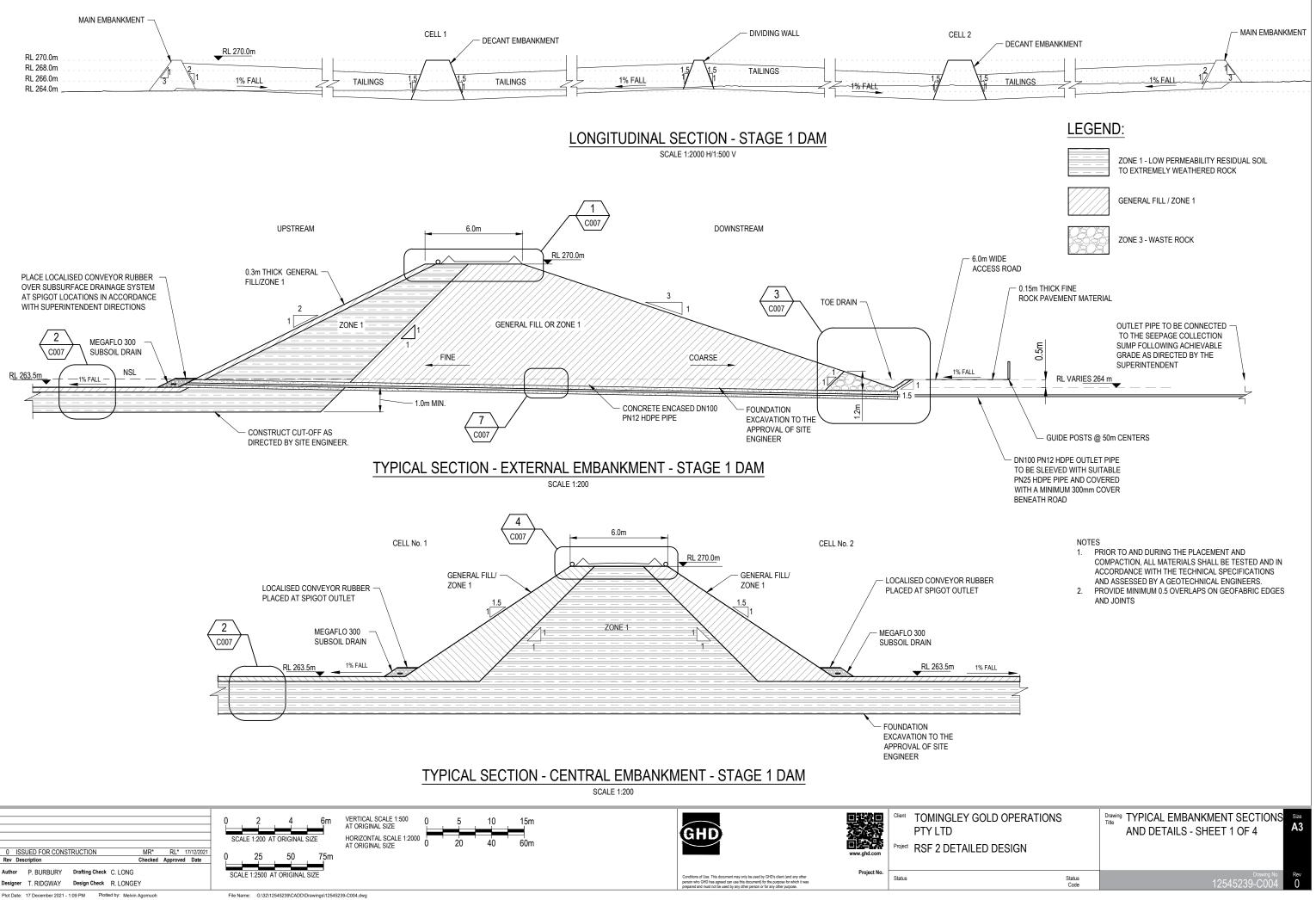


TEST PIT SETOUT POINTS			
POINT	EASTING	NORTHING	
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TP02	613703.700	6393315.000	
TP03	613505.900	6393348.000	
TP04	613534.700	6393509.000	
TP05	613554.000	6393622.000	
TP6	613299.900	6393550.000	
TP07	613318.800	6393661.000	
TP08	613085.400	6393690.000	
TP09	612879.500	6393721.000	
TP10	612870.500	6393621.000	
TP11	613067.700	6393588.000	
TP12	613276.800	6393385.000	
TP13	613043.500	6393417.000	
TP14	613252.200	6393237.000	
TP15	613482.500	6393207.000	
TP16	613730.600	6393476.000	
TP17	613757.100	6393589.000	
TP18	613015.100	6393279.000	
TP19	612819.500	6393320.000	
TP20	612839.700	6393455.000	

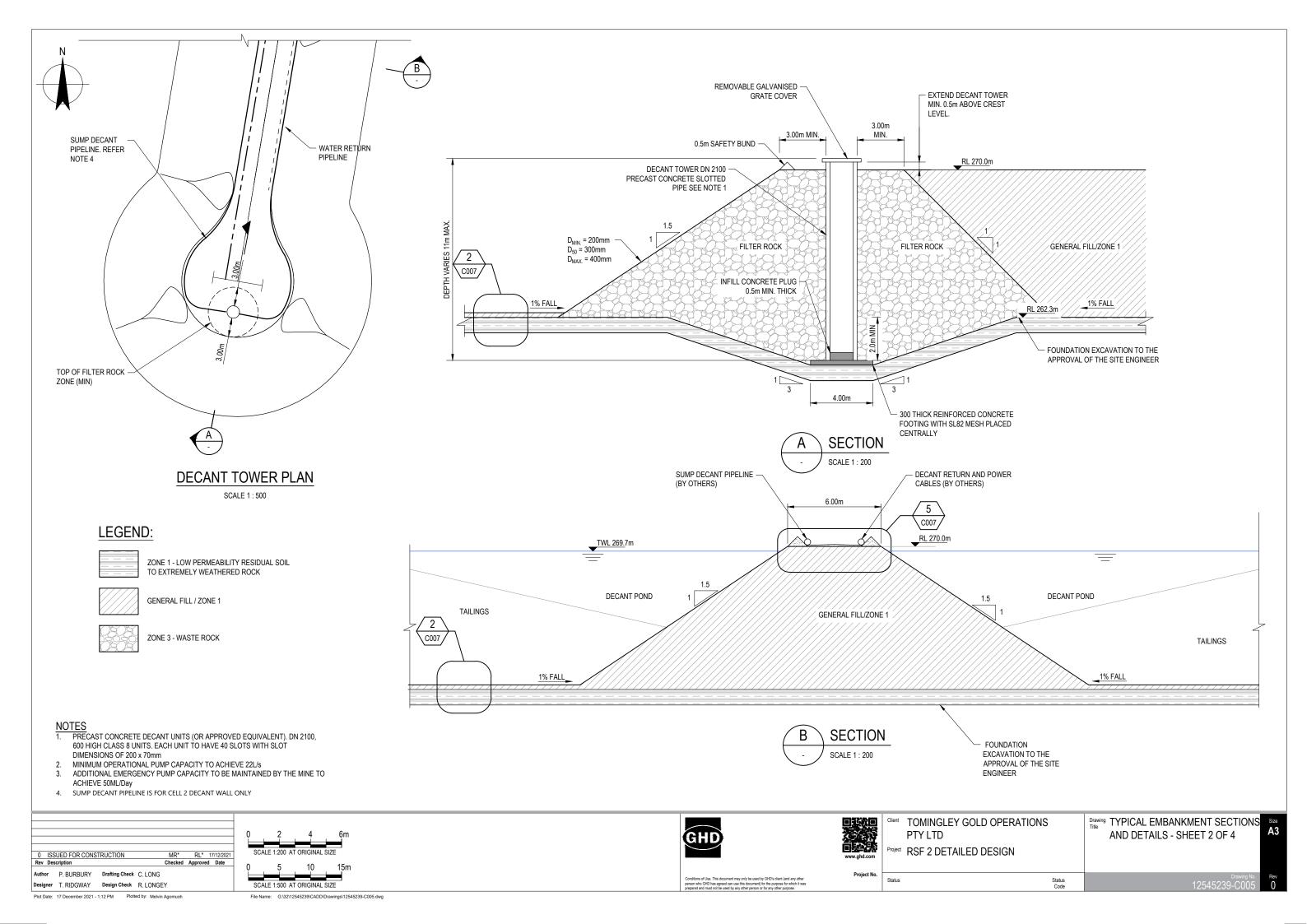
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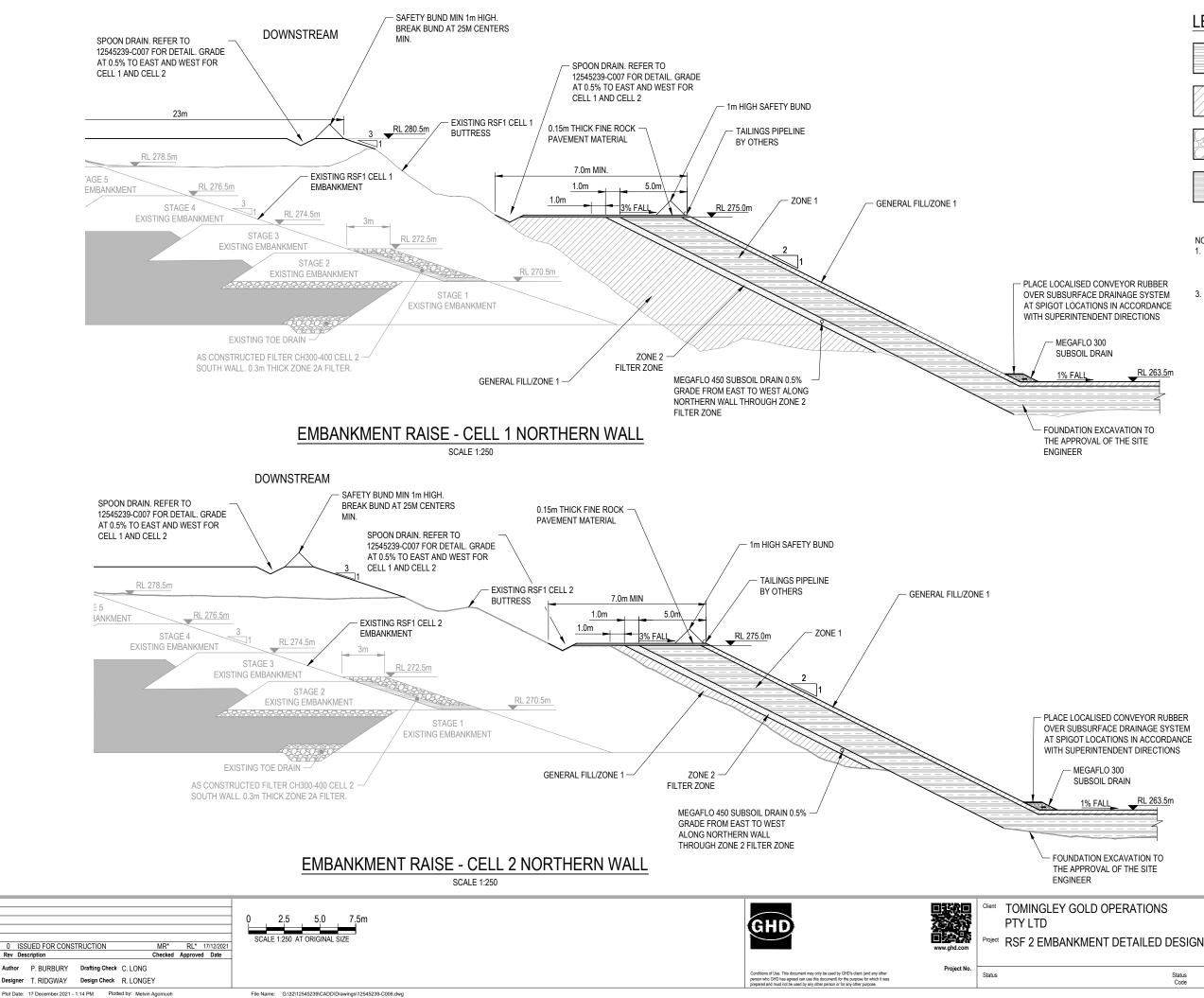
BOREHOLE SETOUT POINT		UT POINTS
POINT	EASTING	NORTHING
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BH03	612974.400	6393189.000
BH04	612790.100	6393360.000
BH05	612841.400	6393664.000
BH06	612974.400	6393714.000
BH07	613320.900	6393651.000
BH08	613552.700	6393609.000
BH09	613785.600	6393577.000
BH10	613739.800	6393286.000

