



**REPORT**

TO

**MOORE THEOLOGICAL COLLEGE**

ON

**GEOTECHNICAL INVESTIGATION**

FOR

**PROPOSED LIBRARY BUILDING**

AT

**CORNER KING STREET AND CARILLON AVENUE,  
NEWTOWN**

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**Jeffery and Katauskas Pty Ltd**  
CONSULTING GEOTECHNICAL AND ENVIRONMENTAL ENGINEERS

Postal Address: PO Box 976, North Ryde BC NSW 1670  
Tel: 02 9888 5000 • Fax: 02 9888 5001 • Email: [engineers@jkgroup.net.au](mailto:engineers@jkgroup.net.au) • ABN 17 003 550 801



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**TABLE A: SUMMARY OF MOISTURE CONTENT TEST RESULTS**

**TABLE B: SUMMARY OF POINT LOAD STRENGTH INDEX TEST RESULTS**

**BOREHOLE LOGS 201 AND 202 WITH ROCK CORE PHOTOGRAPHS**

**FIGURE 1: INVESTIGATION LOCATION PLAN**

**FIGURE 2: GRAPHICAL BOREHOLE SUMMARY**

**VIBRATION EMISSION DESIGN GOALS SHEET**

**REPORT EXPLANATION NOTES**

**APPENDIX: BOREHOLE LOGS 9 TO 12, 102, 103, 9188JV/4 &5, WITH ROCK  
CORE PHOTOGRAPHS, TP104 SECTION, OLD EXPLANATORY NOTES**



## **1 INTRODUCTION**

This report presents the results of a geotechnical investigation for the proposed library building between King Street and Carillon Avenue, Newtown, NSW. The investigation was commissioned by Mr Mark Louw of Allen Jack + Cottier (Architects) on behalf of Moore Theological College, on the basis of our proposal, Ref: P30491VTProp.

Jeffery and Katauskas Pty Ltd have previously undertaken geotechnical investigations at the sites of existing buildings at 1 King Street (Report Ref. 21871VTTrpt issued 10 March 2008, at 2-16 Carillon Avenue (Report Ref. 21871VTLet of 27 August 2008) and at 13A-19 King Street (Report Ref. 9188JV/ms, dated 30 November 1992). A copy of the factual information from each of these reports and a sheet defining the rock descriptions on the logs are contained in the Appendix to this report. These previous investigations revealed a subsurface profile generally consisting of shallow topsoil/fill and very stiff to hard residual clays of medium to high plasticity, which graded into weathered shale or siltstone at depths ranging from 0.2m to 3.6m below existing levels at the time of investigation. The rock was predominantly of low to medium strength but contained extremely low to very low strength bands or layers in places.

We understand that the proposed library will require demolition or partial demolition of several buildings and site improvements to allow construction of new multi-level building complex, possibly with up to seven above ground and two below ground levels. An automatic storage retrieval system, approximately 20m by 10m in plan dimension is also to be constructed within a section of the building and will extend a further 8m below the lower basement level. Other details of the development were not available at the time of report preparation. Structural loads have not been determined or supplied but we expect that some moderate to high loads would apply for the development.



The scope of this investigation was limited to obtaining geotechnical information on subsurface conditions at a further two locations as a basis for comments and recommendations on excavation and support, retaining walls, footings, floor slabs, and pavements. The recommendations may require review and possible amplification once more exact development details, such as final floor levels, structural loads etc. are finalised.

A summary of the principal geotechnical issues for the proposed development are presented in Section 4.1.

## **2 INVESTIGATION PROCEDURE**

The boreholes (BH201 and BH202) were auger drilled using a crawler mounted JK300 drilling rig, to depths of 5.85m and 6.4m below existing surface levels. The boreholes were then extended by rotary diamond coring techniques, using an NMLC triple tube core barrel with water flush, to termination at a common depth of 16.1m. Prior to drilling, each test location was checked by a specialist sub-contractor for buried services using electronic detection equipment, after referring to Dial Before You Dig services drawings. The borehole locations, which are shown on the attached Figure 1 were set out by taped measurements from surface features. The location of the boreholes was partly dictated by access constraints imposed by existing site developments.

The apparent compaction of the fill and strength of the subsurface soils was assessed from Standard Penetration Test (SPT) 'N' values supplemented by hand penetrometer readings on recovered split tube clayey samples. The strength of the weathered rock was assessed from observations of the auger penetration resistance using a tungsten carbide (TC) bit, together with examination of the recovered rock cuttings and subsequent laboratory moisture content tests. The strength of the



cored bedrock was assessed by examination of the recovered rock core and subsequent correlation with the results of rock strength testing.

Monitoring for groundwater was carried out in the boreholes during and on completion of individual boreholes. No longer term monitoring of groundwater levels has been carried out.

Our engineering geologist, Mr Janak Patel, set out the borehole locations, nominated the sampling and testing, and prepared the borehole logs. The borehole logs are included with this report, together with a Standard Set of Notes, which describes the methods and procedures employed in the investigation and their limitations and the logging terms and symbols used. The surface levels, as shown on the borehole logs, were interpolated from the spot levels shown on the supplied survey plan (Project Surveyors Drawings 15995-18565-3), and as such, should be considered as approximate. The datum is Australian Height Datum (AHD).

Selected disturbed samples were recovered from the site and returned to Soil Test Services (STS), a NATA registered laboratory, for moisture content tests. The test results are summarised in the attached Table A. The rock core was also returned to STS, where it was photographed and selected sections of core subjected to Point Load Strength Index Tests ( $I_{s(50)}$ ). The core photographs are attached opposite the relevant borehole log and the Point Load Strength Index tests are indicated on the borehole logs and are summarised in Table B.

### **3 RESULTS OF INVESTIGATION**

#### **3.1 Site Description**

The Moore Theological College complex is located within a relatively large site, bounded to the north by Carillon Avenue and by King Street to the east and south.



The college property extends to the west of Little Queen Street. Little Queen Street and Campbell Street sub-divide the college site into three separate areas.

The surrounding topography consists of undulating hilly to relatively flat terrain. The college site and environs has been substantially modified by previous earthworks to form cut and fill platforms for the buildings and roads.