O OPERATIONS	Drawing GEOTECHNICAL INVESTIGATION PLAN	Size A3
DESIGN		
Status Code	Drawing No. 12545239-C003	Rev 0



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LEGEND:



ZONE 1 - LOW PERMEABILITY RESIDUAL SOIL TO EXTREMELY WEATHERED ROCK



GENERAL FILL / ZONE 1



ZONE 3 - WASTE ROCK

ZONE 2 FILTER ZONE

NOTES

- PRIOR TO AND DURING THE PLACEMENT AND 1
- COMPACTION, ALL MATERIALS SHALL BE TESTED AND IN ACCORDANCE WITH THE TECHNICAL SPECIFICATIONS AND ASSESSED BY A GEOTECHNICAL ENGINEERS. EXISTING RSF 1 SEEPAGE OUTLETS TO BE REDIRECTED 3
- AND CONNECTED INTO THE NORTH WEST COLLECTION SUMP

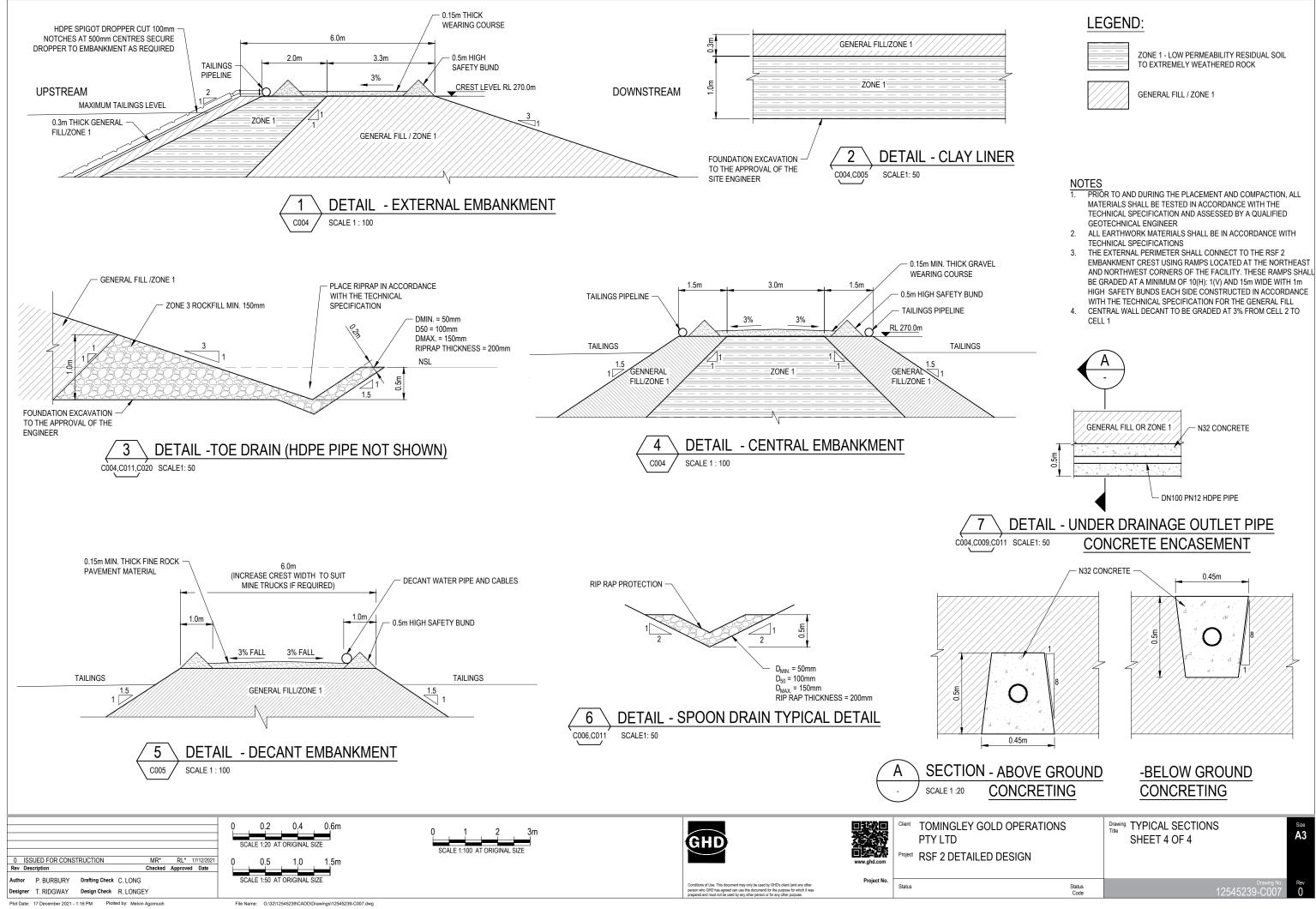
RL 263.5m

OPERATIONS	

Drawing TYPICAL SECTIONS SHEET 3 OF 4

Status Code

Size A3







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TABLE 1 - MATERIAL CONSTRUCTION REQUIREMENTS

MATERIAL	DESCRIPTION	CONSTRUCTION REQUIREMENTS
ZONE 1	EXTREMELY WEATHERED SANDY CLAY OR CLAYEY SAND MATERIAL WON FROM THE FOUNDATION EXCAVATIONS, MEETING SPECIFICATIONS SETOUT IN TABLE 2.	 SPREAD IN LAYERS 150mm THICK AFTER COMPACTION. COMPACT WITH MINIMUM 6 PASSES OF A 10 TONNE VIBRATING PAD FOOT ROLLER TO NO LESS THAN 98% MAX. DRY DENSITY (MDD). SANDY CLAY TO BE PLACED BETWEEN 1% DRY & 3% WET OF OPTIMUM MOISTURE CONTENT (OMC)
ZONE 1/GENERAL FILL	EXTREMELY WEATHERED SANDY CLAY OR CLAYEY SAND TO GRAVELLY CLAY MATERIAL WON FROM THE FOUNDATION EXCAVATIONS, MEETING SPECIFICATIONS SET OUT IN TABLE 2.	 SPREAD IN LAYERS 300mm THICK AFTER COMPACTION COMPACT WITH MINIMUM 6 PASSES OF A 10 TONNE VIBRATING PAD FOOT ROLLER NO LESS THAN 95% MAX. DRY DENSITY (MDD). SANDY CLAY TO BE PLACED BETWEEN 1% DRY AND 3% WET OF OPTIMUM MOISTURE CONTENT (OMC)
ZONE 2	SAND MATERIAL SOURCED FROM CRUSHED ROCK IN ACCORDANCE WITH THE MATERIAL SPECIFICATIONS SET OUT IN SECTION 6.3 OF THE TECHNICAL SPECIFICATION.	FILTER SAND TO BE PLACED IN ACCORDANCE WITH THE SPECIFICATION SET OUT IN SECTION 6.3 OF THE TECHNICAL SPECIFICATION.