The proposed library site is located at the crest of a hill, which generally slopes down to the north-east, south-east and to the north-west. The detail survey indicates that ground surface levels fall across the site from around RL 42.7m at its south-west frontage on King Street, to about RL 41.2m near its north-west frontage on Carillon Avenue. The King Street frontage falls to the east to about RL 42m at its intersection with Carillon Avenue. Carillon Avenue falls away to the west.

At the time of investigation, the site contained several two-storey, brick and rendered terraces, and two or three storey brick college buildings. Landscaped and grassed areas are locally supported, in places, by masonry block, timber, brick and rendered retaining walls between 0.5m and 1m in height. The paths and driveways are surfaced mainly with concrete. A variety of trees are scattered throughout the site area and along the Carillon Avenue frontage.

The existing two and three storey building at 2-16 Carillon Avenue is located on the side of a hill and on a site that falls away about 3.5m in level from its eastern end down to its north-west corner. The floor level of the eastern, two storey section of the building is at RL 43.0m and is close to grade on its east side. There is a covered concrete patio and relatively flat area along its north side; this area is retained by a brick wall, which runs along the street frontage, then turns to the south running back to the patio. The wall is bounded by concrete entry steps which step up about 1.5m to the patio floor. Further to the west, the building steps down and appears to be set about 2m into the hillside slope. This section of the building is of two and three



storeys. An approximately 2m high batter slope runs north to south in the foundation space below the ground floor around 20m from the west end of the building. The lower ground floor level is at RL 39.4m.

The remaining external area to the west of the entry steps contains a garden bed with large shrubs and is retained on its west side by an approximately 1m high mortared sandstone wall. The area at the toe of the wall is relatively flat to gently sloping and is mostly grassed, with concrete paths and garden beds. The areas adjacent to the east and west sides of the building are covered with concrete paving. There is a narrow garden bed adjacent to the eastern portion of the southern side of the building. Further south, a concrete road runs past the garden and along the side of the western portion of the building, toward the west.

The neighbouring site to the south-west, No 13-15 King Street, is located on the eastern property boundary, and contains a two-storey rendered building with a frontage on King Street. Its rear yard is relatively flat and grassed with a few trees.

### **3.2 Subsurface Conditions**

Reference should be made to the borehole logs for specific details of the significantly variable subsurface conditions encountered at each test location. Graphical summaries of the borehole information are presented in Figure 2.

In general terms, the current boreholes (BHs 201 and 202) encountered existing pavements, topsoil/fill, shallow fill over residual silty clays, which grade into weathered bedrock at 1.85m in BH201 and 1.0m in BH202 below existing levels at the time of drilling. For a more detailed description of the subsurface profile and groundwater levels encountered at each borehole location, reference should be made to the attached borehole logs.



The more pertinent details of the encountered variable subsurface conditions at BHs 201 and 202 are presented below:

- **Existing Pavements** consisting of concrete, 300mm or 150mm thick covered the ground surface.
- **Fill** was revealed below the pavements, extending to depths between 0.3m and 0.5m below existing levels. The fill comprised sandy gravel. Based on our observations and the SPT 'N' values, the fill was generally assessed to be poorly compacted. The fill extended down to the residual silty clay.
- **Residual Silty Clays** were of medium to high plasticity and contained a trace of ironstone gravel. The silty clays were predominantly of stiff to very stiff strength, with moisture contents generally greater than the plastic limit at the base of the existing fill. The silty clays graded into shale.
- **Weathered Shale Bedrock** was found below the silty clay profile. The shale was predominantly distinctly to slightly weathered, initially of very low to low or low strength. The shale improved in strength with depth. Medium strength or stronger shale was intersected at 5.7m in BH201 and at 5.5m in BH202. The rock was cored from 5.85m in BH201 and 6.40m in BH202. Defects within the cored rock included some extremely weathered seams or clay seams (between 10mm and 70mm thick), or bedding planes, fractured, crushed or extremely low strength bands, and several (50° to 90°) joints. The core loss zones are inferred to be extremely weathered seams or fractured bands.
- **Groundwater seepage** was not encountered during and on completion of auger drilling: Groundwater levels were measured at 3.3m in BH201 and at 3.4m in BH202 on completion of coring the rock with water flush. It should be noted



that the coring procedure involves the introduction of water into the ground; therefore groundwater levels after completion of drilling are expected to be artificially high and are unlikely to be the 'true' groundwater level.

### **3.3 Laboratory Test Results**

The moisture content tests on samples of the rock correlated well with our field assessment of rock strength. The approximate Unconfined Compressive Strengths (UCS) of the rock core, as shown on Table B, varied significantly from 6MPa to 26MPa for the shale, with an average of about 11MPa.

## **4 COMMENTS AND RECOMMENDATIONS**

### **4.1 Summary of Principal Geotechnical Issues and Further Work**

Based on the results of the subsurface investigations carried out, the principal geotechnical issues for the development are summarized to be as follows:

- The proposed development will involve substantial changes to the site including demolition of the existing buildings, retaining walls and other structures, pavements, and excavations of substantial volumes of soil and rock. Good engineering design, construction and maintenance practices should be adopted to maintain stability to adjoining buildings and structures during excavation and in the long term, as well as reducing the risk of vibration damage to adjoining buildings and structures during excavation.
- The groundwater levels were observed generally within the shale at 5.5m depth in the previously drilled BH9 on completion of auger drilling. Groundwater levels were measured at 3.3m in BH201 and at 3.4m in BH202; these are unlikely to be the 'true' groundwater level as water was introduced during coring. Nevertheless groundwater inflows into the excavations for the two



basement levels and for the below ground storage should be anticipated through defects in the shale.

- The proposed buildings of moderate to high loads should be founded on the underlying shale bedrock. Where bedrock is exposed or at shallow depth after site earthworks, pad or strip footings may be used, but piles will be required where the depth to rock is deeper than about 1.5m.
- The previous and the current boreholes indicate that existing fill is generally poorly compacted. We are unaware of records that document the manner of placement, compaction specification, and control of the fill. In the absence of compaction control records, this fill should not be relied upon to provide foundation support to on-ground slabs for the basement entry ramps unless it is fully re-compacted (or replaced) to an engineering specification in a controlled manner (refer to Sections 4.8.1 and 4.8.2).
- The proposed entry ramp on-ground slabs and pavements may be constructed on the treated fill subgrade, provided it is prepared and proof rolled as detailed in Section 4.8.1. However, even following proof rolling, and treatment as required, of the fill there will still be a risk of poor pavement performance due to the underlying uncontrolled fill. The only way to reduce such risks would be to excavate and replace the uncontrolled fill below the pavement area unless available records or additional testing confirms that the fill is sufficiently well-compacted.

Further comments on the above and other issues are provided within the following sections of this report. A summary of additional geotechnical work recommended are provided in Section 5. We believe sufficient information has been gained to be reasonably confident as to subsurface conditions. However, it will be essential during excavation and construction works that regular geotechnical inspections be commissioned to check initial assumptions about excavation and foundation conditions and possible variations that may occur between inspected and tested locations and to provide further relevant geotechnical advice. Irregular or 'milestone'



inspections by a geotechnical engineer are often not adequate for excavation, shoring and foundation works. It is recommended that the Client be made aware of the need to commission a geotechnical engineer for regular frequent inspections. The comments provided in this report should be reviewed following these inspections. A meeting of the design team may be of benefit in order to discuss the geotechnical issues and solutions in more detail.

#### **4.2 Dilapidation Surveys and Adjacent Buildings**

Prior to commencement of excavation, we recommend that dilapidation survey reports be carried out on the neighbouring buildings and structures to the west, within 15m of the proposed excavations. The reports would provide a record of existing conditions prior to commencement of the work. A copy of each report should be provided to the adjoining property owner who should be asked to confirm that it represents a fair assessment of existing conditions. The reports should be carefully reviewed prior to excavation commencing. We can complete these dilapidation reports if you wish to commission us.

Excavations and retention systems will need to be carefully planned and scheduled so as not to have any adverse effects on the buildings and structures adjoining or above the excavation. During the excavation, every care should be taken to not undermine or render unstable the footings of any adjoining buildings and boundary structures and to maintain stability in the long term, as well as reducing the risk of vibration damage to adjoining buildings and structures during excavation.

#### **4.3 Excavation**

Following demolition of existing buildings, site improvements and pavements and stripping of vegetation, we envisage that for the two level sections of the basement, the excavations would after demolition and stripping, penetrate the existing fill,



residual clays, interbedded clays and shale, and shale bedrock generally of extremely low to low strength. The deeper storage level and locally deeper excavations for footing, service trenches and the lift wells would probably encounter low to medium strength or stronger shale.

Should any large trees require removal, allowance should be made for readjustment of the moisture content of the potentially reactive silty clay foundation materials which probably extend below neighbouring buildings. For example, if the trees are removed during a relatively 'dry' period, the footings supporting the buildings would be subject to potential uplift pressures associated with possible subsequent swell of the clay fill or silty clay subgrade as it 'wets up'.

The soils can be readily excavated by a small to medium size excavator, a front end loader or dozer. Excavation in extremely low to low strength shale can normally be achieved using either a Caterpillar D7 dozer or equivalent, with some light to medium ripping, or by a ripping hook fitted to medium to large excavators. Much of this material can probably also be excavated using a large bucket excavator. However, localised stronger iron indurated or ironstone bands/zones were encountered in the poorer quality shale, which will require the use of heavier specialised equipment (eg rock hammers or larger dozers or heavy ripping). Excavation through the medium strength or stronger shale will be more difficult, requiring large rock saws in combination with heavy ripping using at least a Caterpillar D10 or similar dozers. A generous allowance should be made for rock hammer assistance to the ripping. Hydraulic rock breaking equipment would also be required for detailed excavations such as footings or services.

The excavatability of the rock and the selection of appropriate excavation equipment have been assessed on the basis of the rock core strength and limited information on the nature and inclination of rock defects. Assessment of excavation characteristics and productivity is not an exact science and contractors must make their own



evaluation based on experience with specific equipment, preferably after inspection of the rock cores (we only store these for one month after the formal report is issued unless other arrangements are made). The ease with which excavation of rock is achieved depends upon the equipment used, the skill and experience of the operator and the characteristics of the rock. The contractor must make his own judgement on all of these factors.

#### **4.3.1 Potential Vibration Risks**

If hydraulic impact hammers are used, considerable caution should be taken during rock excavation, as there will likely be direct transmission of ground vibrations to adjoining structures and buildings. Depending on the locations of buildings and other structures in relation to the excavations, continuous electronic vibration monitoring (i.e. measurement of peak particle velocities) may be required during the period of excavation.

If large rock hammers are to be used, we recommend that the initial excavation in rock should preferably be commenced away from likely critical areas and instrument vibration monitoring undertaken to confirm whether the vibration limits are likely to be exceeded and to provide guidance on how far the rock hammer should be kept away from the site boundaries. Guideline levels of vibration velocity for evaluating the effects of vibration in structures are given in the attached Vibration Emission Design Goals sheet. This limit of vibrations should be reviewed once more definite details of the excavation and development staging are known to confirm that they are still suitable. By monitoring vibrations in this way, it will allow some freedom to the excavation contractor in the equipment he adopts, so that a balance can be made between productivity and vibration reduction.

Vibrations induced by excavations can be reduced by alternative methods such as the following.



- Start the rock excavation away from likely critical areas.
- Maintain rock hammer orientation into the face and enlarge excavation by breaking small wedges off faces.
- Operate hammers in short bursts only, to prevent amplification of vibrations.
- Use smaller equipment (offset by a loss in productivity and economy and greater duration of the nuisance).
- Excavate a cut off trench around the site to reduce vibrations from excavation activities; this can be done progressively with the rock saw.
- Use line drilling, especially along excavation boundaries, to aid breaking and trimming.

As a very general guide, we have found on other sites that grinders or rock saws are typically required within about 5m to 10m of the buildings and structures. However the distance is very dependent on specific rock characteristics at each site, the equipment used and the condition of adjoining structures and, therefore, vibration monitoring is essential.

In addition, we recommend that only excavation contractors with appropriate insurances and experience on similar projects be used. The contractor should also be provided with a copy of this report to make his own judgement on the most appropriate excavation equipment.

#### **4.3.2 Excavation Batters**

The excavations in the shallow clay fill, silty clay of at least very stiff strength and extremely low to low strength shale may be battered at 1V in 1H. Surcharge loadings (footings, vehicles, etc) should not be within the zone of influence of the excavation. As a guide, surcharge loadings should be no closer than 2H from the top of any batter or the face of any excavation (including footing excavations),



where H is the vertical height of the batter or depth of the excavation in the fill, silty clay and low strength or weaker shale.