TABLE 2 - GHD SPECIFICATION

TABLE 3 - FOUNDATION PREPARATION

MATERIAL

STORAGE FOUNDATION

TABLE 4 - MATERIAL TESTING REQUIREMENTS FOR SANDY CLAY/ CLAYEY SAND

REQUIRED PROPERTIES OF ZONE 1 FILL AFTER COMPACTION MINIMUM % BY WEIGHT PASSING 0.075mm AS 1152 SIEVE 30 MINIMUM % BY WEIGHT PASSING 0.3mm AS 1152 SIEVE 70 MINIMUM % BY WEIGHT PASSING 1.18mm AS 1152 SIEVE 80 MINIMUM PLASTICITY INDEX (%) (AS 1289.3.3.1) 15 LAYER THICKNESS 150mm PERMEABILITY K < 1 X 10⁽⁻⁹⁾ m's (SATURATED) REQUIRED PROPERTIES OF GENERAL FILL AFTER COMPACTION 20 (REFER NOTE 2) MINIMUM % BY WEIGHT PASSING 0.075mm AS 1152 SIEVE MINIMUM % BY WEIGHT PASSING 0.3mm AS 1152 SIEVE 40 MINIMUM % BY WEIGHT PASSING 1.18mm AS 1152 SIEVE 60 LAYER THICKNESS 300mm PERMEABILITY K < 1 X 10⁽⁻⁷⁾ m/s (SATURATED). (REFER NOTE 2).

TEST DETAILS FIELD DENSITY & HILF COMPACTION OR STANDARD COMPACTION ATTERBERG LIMITS PARTICLE SIZE DISTRIBUTION EMERSON CLASS TESTING - PROCESS WATER EMERSON CLASS TESTING - FRESH WATER TRIAXIAL PERMEABILITY EMBANKMENT FOUNDATIO

PUSH TUBE SAMPLING AND TRIAXIAL TESTING

NOTE 2: IF MATERIALS MEET THE REQUIRED PERMEABILITY SPECIFICATION THEN THIS CAN SUPERCEDE THE REQUIRED GRADING SPECIFICATION.

	0 ISSUED FOR CONSTRUCTION MR* RL* 17/12/2021 Rev Description Checked Approved Date	2 Salamanca Square Hobart TAS 7000 Australia GPO Box 667 Hobart TAS 7001 T 61 3 6210 0600 F 61 3 6210 0601 E hbamail@ghd.com W www.ghd.com	WWW.ghd.com	Client TOMINGLEY GOLD OPERATIONS PTY LTD Project RSF 2 DETAILED DESIGN	Drawing MATERIALS SPECIFICATION
Designer T. RIDGWAY Design Check R. LONGEY Status Status Status Code 12545239-C	Author P. BURBURY Drafting Check C. LONG Designer T. RIDGWAY Design Check R. LONGEY		Project No.	Status Status Code	Drawing N 12545239-C00

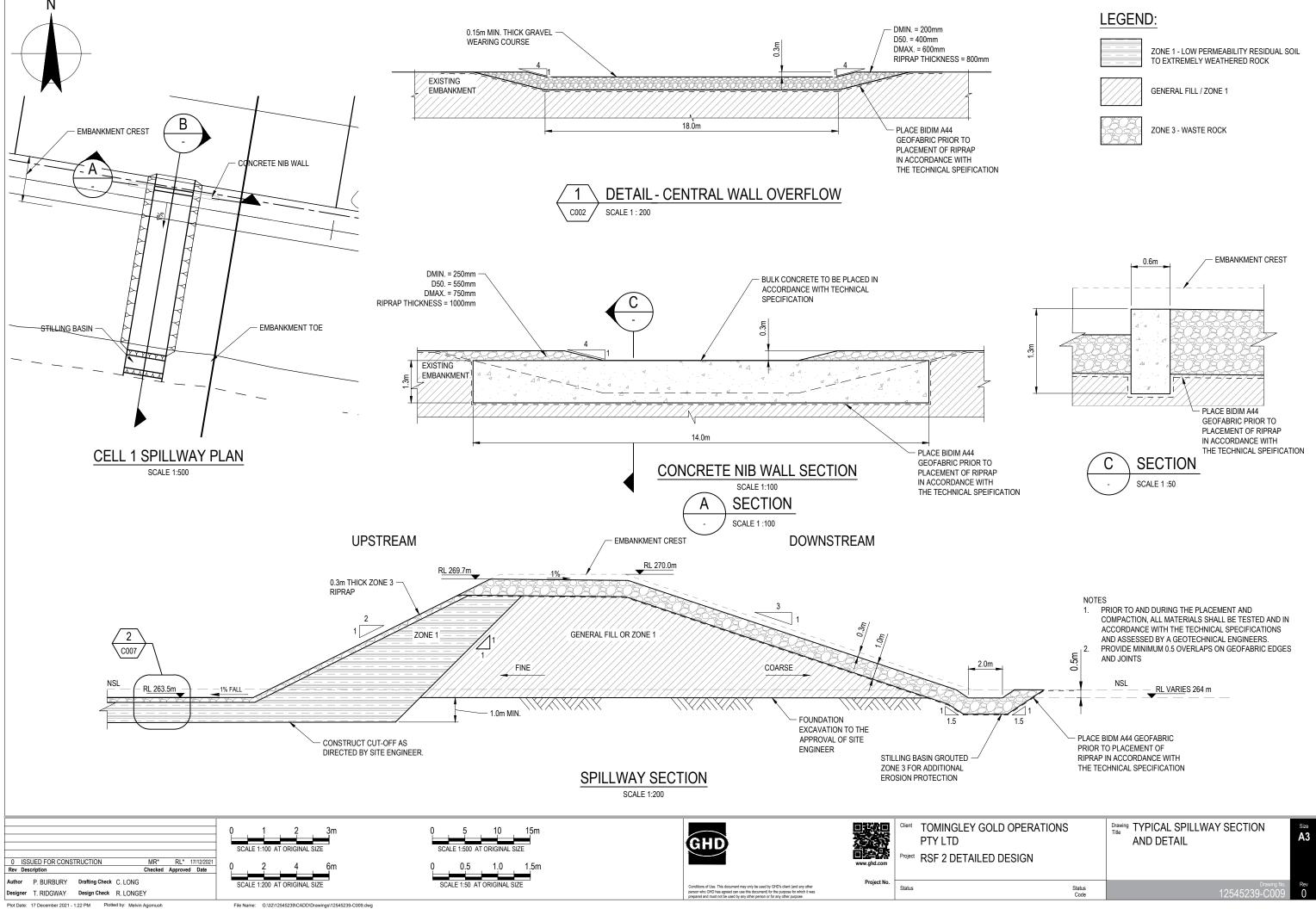
REQUIREMENTS

STORAGE FOUNDATION TO BE PREPARED IN ACCORDANCE WITH THE TECHNICAL SPECIFICATION

MINIMUM TEST FREQUENCY/ VOLUME PLACE (m³)

	EVERY 300mm AND 1,000m ³ /5,000m ³)			
	1,000m ³ /5,000m ³			
	1,000m ³ /5,000m ³			
	5,000m ³ /10,000m ³			
	5,000m ³ /10,000m ³			
	10,000m ³ /20,000m ³			
	5,000m ³ /10,000m ³			
ON	ON TESTING REQUIREMENTS			