Flatter batters may be required where groundwater seepage is encountered. Where possible, water should be drained away from batter slopes and prevented from discharging over batter faces.

Permanent batters would need to be flatter (that is, no steeper than 1V in 2H) and protected from erosion by vegetation or other means.

Good quality shale of at least medium or higher strength (at 6.6m in BH11, at 4.8m in BH12, at 5.7m in BH201 and 5.5m in BH202) may possibly be cut to a temporary batter of about 1V in 0.5H or slightly steeper, depending on the orientation of the steeply inclined (mainly 60° to 70°) joints and individual cut faces. However, some allowance should be made for the potential larger scale instability (eg. continuous joints, etc) that occasionally exists within shale bedrock. These continuous joints can be as flat as 40° to 50° and run in north-west/south-east or north-east/south-west directions. Should these joints exist, flatter batters (possibly of the order of 1V in 1H or flatter) or large capacity rock anchors can be required; the cost of the latter would be relatively high and delays to the excavation process with consequential cost implications would occur.

The stability of all cuts and even those in good quality, medium strength or stronger shale bedrock must be subject to confirmation by an inspection by a geotechnical engineer. No excavation face should be allowed to advance more than 1.5m vertically between inspections and the excavation should be staged or stepped so that a whole face is not excavated 1.5m vertically between visits. If adverse defects are identified by the geotechnical engineer during the inspections, then stabilisation or flatter batters will be required. If there are only occasional bedding and joint defects in the rock, the face may only require protection by dowels, mesh



and shotcrete or the permanent basement walls. The extent of shotcrete to temporarily protect the rock faces prior to construction of the permanent walls should be confirmed during the geotechnical inspections. Stabilisation may also require the use of rock bolts, mesh and/or shotcrete protection to support the large blocks or other rock face areas or where near vertical cuts are proposed. The permission of adjoining property owners, to install rock bolts or dowels below their property, should be obtained in advance of construction in order that there is no delay in providing support should adverse conditions be encountered. It would be unusual to complete such an excavation without some form of support being required to the rock faces, though this may take forms other than rock bolting.

Where space permits, the permanent walls would then be constructed at the toe of the temporary batters in the soils, poorer quality shale and the vertical shale cuts and subsequent backfilling undertaken. Caution will be required during backfilling to prevent over-compaction adjacent to the walls and thereby causing excessive forces on the walls.

Where the basement excavations are extend right-up to existing buildings and the site boundaries, insufficient space would be available to excavate temporary batters. Where the batter slopes cannot be accommodated or are not preferred, then the vertical excavation in fill and any poor quality shale will need to be supported by appropriate shoring systems or properly engineered retaining walls, with due allowance for the slope of the ground behind the walls. Any necessary vertical support system will need to be installed prior to excavation.

#### **4.4 Groundwater and Drainage**

We expect that localised seepage may possibly occur into the excavations along the soil/bedrock boundary and along existing defects, such as bedding planes and joints, which exist in the rock. Localised seepage may also occur through the fill or



permeable gravelly layers in the clay, especially during and following periods of heavy rainfall. We anticipate that seepage would be controllable using conventional sump and pump techniques.

Complete and permanent drainage and appropriate waterproofing are recommended for the walls and floors close to or in contact with the excavated areas.

Where basement excavations are proposed, some under-floor drainage will be required for on-ground slabs constructed over the shale, though this should be reviewed following after inspection of the completed excavation. The drains should incorporate a sump and gravity or an automatic pump-out system for discharge of collected seepage to the stormwater system.

The silty clay subgrade is likely to soften with an increase in moisture content. Therefore, good and effective site drainage should be provided both during construction and for long term site maintenance. Earthworks platforms should be graded to maintain cross-falls during construction. The principal aim of the drainage is to promote run-off and reduce ponding. A poorly drained clay subgrade will also become untrafficable when wet. We recommend that if soil 'softening' occurs, the subgrade be over-excavated to below the depth of moisture 'softening' and that the excavated material be replaced with engineered fill, compacted as specified in Section 4.8.2.

#### **4.5 Shoring Systems and Retaining Walls**

A suitable method of retention to support vertical cuts, prior to bulk excavation, would be bored cast in-situ or augered, grout injected (CFA), soldier pile walls with infill panels where movement is not of concern, or alternatively, contiguous pile walls. Construction of the contiguous pile walls should be of high quality, taking the uttermost care to prevent soil loss through gaps that may occur between the piles as



this would add to the possibility of settlement occurring outside the excavation. Such gaps should be rectified without delay, such as by mass concrete infill.

Conventional driven sheet-pile walls would not be suitable as there is a need to minimise noise and avoid ground vibration damage to the neighbouring buildings; sheet piles do not penetrate weathered rock effectively.

We advise that cantilevered walls may be used for supporting retained heights of around 3m to 4m and only where some higher lateral and vertical movements of adjoining ground can be tolerated. If greater height walls are required, or, where only minimal movements can be tolerated, then anchored or propped walls would normally be required.

The piles of the shoring walls should be suitably embedded below the base of the excavation. Props or anchors will also be needed to restrain the upper sections of the walls and these must be installed progressively and immediately once the propping point has been uncovered, and prior to excavation adjacent to neighbouring structures and sensitive services which are located within the 2H zone of influence of the excavation perimeter (discussed in Section 4.3.2).

Drilling of rock sockets will be difficult through the iron indurated bands and medium to high strength rock requiring the use of heavy drilling rigs equipped with rock augers and a coring bucket. Some groundwater inflow is expected into bored pile footings and we expect that this inflow will be controllable by conventional pumping methods. Alternatively, concrete may be poured using tremie methods.

#### **4.5.1 Retaining Wall Design Parameters**

Design of the retaining walls may be on the basis of an 'active' lateral pressure coefficient,  $K_a$ , of at least 0.35 for the fill, clayey soils, extremely low, and extremely



low to very low strength shale, provided some deflection is tolerable. The K value may be reduced to about 0.2 for shale of at least low strength rock. Subject to geotechnical inspection, no K value need to be taken into account for the shale of at least medium strength. Approximate bulk unit weights of  $20\text{kN/m}^3$  for the soils and  $22\text{-}23\text{kN/m}^3$  for extremely low to low strength rock may be adopted. Walls which are to be subsequently propped by the permanent structure (e.g. by the upper ground floor slab) should be designed based on a higher lateral pressure coefficient, K, of at least 0.6 (or about 0.4 for low strength shale). These coefficients assume almost horizontal ground surfaces behind the crest of the walls.

For propped or anchored walls, we recommend the use of a trapezoidal lateral earth pressure of at least  $4H$  (kPa), where H is the retained height in metres in the soils and shale. For propped or anchored walls in areas, which are highly sensitive to lateral movement (such as adjacent to neighbouring building footings located within  $2H$  metres of the excavation), a greater trapezoidal lateral earth pressure of at least  $8H$  (kPa) should be used. These  $4H$  and  $8H$  pressures should be assumed to be uniform over the central 50% of the full, retained height in the soils and shale of low strength or weaker. Alternatively, more sophisticated computer based shoring design (such as Wallap) generally results in cost savings compared to designs based on simplified assumptions regarding earth pressure distributions. These detailed numerical analyses can model the progressively anchored or propped shoring walls as they are constructed. The lateral earth pressure coefficients ( $K_a$  and K) nominated for the cantilever wall may be adopted to confirm the minimum depth of embedment of the wall toe and the likely order of magnitude of wall movements during the various phases of construction when using Wallap.

The recommended lateral earth pressure coefficients and trapezoidal pressures assume almost horizontal ground surfaces behind the crest of the walls. If inclined backfill surfaces are to be designed, then the above factors would have to be



increased or the inclined section of backfill should be taken as a surcharge load in the design.

Applicable hydrostatic pressures should be added to the lateral earth pressures, unless specific measures are taken to introduce complete and permanent drainage of the ground behind the walls. Any surcharge affecting the walls (e.g. footings, retaining walls and their backfill, the ground slope behind the wall, etc.) should also be taken into account in design.

Anchors may be designed for an allowable bond stress of 350kPa for shale bedrock of at least low strength. All ground anchors should be proof tested to 1.3 times the working load under the supervision of an experienced engineer independent of the anchor contractor. Anchors must be bonded behind a 45° line drawn upwards from the base of the excavation. Anchor group interaction must also be taken into account. Permanent anchors should have appropriate corrosion provisions.

#### **4.5.2 Excavation Induced Movements**

It is inevitable that the excavation will induce movements of the adjacent ground that falls within the area of influence of the excavation.

Lateral and horizontal movements could occur within about 2H back from the anchored wall. With a less rigid support system, excavation induced movements should be expected to be of a higher order. Settlements may also be caused by the wall construction itself (e.g. loss of ground during anchor drilling, etc).

As excavation of the rock progresses, the rock mass will also tend to move inwards towards the excavation along bedding planes, clay seams, etc. as it is stress relieved. With increasing depth of excavation, the bed undergoing excavation will also drag overlying beds with it as the lower bed moves towards the excavation. The extent of



movement will depend on the strength of the rock between the bedding planes and the spacing of joints or other defects. As the beds move inwards, joints, etc. will start opening behind the excavated face and any structures on or in the rock also move. These stress-relief movements will decrease away from the excavated face, however, their magnitude will increase as the depth of excavation increases.

Experience with excavations in residual clay and weathered shale indicates that lateral and vertical ground movements of around 2 to 5mm/m of excavation depth may occur, mostly as a result of stress relief, depending on the rigidity and construction practice of the shoring system.

It may not be practicable to prevent significant vertical and lateral ground displacements immediately beyond the limits of the excavation, so the effects of the inevitable excavation induced movements on the adjoining buildings and structures and also on the permanent structure should be assessed.

The objective with properly engineered retaining walls is to keep the adjacent ground movements within tolerable limits. The actual wall movements are highly dependent on the construction sequence, detailing and quality of installation and should be assessed by the structural engineer for the system to be adopted. Hence, any existing adjoining structures, or buried services, which fall within the area of influence of the excavations, should be assessed for risks of damage due to excavation-induced movements and whether underpinning is required. The underpinning should be designed for lateral earth pressures, any surcharge loadings and hydrostatic pressures.

The risk of architectural or structural damage to adjoining buildings and structures will depend on their sensitivity to horizontal and vertical deformations, structural load, type and founding elevations of the floor slabs and footings and foundation



conditions. All these factors should be carefully investigated and evaluated prior to excavation commencing.

In addition, we recommend that an excavation/retention methodology be prepared prior to bulk excavation commencing. The methodology must include but not be limited to proposed excavation, retention and underpinning techniques, the proposed excavation equipment, excavation/retention/underpinning sequencing, geotechnical inspection intervals or hold points, vibration monitoring procedures, monitor locations, monitor types, contingency plans in case of non-compliance. Preferably, this methodology should be shown on the structural engineer's drawings. The excavation/retention/underpinning methodology should be reviewed and approved by the geotechnical engineer.

#### **4.6 Footing Design**

Building and basement retaining wall footings may be supported by strip or pad footings, or bored, cast in-situ piles or augered, grout injected piles founded in the underlying siltstone and shale bedrock to limit the potential for differential settlements.

Strip and pad footings or bored piles or augered, grout injected (CFA) piles may be designed for maximum allowable working bearing pressures for the siltstone and shale given in Table 1. Rock sockets in piled footings below the indicative founding levels specified above may be designed for a safe adhesion value of 10% of the appropriate safe bearing pressure under compressive vertical loading. Two-thirds of these adhesion values may be adopted in uplift. These adhesion values assume excavation is not carried out within the zone of influence of the footing. The bearing and adhesion values assume footing bases have been cleaned of loosened or softened materials and sockets are free of smeared material (a special roughening tool is normally required to achieve this in bored piers).



For footings fully embedded into the underlying bedrock below the lowest building floor level, an allowable lateral stress in the rock socket equal to one third of the allowable bearing pressure may be adopted. These passive resistance values assume excavation is not carried within the zone of influence of the wall toe and the rock does not contain unfavourable defects etc. The upper 0.3m depth of the socket should not be taken into account to allow for disturbance effects during excavation.