1 EVERY 500m

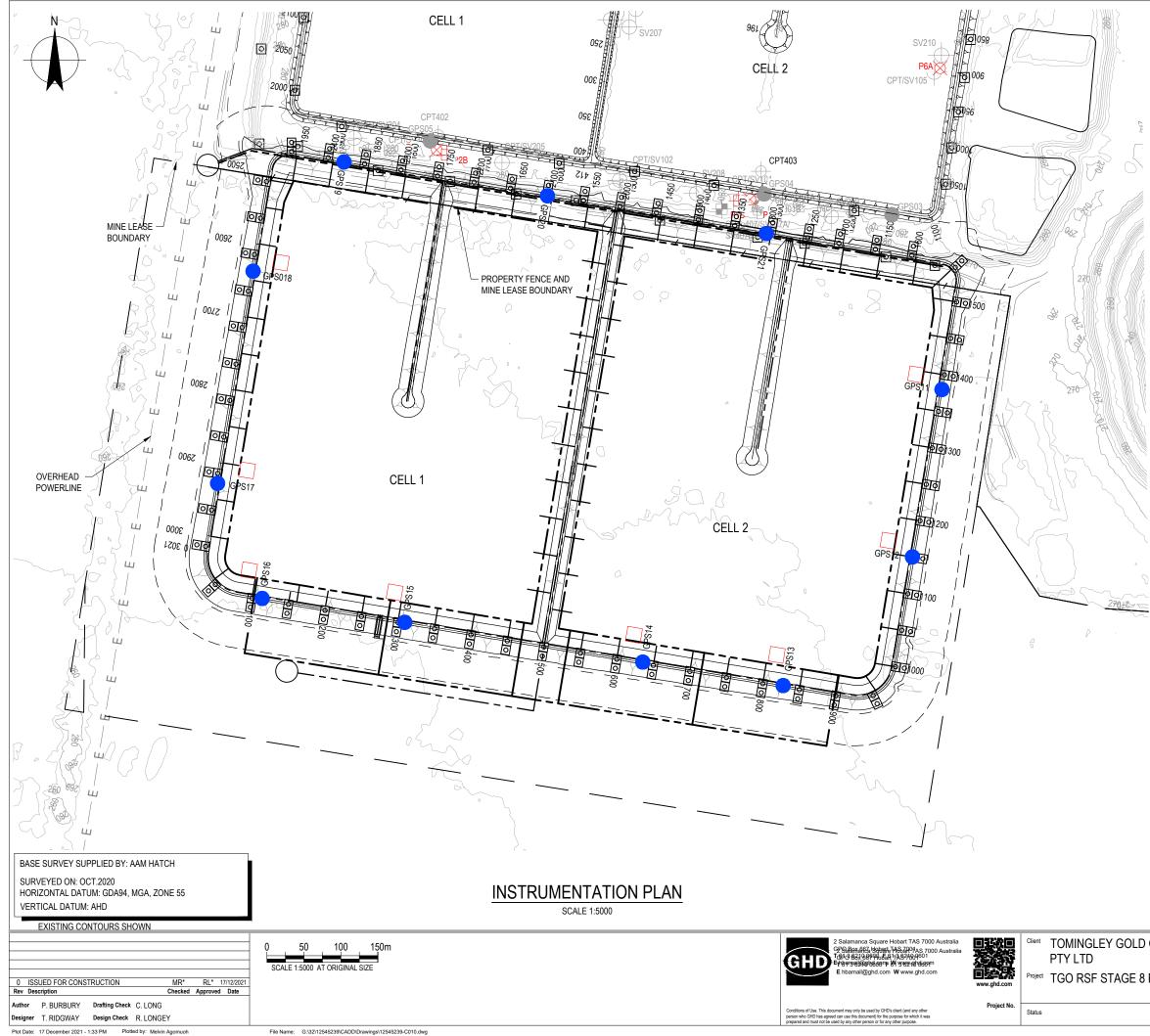


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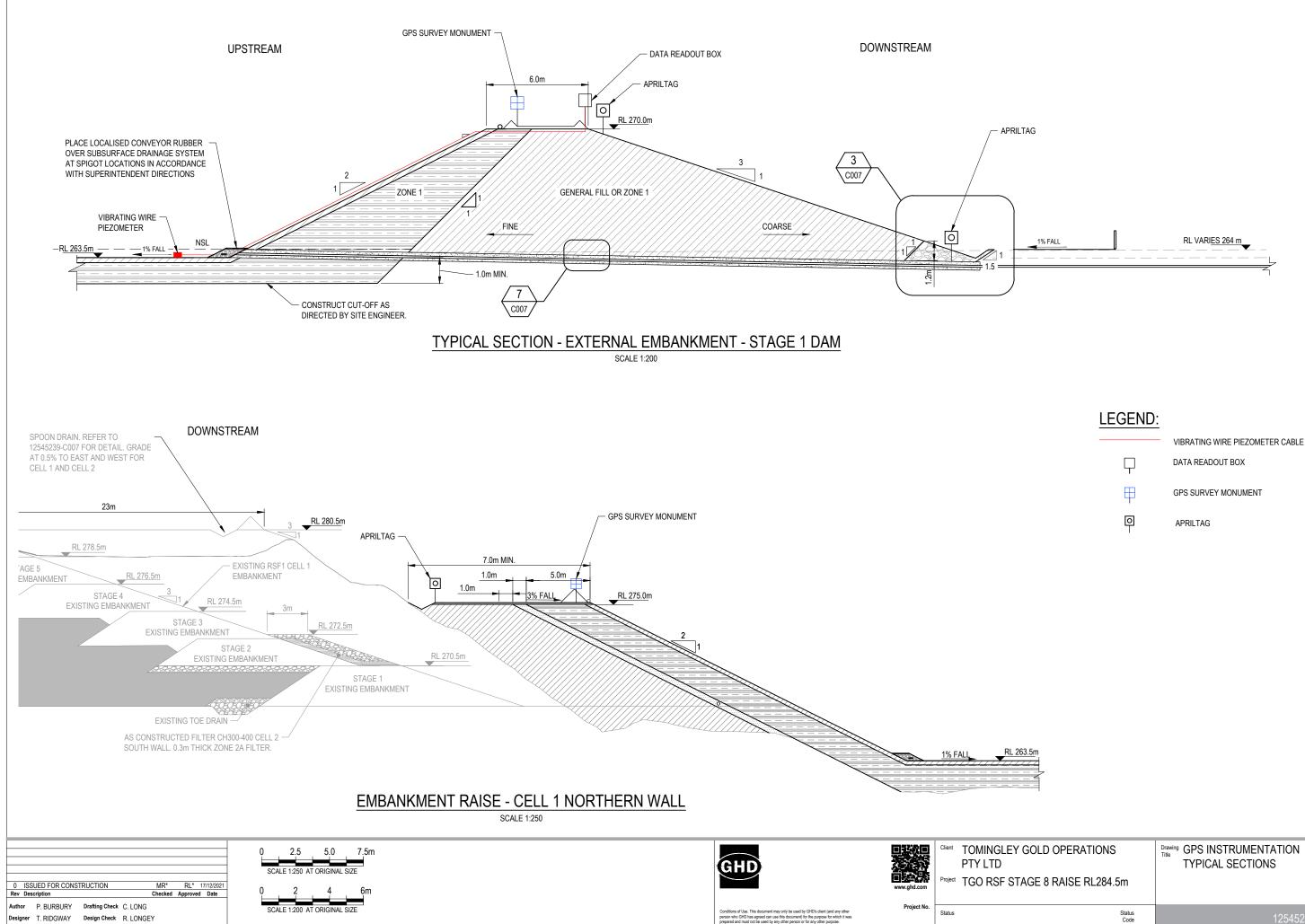
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GPS13	613558.114	6393094.898
GPS14	613368.352	6393126.634
GPS15	613046.595	6393180.445
GPS16	612853.942	6393212.664
GPS17	612793.650	6393368.276
GPS018	612841.610	6393655.048
GPS19	612964.450	6393802.685
GPS20	613239.498	6393756.948
GPS21	613535.455	6393705.913

LEGEND:

	VIBRATING WIRE PIEZOMETER CABLE
\boxtimes	SET A PIEZOMETERS - PUSH IN FROM STAGE 4 CREST
+	SET B PIEZOMETERS - PUSH IN FROM BUTTRESS
0	APRILTAG LOCATION
GPSXX	GPS SURVEY MONUMENT - NEW
GPSXX	GPS SURVEY MONUMENT - EXISTING
	VIBRATING WIRE PIEZOMETER

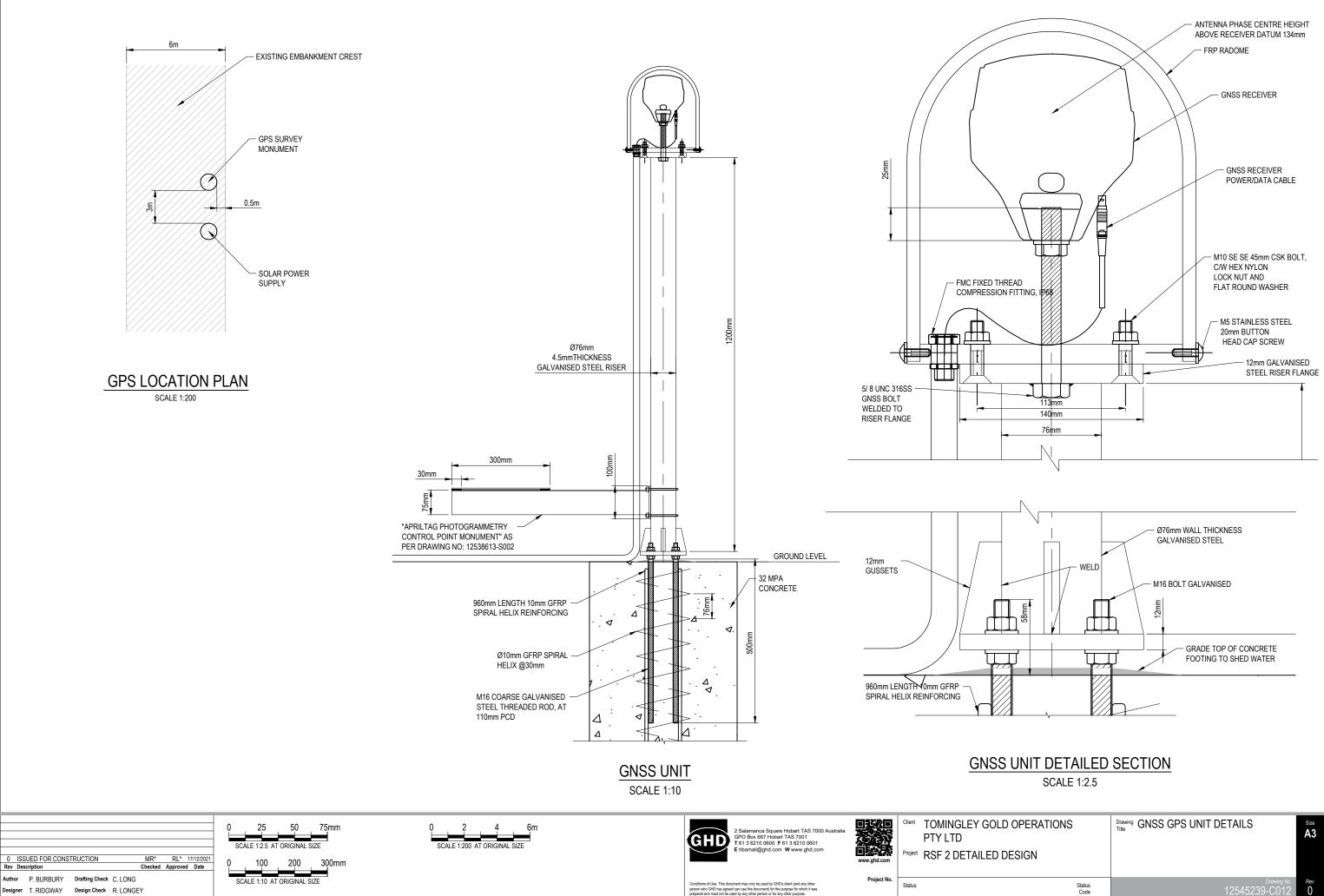
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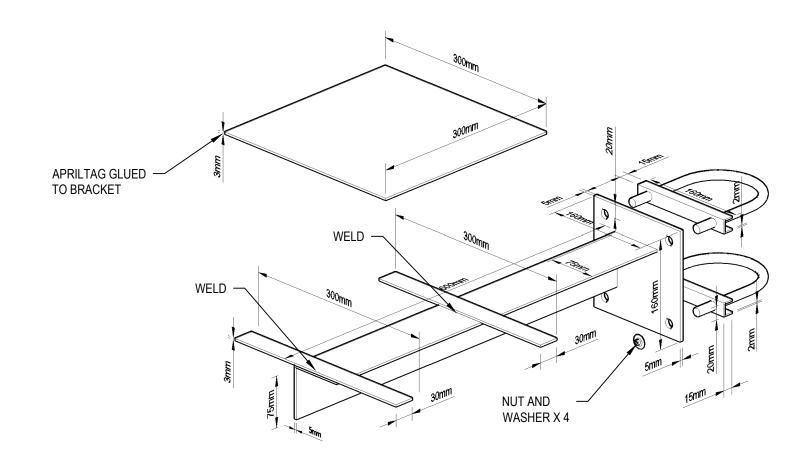
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D OPERATIONS	Drawing GPS INSTRUMENTATION TYPICAL SECTIONS	Size A3
8 RAISE RL284.5m		
Status Code	Drawing No. 12545239-C011	Rev 0
	12545239-C011	0