**Table 1 – Footing Bearing Pressures and Depth**

<b>Borehole Number</b>	<b>Depth (in metres) below existing ground level for Safe Bearing Pressure of 700kPa</b>	<b>Depth (in metres) below existing ground level for Safe Bearing Pressure of 1500kPa</b>	<b>Depth (in metres) below existing ground level for Safe Bearing Pressure of 3500kPa</b>
9	2.8	2.8	(2.8)
10	3.9	5.9	6.4
11	1.8	6.9	(6.9)
12	1.7	5.1	5.1
102	0.7	4.1	-
103	2.0	4.3	-
201	2.1	3.0	5.8
202	1.5	1.5	5.8
9188JV/5	2.2	3.7	(4.3)

**Note** - The bracketed founding depths for the higher bearing value of 3500kPa are likely to be appropriate for the low to medium strength or stronger rock; however, additional proving would be required in diamond cored boreholes, with rock strength testing of the recovered cores.

Where footings are founded close to the top of a rock face, the allowable bearing pressure below these footings will need to be carefully assessed. The safe bearing pressure would need to take into account rock strength, the inclination of the rock face, jointing and the influence of clay seams as well as the magnitude and inclination of the applied loadings.



If the designer wishes to adopt the limit state design methods, such as in the Piling Code, AS2159-1995, then the ultimate values of end bearing pressure may be estimated by multiplying the above recommended allowable bearing and lateral stress values by Factors of Safety of 3. A Factor of Safety of 2 should be applied to the shaft adhesion values. We recommend that the ultimate values be multiplied by a geotechnical strength reduction factor,  $\Phi_g$ , of 0.5. Higher reduction factors may be adopted but these will depend on the intensity and type of proving of the footings and their foundation. An appropriate load factor should also be applied to the proposed footing loadings.

The rock bearing pressures given in Table 1 are based on a serviceability criteria of deflections at the footing base/pile toe of less than or equal to 1% of the least footing dimension (or pile diameter). Footing settlements may be estimated using the Elastic Moduli given in Table 2.

**Table 2 – Elastic Moduli for Footings in Rock**

<b>Strata</b>	<b>Bulk Unit Weight (kN/m<sup>3</sup>)</b>	<b>Poisson's Ratio</b>	<b>Elastic Modulus (MPa)</b>
Shale – extremely low to very low strength with iron indurated bands	22	0.25	100 – 150
Shale – low strength	23	0.25	400 – 500
Shale – low to medium strength	23	0.2	500 – 700
Shale – medium or medium to high strength	23	0.2	1000 – 2000

Footings on rock can also be designed using 'Limit State Design' principles as detailed in the paper "Foundation on Sandstone and Shale in the Sydney Region' by Pells, Mostyn and Walker, Australian Geomechanics, Number 33, Part 3, December



1998 (Pages 17-29). It must be emphasised that the use of limit state design to adopt relatively high bearing pressures (above the serviceability criteria described above) is not currently standard practice, and there is an increased risk of inadequate footing performance.

If construction proceeds during a relatively 'dry' period, the beams between piles should be designed to withstand potential uplift pressures associated with possible subsequent swell of the clay fill or silty clay subgrade as it 'wets up'. Alternatively, the beams should be underlain with void formers or similar (at least 70mm thick) to minimise the impact of uplift pressures. A degree of uplift protection can be achieved by tyning/loosing the soil below the ground beams for say 120mm depth.

#### **4.6.1 Footing Construction**

In order to minimise potential problems, we recommend that a pre-construction meeting be held so that all parties involved understand the proposed footing design and construction requirements and how to identify the weathered rock materials at the indicative founding levels so as to minimise over-drilling of the piles during construction.

If bored or augered grout piles are to be socketted into the shale/siltstone then we recommend that heavy drilling rigs with rock augers be used to drill the piles. Heavy drill rigs with coring buckets may be required for drilling through medium strength or stronger rock or through the iron indurated bands.

Some groundwater seepage can be expected during the construction of piers and we recommend that trials should be undertaken to confirm piers can be successfully constructed at the site, otherwise augered, grout injected piles should be used. Piers should be dewatered (by conventional pumping methods) prior to concreting or the concrete may be poured using tremie methods.



All footings should be drilled, cleaned, inspected and poured with minimal delay, on the same day or the base of the footing should be protected by a concrete blinding layer after cleaning of loose spoil and inspection. Water should be prevented from ponding in the base of footings as this will tend to soften the foundation material, resulting in further excavation and cleaning being required.

In addition to inspection, the shale foundation may also need to be spoon tested or cored in boreholes if footings are designed using a safe bearing pressure of 3.5MPa. This testing is to confirm that seams or defects present below the founding levels are within tolerable limits. The presence of such seams would require a reduction in allowable bearing capacity or an increase in footing depth. The amount of testing should be addressed when structural design is more advanced.

The initial stages of footing excavation/drilling, particularly if bored piles are adopted, should be inspected by a geotechnical engineer/engineering geologist to ascertain that the recommended foundation material has been reached and to check initial assumptions about foundation conditions and possible variations that may occur between borehole locations. The need for further inspections can be assessed following the initial visit.

#### **4.7 Basement Floor Slab**

On-ground floor slabs may be constructed over the shale and no special treatment is required other than the removal of loose and softened material. Areas, which have to be built-up to infill low points in the excavations should be filled with properly compacted sub-base material.

Although we expect that some under-floor drainage will be required, this should be reviewed following further monitoring of groundwater seepage during and on



completion of the excavations. The under-floor drainage (such as perimeter drains and/or a free draining gravel bed) should be installed with sumps for gravity or automatic pumped discharge of groundwater. If under-floor drainage is not installed, then the on-ground floor slab may be subjected to uplift pressures from the groundwater; this may require additional mass or ground anchors.

The basement floor slab, where subject to traffic loadings, should have a sub-base layer of at least 100mm thickness of crushed rock to RTA QA specification 3051 (1994) unbound base material (or equivalent good quality durable fine crushed rock) which is compacted to at least 100%SMDD.

#### **4.8 On-Ground Slabs and Pavements**

The on-ground slabs and pavements for basement entry ramps may also be founded on engineered fill or the proof rolled and treated clayey subgrade on condition that the subgrade is prepared in accordance to the recommendations provided in Sections 4.8.1 and 4.8.2. On-ground slabs constructed over the fill or clays should be isolated from those slabs, walls and columns founded on the sandstone.