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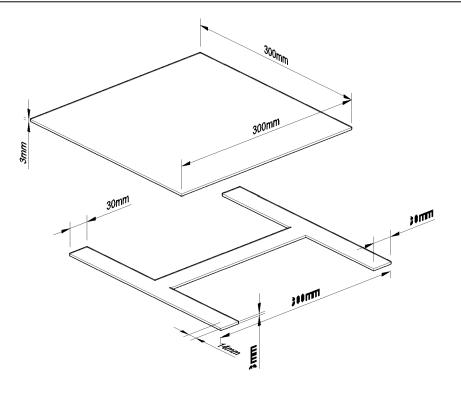


SURVEY REFERENCE APRIL TAG MOUNT

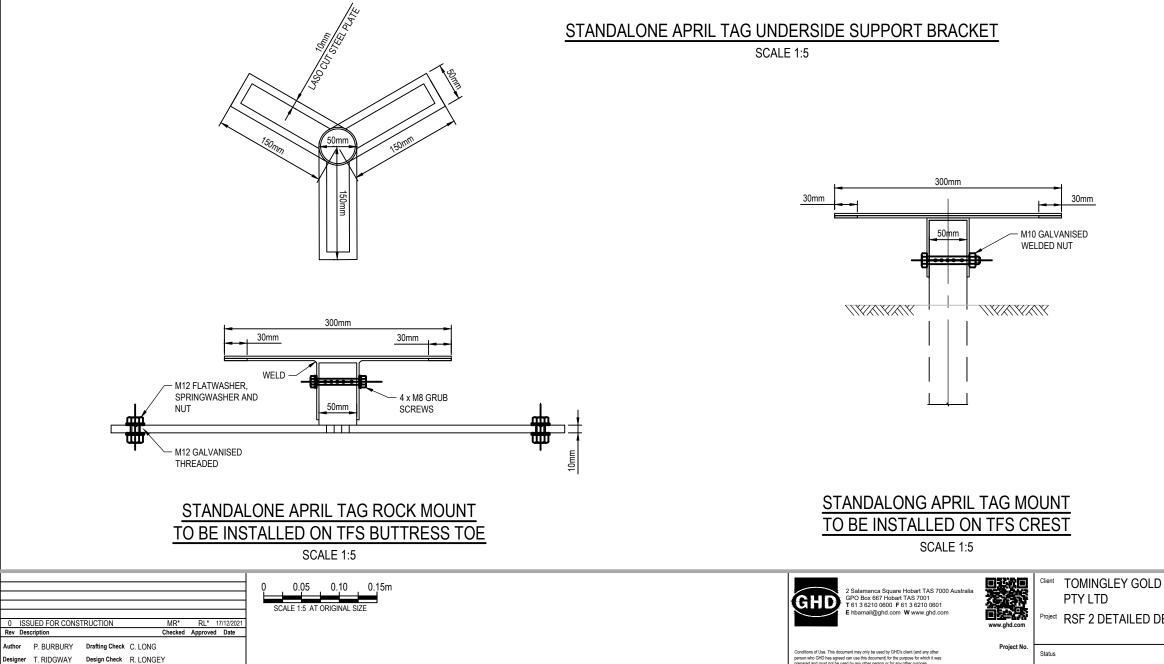
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O OPERATIONS	Drawing SURVEY REFERENCE APRIL TAG MOUNT	Size A3
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O OPERATIONS	Drawing STAND ALONE APRIL TAG MOUNT DETAILS	Size A3
DESIGN		
Status Code	Drawing No. 12545239-C014	Rev 0

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2 Salamanca Square Hobart TAS 7000 Australia GPO Box 667 Hobart TAS 7001 T61 36210 0600 F61 36210 0601 E hbamail@ghd.com W www.ghd.com	Www.ghd.com	Client Project	TOMINGLEY GOLD PTY LTD RSF 2 DETAILED DE
Conditions of Use. This document may only be used by GHD's client (and any other person who GHD has agreed can use this document) for the purpose for which it was prepared and must not be used by any other person or for any other purpose.	Project No.	Status	

CHAINAGE	600.00	620.00	660.00	680.00	700.00	720.00	740.00	760.00	780.00	800.00	820.00	840.00	860.00	880.00	00.006	920.00 921.31 940.00	960.00	980.00	999.85	1020.00	000101
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Author P. BURBURY Drafting Check C. LONG Designer T. RIDGWAY Design Check R. LONGEY													person w	ho GHD has agreed o	can use this docume	d by GHD's client (and any other nt) for the purpose for which it was n or for any other purpose.	P	Project No.	Status		

DATUM RL. 234.00																															
LEVEL DIFFERENCE CUT - / FILL +	5.92	5.82	5.74	5.69	5.69	5.63	5.56	5.49	5.39	5.29	5.27	5.21	5.18	5.05	5.02	4.98	4.90	4.72	4.75	2.46	4.20	3.91	2.35	4.64	4.67	4.64	4.60	4.63	4.33	4.49	
DESIGN SURFACE LEVEL	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000-	
	264.08	264.18	264.26	264.31	264.31	264.37	264.44	264.51	264.61	264.71	264.73	264.79	264.82	264.95	264.98	265.02	265.10 265.10		265.25	267.54	265.80	266.09	267.65	265.36	265.33	265.36	265.40	265.37	265.67	265.51	
CHAINAGE	600.00	620.00	640.00	660.00	680.00	700.00	720.00	740.00	760.00	780.00	800.00	820.00	840.00	860.00	880.00	00.006	920.00	940.00	960.00	980.00	999.85	1020.00	1040.00	1060.00	1080.00	1100.00	1120.00	1140.00	1160.00	1180.00	

DESIGN SURFACE —

EXTERNAL EMBANKMENT - LONGITUDINAL SECTION (0 - 600) SCALE 1:2000

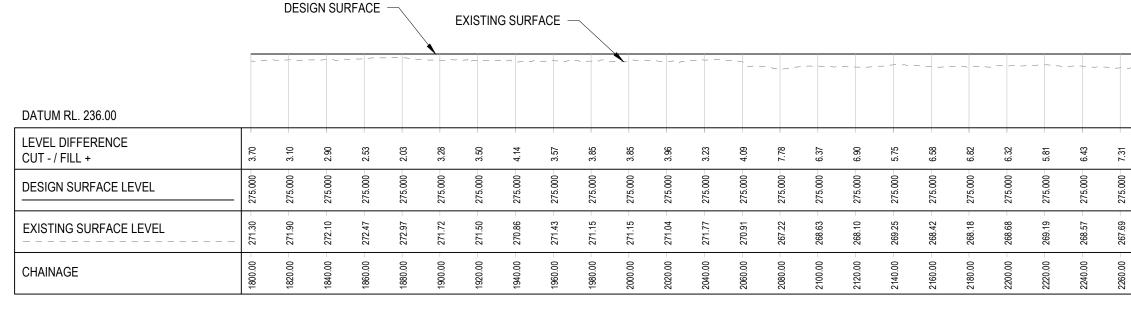
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270.000	270.000	270.000	270.000	270.000	270.000 270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000
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- - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - - -	EXISTING SURFACE EXISTING SURFACE 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012 0000012	EXISTING SURFACE 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2 000001/2	EXISTING SURFACE 61:9 00000/Z 62:9 00000/Z 99:8:8 0000/Z 99:8:8 0000/Z 99:8:8 0000/Z 99:8:8 0000/Z 99:8:8 00000/Z </td <td>EXISTING SURFACE EXISTING SURFACE EXISTING</td> <td>EXISTING SURFACE EXISTING SURFACE 0099 00001Z 0099 00001Z 0099 00001Z 0099 00001Z 0199 000001Z 01982 0</td>	EXISTING SURFACE EXISTING	EXISTING SURFACE EXISTING SURFACE 0099 00001Z 0099 00001Z 0099 00001Z 0099 00001Z 0199 000001Z 01982 0

GOLD OPERATIONS	Drawing LONGITUDINAL SECTIONS SHEET 1 0F 5	Size A3
ILED DESIGN		
Status Code	Drawing No. 12545239 - C015	Rev 0