The design of pavements will depend on subgrade preparation, subgrade drainage, the nature and composition of new fill imported to the site, as well as vehicle loadings and use.

Lightly loaded pavements may tentatively be designed using a lower bound characteristic CBR value of 2.5% or a coefficient of subgrade reaction of 20kPa/mm (750mm plate) or a long term Young's modulus of 10MPa for the proof rolled and treated clay subgrade. These preliminary design values should be confirmed by CBR tests once initial earthworks design is complete and by inspection and testing during construction.



Concrete pavements and on-ground floor slabs subject to traffic loadings should be supported on a sub-base layer of RTA Specification 3051 unbound or equivalent good quality crushed rock, compacted to a density of at least 100% SMDD.

Concrete pavements should be provided with effective shear connection at joints by using dowels or keys. Concrete pavements should preferentially be used in areas where heavy vehicles manoeuvre such as garbage bin and truck unloading areas.

#### **4.8.1 Subgrade Treatment**

Where slab support is required, the existing fill should (subject to further geotechnical assessment of fill compaction control records) be excavated at least 2m beyond the perimeter of the slab, if possible, and re-compacted to form a properly compacted, engineered fill (refer to Sections 4.8.2).

Excavation and re-compaction of the fill would not be required where slabs are to be fully suspended and do not rely on the fill for support.

Any remaining existing fill may be left in place below proposed pavements on the condition that the subgrade is proof rolled and appropriately treated. However, there is a chance that some settlement may still occur under pavements bearing on the existing fill, even after it is treated by proof rolling.

Following excavation to the proposed design levels, the exposed soil subgrade should be proof rolled using a 5 tonne dead weight smooth drum vibratory roller under the supervision of an experienced earthworks superintendent, geotechnician or geotechnical engineer to check for any unstable areas. Proof rolling would not be required below slabs, which are to be fully suspended and do not rely on the underlying subgrade for support. During proof-rolling care should be taken to avoid



vibration damage to any neighbouring structures or services or improvements. The vibrations should be monitored and the vibrations may need to be reduced or ceased if there is a risk of damage. Where unstable areas are encountered the area should be locally excavated down to a sound base and replaced with engineered fill as detailed in Section 4.8.2.

If 'dry' conditions prevail at the time of construction, the clayey subgrade may become desiccated or have shrinkage cracks prior to sealing with sub-base or base materials. If this occurs then the subgrade must be watered and rolled until the cracks disappear.

We recommend that reference be made to AS2870 for drainage and vegetation precautions on reactive sites.

#### **4.8.2 Engineered Fill**

Engineered fill should preferably comprise well-graded granular material (ripped or crushed shale or sandstone), free of deleterious substances and having a maximum particle size of 75mm. The sandy fill materials may be re-used, however, the clay fill and clay materials are less desirable but may be re-used provided unsuitable ('over-wet' and 'over-size') material and any deleterious material is excluded. The well-graded granular fill for backfilling excavations or for raising site levels should be compacted in layers of not greater than 150mm loose thickness, to a density between 98% and 102% of Standard Maximum Dry Density (SMDD). Clayey fill should be compacted to a similar density but within  $\pm 2\%$  of Standard Optimum Moisture Content (SOMC). However, it would be wise to have a capping layer of better quality imported fill over the clay fill materials. The use of clay materials for engineered fill will entail more rigorous earthworks supervision and compaction control.



All fill should either be retained or battered to a slope of compacted fill of no steeper than 1V in 2H to prevent instability. The fill should also be 'keyed in' the existing side batters. All engineered fill areas should be over-filled and compacted and then the loose outer face of the fill should be cut back so that only well-compacted fill remains. We recommend a horizontal compacted fill platform extend beyond the slab/pavement periphery by at least 2m, where possible. All exposed fill should be protected from erosion by quickly establishing a grass cover.

Density testing should be carried out at not less than the frequencies given in AS3798. At least Level 2 testing (but Level 1 where fill is to support movement-sensitive floor slabs/pavements) of earthworks should be carried out in accordance with AS3798. Preferably, the geotechnical testing authority should be engaged directly on behalf of the client and not as part of the earthworks contract. We can complete these tests if you wish to commission us.

The earthworks recommendations provided here should be complemented by reference to AS3798.

## **5 SUMMARY OF FURTHER GEOTECHNICAL WORK**

Excavation and retention recommendations provided in this report should be complemented by reference to the Code of Practice Excavation Work, Cat. No. 312 by WorkCover NSW.

As detailed in this report, further geotechnical work is recommended as follows:

- Dilapidation survey for the neighbouring buildings/structures, especially if rock hammers are used.
- Assessment of the effects of excavation on the nearby building footings and whether underpinning is required.



- Quantitative monitoring of transmitted vibrations during rock excavation using rock hammers.
- Assessment of groundwater inflow to confirm drainage requirements following excavation.
- Inspection of the excavations to confirm batters and rock face treatment for cuts in the medium or higher strength rock.
- Inspection of footing excavations to ascertain that the recommended foundation has been reached and to check initial assumptions regarding foundation conditions and possible variations that may occur.
- Inspect proof rolling of fill/silty clay subgrade to detect soft spots requiring treatment.
- Carry out laboratory CBR testing of silty clay subgrade parameters for pavement design.

We recommend that Jeffery & Katauskas Pty Ltd view the proposed earthworks and structural drawings and section details in order to confirm they are within the guidelines of this report.

## **6 GENERAL COMMENTS**

The recommendations presented in this report include specific issues to be addressed during the construction phase of the project. As an example, special treatment of soft spots may be required as a result of their discovery during proof-rolling, etc. In the event that any of the construction phase recommendations presented in this report are not implemented, the general recommendations may become inapplicable and Jeffery and Katauskas Pty Ltd accept no responsibility whatsoever for the performance of the structure where recommendations are not implemented in full and properly tested, inspected and documented.



The long-term successful performance of floor slabs and pavements is dependent on the satisfactory completion of the earthworks. In order to achieve this, the quality assurance program should not be limited to routine compaction density testing only. Other critical factors associated with the earthworks may include subgrade preparation, selection of fill materials, control of moisture content and drainage, etc. The satisfactory control and assessment of these items may require judgement from an experienced engineer. Such judgement often cannot be made by a technician who may not have formal engineering qualifications and experience. In order to identify potential problems, we recommend that a pre-construction meeting be held so that all parties involved understand the earthworks requirements and potential difficulties. This meeting should clearly define the lines of communication and responsibility.