		DESIG	N SURF	ACE -		EX	ISTING	SURFA	CE —														~ ~ ~ ~	~			RL 275.	<u>0m</u>		- + -	
DATUM RL. 235.00									_ + _		\								<												
LEVEL DIFFERENCE CUT - / FILL +	4.40	4.36	4.29	4.13	4.02	3.94	3.86	3.82	3.69	3.64	3.57	3.48	3.37	3.24	3.07	3.12 4.86	7.35	7.88 7.80	ע די ע	4.72 2.86	2.93	4.36	4.57	3.88	4.43	2.38	3.11	2.97	2.90	2.98	3.70
DESIGN SURFACE LEVEL	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000 271.875	274.375	275.000 275.000	275 000	275.000 275.000 275.000	275.000	275.000	275.000	275.000	275.000	275.000	275.000	275.000	275.000	275.000	275.000
EXISTING SURFACE LEVEL	265.60	265.64	265.71	265.87	265.98	266.06	266.14	266.18	266.31	266.36	266.43	266.52 -	266.63 -	266.76	266.93	266.88 - 267.02 -	267.03	267.12 - 267.20 -	060 <u>4</u> 4	270.28 - 272.14 -	272.07	270.64	270.43	271.12 -	270.57	272.62	271.89	272.03	272.10	272.02	271.30
CHAINAGE	1200.00	1220.00	1240.00	1260.00	1280.00	1300.00	1320.00	1340.00	1360.00	1380.00	1400.00	1420.00	1440.00	1460.00	1480.00	1485.00 - 1500.00 -	1520.00	1525.00 - 1540.00 -	1560.00	1564.29 - 1580.00 -	1600.00	1620.00	1640.00	1660.00	1680.00	1700.00	1720.00	1740.00	1760.00	1780.00	1800.00

EXTERNAL EMBANKMENT - LONGITUDINAL SECTION (1200 - 1800)

SCALE 1:2000



EXTERNAL EMBANKMENT - LONGITUDINAL SECTION (1800 - 2400)

SCALE 1:2000

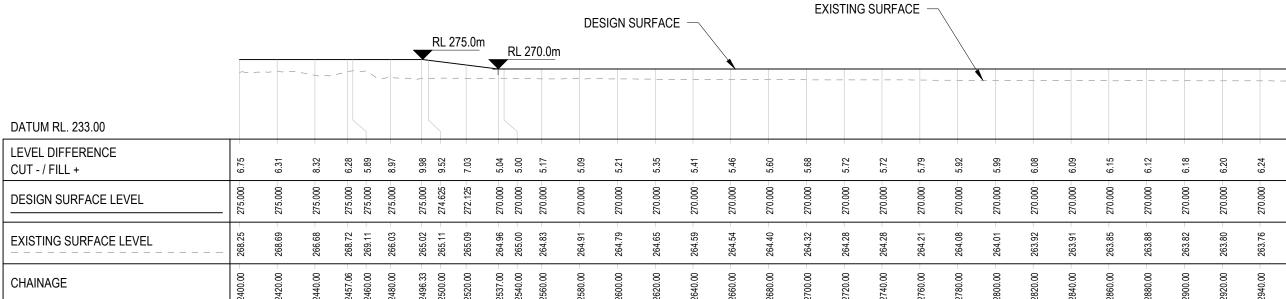


	7.32 - 7.46 - 7.46 - 7.40 - 7.40 - 6.70 - 6.75 - 6.75 - 6.75 - 6.75 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.60 - 7.	275.000 - 275.000 - 275.000 - 275.000 - 275.000 -	267.68 - 267.54 - 268.30 - 267.60 - 268.25 -	2320.00 - 2340.00 - 2360.00 - 2380.00 - 2400.00 -
	7.46	275.000	267.54	340.00 -
~	7.32	275.000	267.68	320.00
~ ~ ~	7.87	275.000	267.13	2300.00
	7.39	275.000	267.61	2280.00
	7.31	275.000	267.69	2260.00

RL 275.0m

D OPERATIONS	Drawing LONGITUDINAL SECTIONS SHEET 2 0F 5	Size A3
DESIGN		
Status Code	Drawing No. 12545239 - C016	Rev 0

														5.4	5.6	5.6			5.7	5.6	5.6	6.0	6.0	
DESIGN SURFACE LEVEL	275.000	275.000	275.000	275.000 - 275.000 -	275.000	275.000 274.625	272.125	270.000 270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000
EXISTING SURFACE LEVEL	268.25	268.69	266.68	268.72 269.11	266.03 -	265.02 265.11	265.09	264.96 265.00	264.83	264.91	264.79	264.65	264.59	264.54	264.40	264.32	264.28	264.28	264.21	264.08	264.01	263.92	263.91	263.85
CHAINAGE	2400.00	2420.00	2440.00	2457.06 2460.00	2480.00	2496.33 2500.00	2520.00	2537.00 - 2540.00 -	2560.00	2580.00	2600.00	2620.00	2640.00	2660.00	2680.00	2700.00	2720.00	2740.00	2760.00	2780.00	2800.00	2820.00	2840.00	2860.00
								<u>EX</u>	TERN	NAL E	MBAN	NKME	ENT -	LONC SCALE		DINAL	SEC	TION	(2400) - 302	<u>21)</u>			
SUED FOR CONSTRUCTION MR* RL* 17/12/2021	0 SCA	20 ALE 1:200 J	40 AT ORIGINA	60m SIZE											G	HD 9	2 Salamanca Sr 3PO Box 667 F 761 3 6210 060 1 3 6210 060	obart TAS 700	0 0601			PT	TY LTD	EY GOLD



OPERATIONS	Drawing LONGITUDINAL SECTIONS SHEET 3 0F 5	Size A3
DESIGN		
Status Code	Drawing No. 12545239 - C017	Rev 0

6.12	6.18	6.20	6.24	6.28	6.34	6.50	6.59 6.60
270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000 - 270.000 -
263.88	263.82	263.80	263.76	263.72	263.66	263.50 -	263.41 - 263.40 -
2880.00	2900.00	2920.00	2940.00	2960.00	2980.00	3000.00 -	3020.00 - 3021.49 -

			URFACE LEVEL	FACE LEVEL	DESIG	N SURFACE LEVEL		,									
DATUM RL. 248.00				+											=		
EXCAVATION SURFACE LEVEL	275.00	265.00	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60 -	264.60	264.60
LEVEL DIFFERENCE CUT - / FILL +	5.70	7.22	4.37	4.35	4.39	4.56	4.64	4.75	4.84	4.94	4.95	5.01	5.05	5.10	5.11	5.14	5.17
DESIGN SURFACE LEVEL	275.000	272.500	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000
EXISTING SURFACE LEVEL	269.30	265.28	265.63	265.65	265.61	265.44	265.36	265.25	265.16	265.06	265.05	264.99	264.95	264.90	264.89	264.86	264.83
CHAINAGE	0.00	20.00	40.00	60.00	80.00	100.00	120.00	140.00	160.00	180.00	200.00	220.00	240.00	260.00	280.00	300.00	320.00

DIVIDING WALL EMBANKMENT - LONGITUDINAL SECTION

SCALE 1:1000

264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	264.60	270.00
5.10	5.11	5.13	5.20	5.28	5.36	5.49	5.41	5.63	5.73	5.84	0.00	6.13	6.17
270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000	270.000
264.90	264.89	264.87	264.80	264.72	264.64	264.51	264.59	264.37	264.27	264.16	264.00	263.87	263.83
340.00	360.00	380.00	400.00	420.00	440.00	460.00	480.00	500.00	520.00	540.00	560.00	580.00	593.17
	00 - 264.90 - 270.000 - 5.10 - 264.60	0.00 - 264.90 - 270.000 - 5.10 - 264.60	0.00 - 264.90 - 270.000 - 5.10 - 264.60	0.00 264.90 270.000 5.10 264.60 0.00 264.89 270.000 5.11 264.60 0.00 264.87 5.13 264.60 0.00 264.87 5.13 264.60 0.00 264.87 5.13 264.60 0.00 264.80 5.13 264.60	0.00 264.90 270.000 5.10 264.60 - 0.00 264.89 270.000 5.11 264.60 - 0.00 264.87 270.000 5.13 264.60 - 0.00 264.87 270.000 5.13 264.60 - 0.00 264.87 270.000 5.13 264.60 - 0.00 264.80 5.13 264.60 - - 0.00 264.80 5.20 5.20 264.60 - 0.00 264.72 270.000 5.20 264.60 - -	0.00 264.90 270.000 5.10 264.60 0.00 264.89 270.000 5.11 264.60 0.00 264.87 270.000 5.13 264.60 0.00 264.87 270.000 5.13 264.60 0.00 264.87 270.000 5.13 264.60 0.00 264.72 270.000 5.28 264.60 0.00 264.72 270.000 5.28 264.60 0.00 264.64 5.28 264.60 0.00 264.64 5.36 - 264.60	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	0.00 264.59 270.000 5.10 264.60 - 0.00 264.87 270.000 5.11 264.60 - 0.00 264.87 270.000 5.13 264.60 - 0.00 264.87 270.000 5.13 264.60 - 0.00 264.87 270.000 5.13 264.60 - 0.00 264.87 270.000 5.26 264.60 - 0.00 264.51 270.000 5.26 264.60 - 0.00 264.51 270.000 5.26 264.60 - 0.00 264.51 270.000 5.28 264.60 - 0.00 264.51 270.000 5.49 264.60 - 0.00 264.51 270.000 5.49 264.60 - 0.00 264.59 5.49 264.60 - - 0.00 264.59 5.49 264.60 - -	00 264.50 270.000 5.10 264.60 00 264.87 270.000 5.11 264.60 00 264.87 270.000 5.11 264.60 00 264.87 270.000 5.13 264.60 00 264.72 270.000 5.13 264.60 000 264.64 270.000 5.20 264.60 000 264.51 270.000 5.28 264.60 000 264.51 270.000 5.28 264.60 000 264.51 270.000 5.49 264.60 000 264.51 270.000 5.49 264.60 000 264.51 270.000 5.49 264.60 000 264.59 270.000 5.49 264.60 000 264.59 246.60 - - 000 5.49 2.44.60 - - 000 264.50 - 264.60 - 000 5.49 2.46.60 - - 000 264.59 -	0.0 264.30 270.000 5.10 264.60 29.66 0.0 264.38 270.000 5.11 264.60 - 0.0 264.87 270.000 5.11 264.60 - 0.0 264.87 270.000 5.13 264.60 - 0.0 264.87 270.000 5.13 264.60 - 0.0 264.81 270.000 5.20 264.60 - 0.0 264.51 270.000 5.20 264.60 - 0.0 264.51 270.000 5.26 264.60 - 0.0 264.51 270.000 5.36 264.60 - 0.0 264.51 270.000 5.49 264.60 - 0.0 264.51 270.000 5.41 264.60 - 0.0 264.57 270.000 5.41 264.60 - 0.0 264.50 5.46 - - - 0.0 264.57 270.000 5.41 264.60 - 0.0 264.57	0.00 264.39 270.000 5.10 264.60 - 0.00 264.49 270.000 5.11 264.60 - 0.00 264.89 270.000 5.13 264.60 - 0.00 264.80 270.000 5.13 264.60 - 0.00 264.80 270.000 5.13 264.60 - 0.00 264.51 270.000 5.13 264.60 - 0.00 264.51 270.000 5.13 264.60 - 0.00 264.51 270.000 5.49 264.60 - 0.00 264.51 270.000 5.49 264.60 - 0.00 264.51 270.000 5.49 264.60 - 0.00 264.51 270.000 5.49 264.60 - 0.00 264.51 270.000 5.49 264.60 - 0.00 264.50 5.49 264.60 - - 0.00 264.50 5.49 264.60 - - 0.00 264.6	0.00 2.64.49 270.000 5.10 264.60 0.00 2.64.89 270.000 5.11 264.60 0.00 2.64.89 270.000 5.11 264.60 0.00 2.64.87 270.000 5.13 264.60 0.00 2.64.87 270.000 5.13 264.60 0.00 2.64.51 270.000 5.28 264.60 0.00 2.64.53 270.000 5.49 264.60 0.00 2.64.53 270.000 5.49 264.60 0.00 2.64.53 270.000 5.41 264.60 0.00 2.64.54 270.000 5.49 264.60 0.00 2.64.54 270.000 5.41 264.60 0.00 2.64.53 270.000 5.41 264.60 0.00 2.64.50 264.60 - - 0.00 2.64.50 2.64.60 - - 0.00 2.64.60 5.41 2.64.60 -	0.0 264.90 270.000 5.10 264.60 0.0 264.83 270.000 5.11 264.60 0.0 264.87 270.000 5.13 264.60 0.0 264.87 270.000 5.13 264.60 0.0 264.87 270.000 5.13 264.60 0.00 264.51 270.000 5.13 264.60 0.00 264.51 270.000 5.13 264.60 0.00 264.51 270.000 5.36 264.60 0.00 264.51 270.000 5.49 264.60 0.00 264.51 270.000 5.49 264.60 0.00 264.51 270.000 5.49 264.60 0.00 264.50 264.60 264.60 264.60 0.00 264.50 264.60 264.60 264.60 0.00 264.60 5.38 264.60 264.60 0.00 264.60 264.60 264.60 264.60 <tr< td=""></tr<>

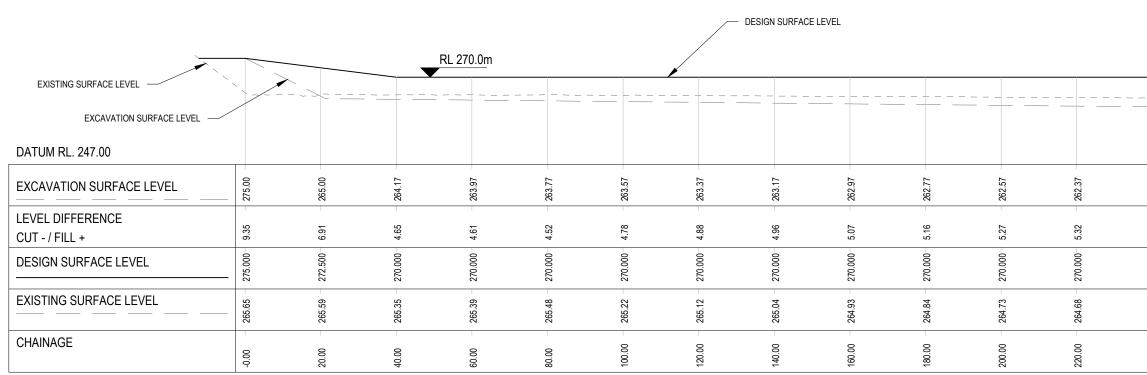
DIVIDING WALL EMBANKMENT - LONGITUDINAL SECTION (CONTINUED)

SCALE 1:1000



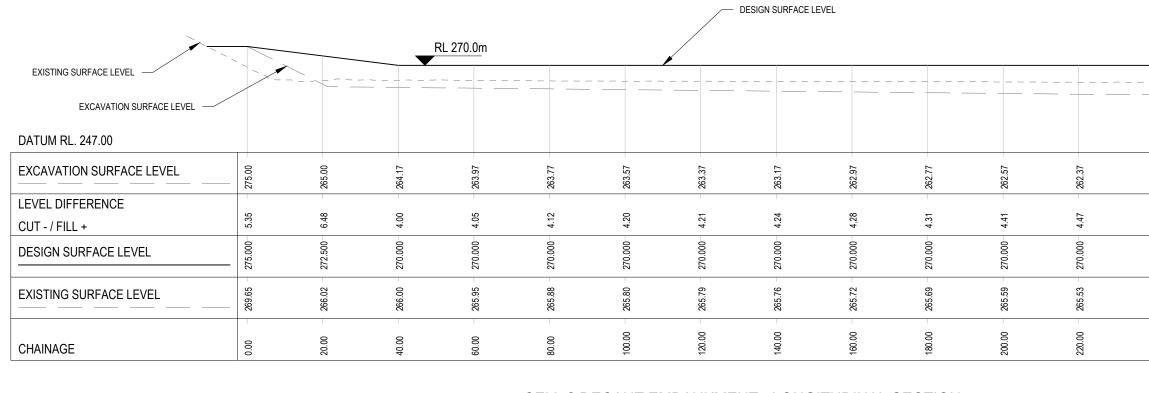
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O OPERATIONS	Drawing LONGITUDINAL SECTIONS SHEET 4 0F 5	Size A3
DESIGN		
Status Code	Drawing No. 12545239 - C018	Rev 0



CELL 1 DECANT EMBANKMENT - LONGITUDINAL SECTION

SCALE 1:1000



CELL 2 DECANT EMBANKMENT - LONGITUDINAL SECTION

SCALE 1:1000

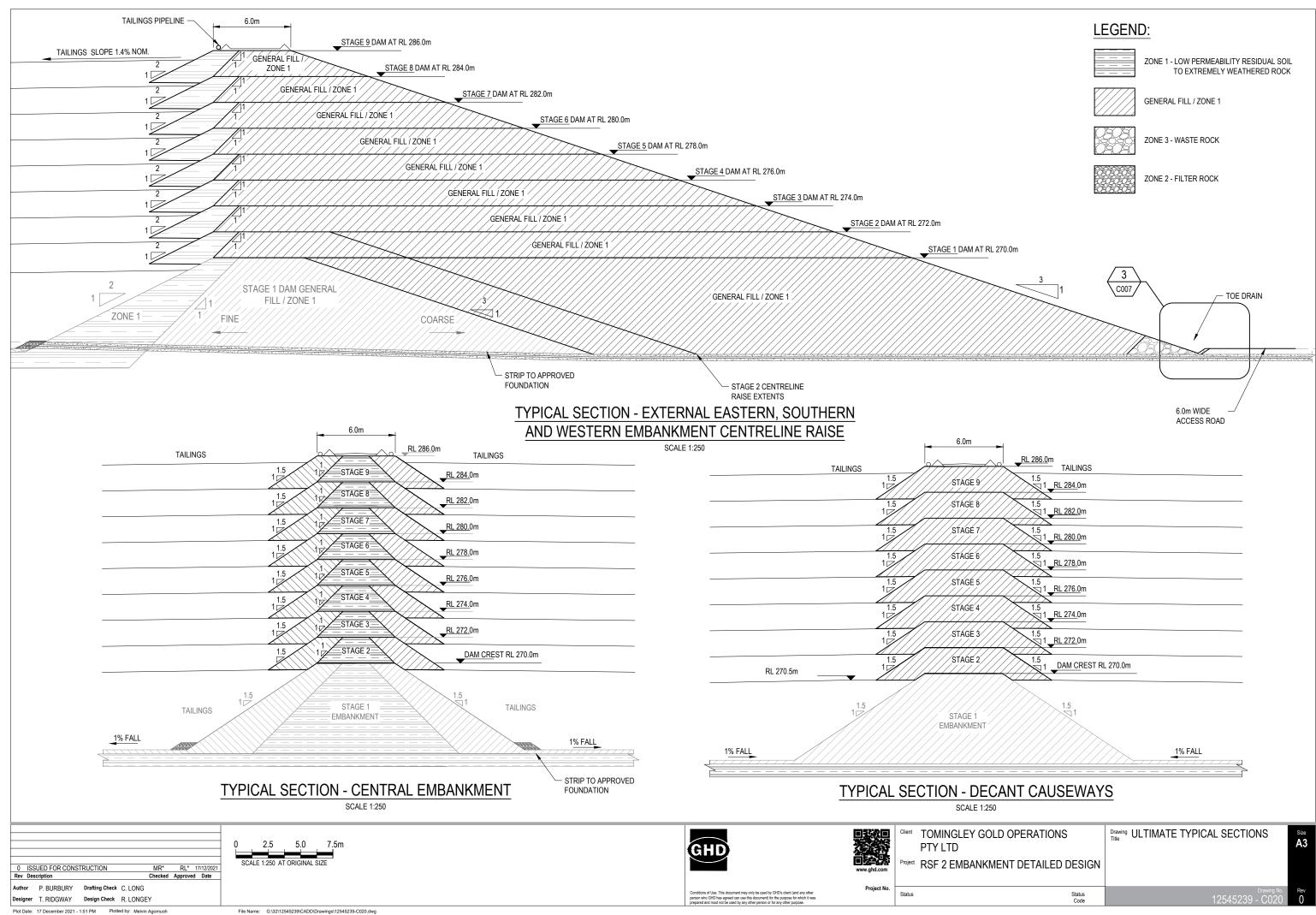


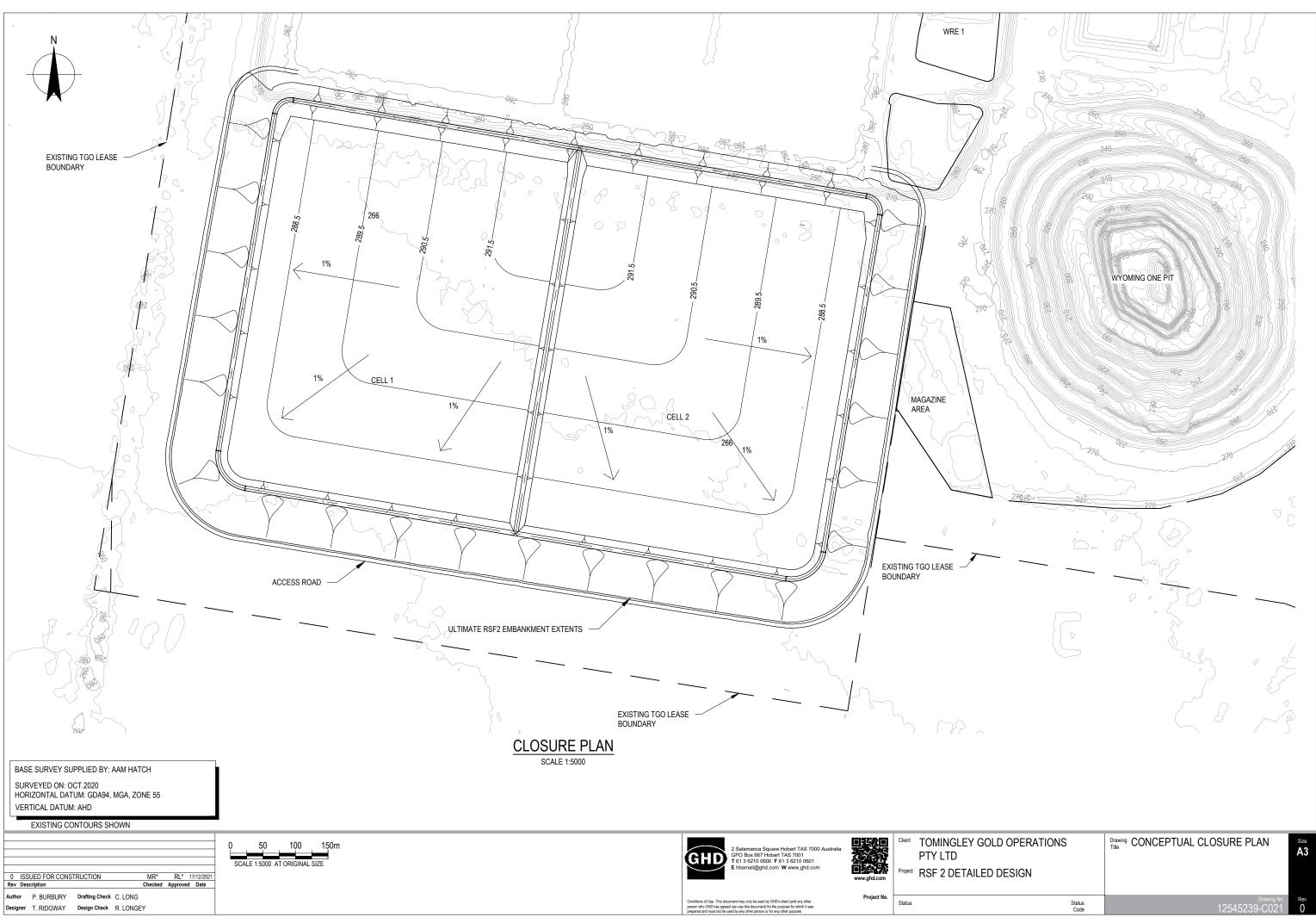
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O OPERATIONS	Drawing LONGITUDINAL SECTIONS SHEET 5 0F 5	Size A3
DESIGN		
Status	Drawing No. 12545239 - C019	Rev

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240.00	260.00	280.00	300.00 301.33

262.30	262.30	262.30	262.30
5.43	5.53	5.69	5.75 5.75
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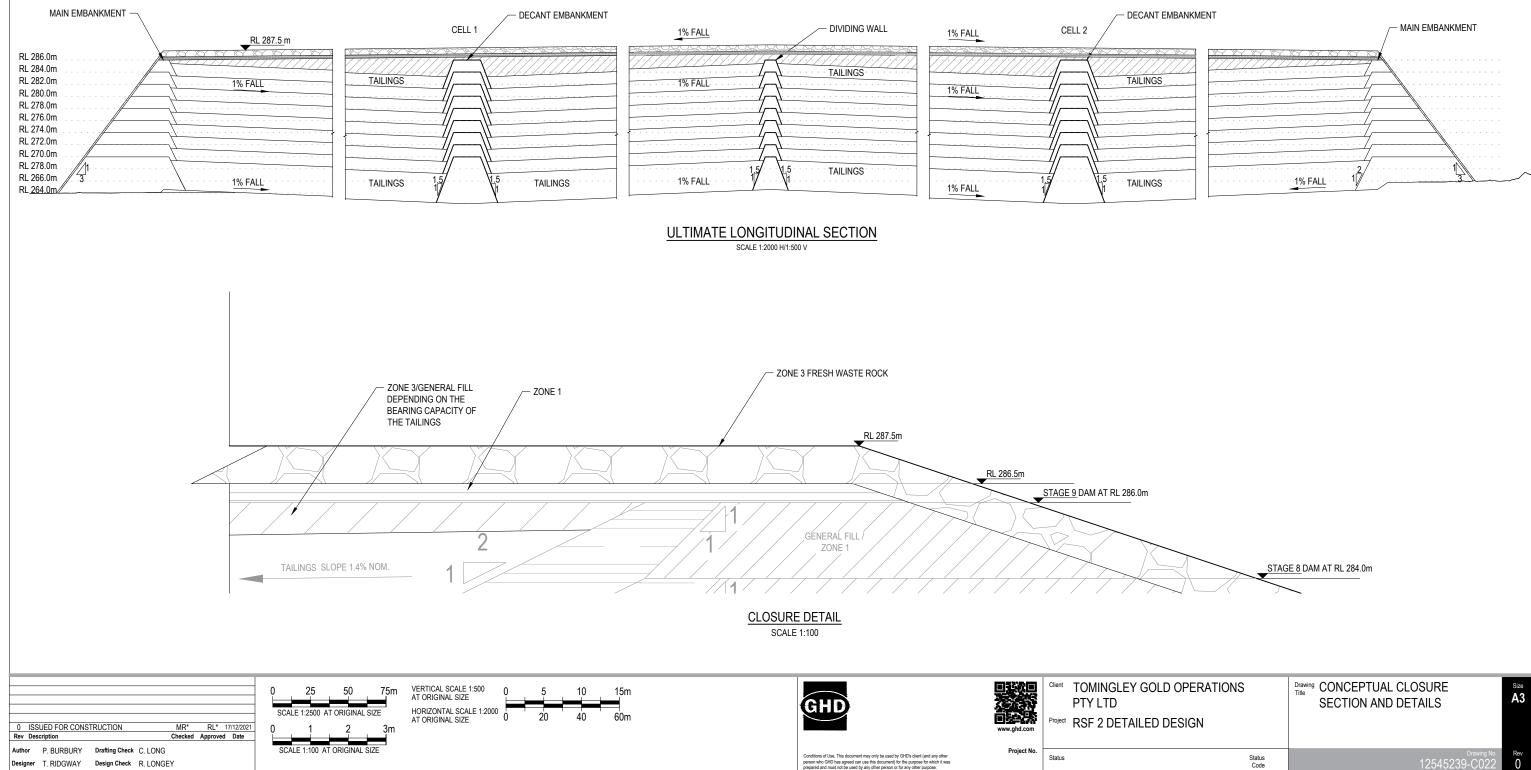


Plot Date: 17 December 2021 - 1:53 PM Plotted by: Melvin Agomuoh

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NOTES 1. DECANT WATER TO BE REMOVED FROM RSF USING EXISTING DECANT INFRASTRUCTURE

- PRIOR TO THE PLACEMENT OF THE CLOSURE MATERIAL, THE TAILINGS ARE TO BE 2. TRACK ROLLED FOR CONSOLIDATION
- WASTE ROCK AND GENERAL FILL IS TO BE USED TO INFILL THE DECANT POND AREA 3. PRIOR TO CAPPING THE RSF LANDFORM ALL CONTAMINATED TAILINGS INFRASTRUCTURE SUCH AS THE TAILINGS DEPOSITION
- 4. PIPELINE IS TO BE DISPOSED IN AN APPROVED LOCATION THE DECANT TOWER IS TO BE INFILLED WITH BULK CONCRETE FOLLOWING THE DECANT POND WATER REMOVAL AND PRIOR TO THE CAPPING OF THE FACILITY 5.
- ADDITIONAL WASTE ROCK IS T BE PLACED ON THE DOWNSTREAM FACE OF THE 6.
- PERIMETER EMBANKMENT FOR EROSION PROTECTION IF REQUIRED THE CAPPING SURFACE AND DOWNSTREAM FACE OF THE EMBANKMENT IS TO BE 7.
- REHABILITATED USING LOCALLY AVAILABLE NATIVE GRASSES



Plot Date: 17 December 2021 - 1:55 PM Plotted by: Melvin Agomuch

File Name: G:\32\12545239\CADD\Drawings\12545239-C022.dwg

LEGEND:



ZONE 1 - LOW PERMEABILITY RESIDUAL SOIL TO EXTREMELY WEATHERED ROCK



GENERAL FILL / ZONE 1



ZONE 3 - WASTE ROCK