Occasionally, the subsurface conditions between the completed boreholes may be found to be different (or may be interpreted to be different) from those expected. Variation can also occur with groundwater conditions, especially after climatic changes. If such differences appear to exist, we recommend that you immediately contact this office.

This report provides advice on geotechnical aspects for the proposed civil and structural design. As part of the documentation stage of this project, Contract Documents and Specifications may be prepared based on our report. However, there may be design features we are not aware of or have not commented on for a variety of reasons. The designers should satisfy themselves that all the necessary advice has been obtained. If required, we could be commissioned to review the geotechnical aspects of contract documents to confirm the intent of our recommendations has been correctly implemented.

The offsite disposal of soil will most likely require classification in accordance with the Department of Environment & Climate Change (NSW) guidelines as Virgin Un-



Excavated Natural Material (VENM), General Solid, Restricted Solid or Hazardous waste. We can complete the necessary classification and testing if you wish to commission us. As testing requires about seven days to complete, allowance should be made for such testing in the construction program unless testing is completed prior to construction. If contamination is found to be present then substantial further testing and delays should be expected. We strongly recommend this issue be addressed prior to commencement of excavation on site.

If there is any change in the proposed development described in this report then all recommendations should be reviewed.

This report has been prepared for the particular project described and no responsibility is accepted for the use of any part of this report in any other context or for any other purpose. Copyright in this report is the property of Jeffery and Katauskas Pty Ltd. We have used a degree of care, skill and diligence normally exercised by consulting engineers in similar circumstances and locality. No other warranty expressed or implied is made or intended. Subject to payment of all fees due for the investigation, the client alone shall have a licence to use this report. The report shall not be reproduced except in full.

Should you have any queries regarding this report, please do not hesitate to contact the undersigned.

Tony Walker  
Associate

Fernando Vega  
Senior Associate  
For and on behalf of

JEFFERY AND KATAUSKAS PTY LTD.

Ref No: 21871VT2  
Table A: Page 1 of 1

**TABLE A**  
**SUMMARY OF MOISTURE CONTENT TEST RESULTS**

AS 1289	TEST METHOD	2.1.1
BOREHOLE NUMBER	DEPTH	MOISTURE CONTENT
	m	%
201	2.70-3.00	9.2
201	4.20-4.50	8.7
201	5.50-5.70	5.5
202	1.30-1.50	6.5
202	2.70-3.00	9.4
202	4.20-4.50	9.2
202	5.50-6.00	4.0

Ref No: 21871VT2  
 TABLE B Page 2 of 2

**TABLE B**  
**SUMMARY OF POINT LOAD STRENGTH INDEX TEST RESULTS**

BOREHOLE NUMBER	DEPTH	$I_{S(50)}$	ESTIMATED UNCONFINED COMPRESSIVE STRENGTH
	m	MPa	(MPa)
202	6.49-6.52	0.3	6
	7.00-7.03	0.6	12
	8.10-8.13	0.5	10
	8.75-8.78	0.8	16
	9.14-9.17	0.5	10
	9.84-6.87	0.5	10
	10.22-10.25	0.5	10
	10.92-10.95	0.4	8
	11.35-11.38	0.4	8
	12.39-12.42	0.5	10
	13.65-13.68	0.4	8
	14.39-14.42	0.4	8
	15.05-15.08	0.3	6
	15.90-15.93	0.4	8

**NOTES:**

1. In the above table testing was completed in the Axial direction.
2. The above strength tests were completed at the 'as received' moisture content.
3. Test Method: RTA T223.
4. The Estimated Unconfined Compressive Strength was calculated from the point load Strength Index by the following approximate relationship and rounded off to the nearest whole number :  

$$U.C.S. = 20 I_{S(50)}$$

Ref No: 21871VT2  
 TABLE B Page 1 of 2

**TABLE B**  
**SUMMARY OF POINT LOAD STRENGTH INDEX TEST RESULTS**

BOREHOLE NUMBER	DEPTH m	$I_{s(50)}$ MPa	ESTIMATED UNCONFINED COMPRESSIVE STRENGTH (MPa)
201	6.20-6.23	0.5	10
	7.17-7.20	0.7	14
	7.85-7.88	0.6	12
	8.16-8.19	0.4	8
	9.33-9.36	1.3	26
	10.22-10.25	0.9	18
	11.36-11.39	0.8	16
	12.10-12.13	0.4	8
	12.82-12.85	0.5	10
	13.52-13.55	0.4	8
	14.03-14.05	0.8	16
	15.66-15.69	0.5	10
	14.90-14.93	0.5	10

**NOTES: See Page 2 of 2**



JOB NO: 21871 VT2

BOREHOLE NO: 201

START CORING @ 5.85<sub>m</sub>

5

5.85<sub>m</sub>

6

7

8

9

10

11

12

13

14

15

16

END OF BOREHOLE @ 16.10<sub>m</sub>



Borehole No.

**201**

2/3

## CORED BOREHOLE LOG

<b>Client:</b>	MOORE THEOLOGICAL COLLEGE
<b>Project:</b>	PROPOSED LIBRARY BUILDING
<b>Location:</b>	1-11 KING STREET & 2-16 CARILLON AVENUE, NEWTOWN, NSW

<b>Job No.</b> 21871VT2	<b>Core Size:</b> NMLC	<b>R.L. Surface:</b> ≈ 42.8m
<b>Date:</b> 10-2-09	<b>Inclination:</b> VERTICAL	<b>Datum:</b> AHD
<b>Drill Type:</b> JK300	<b>Bearing:</b> -	<b>Logged/Checked by:</b> J.P.

Water Loss/Level	Barrel Lift	Depth (m)	Graphic Log	CORE DESCRIPTION Rock Type, grain characteristics, colour, structure, minor components.	Weathering	Strength	POINT LOAD STRENGTH INDEX I <sub>s</sub> (50)	DEFECT DETAILS			
								DEFECT SPACING (mm)	DESCRIPTION Type, inclination, thickness, planarity, roughness, coating.		
								EL VL L M H VH EL	500 300 100 50 30 10		
		5		START CORING AT 5.85m							
FULL RET- URN		6		SHALE: grey, with light grey laminae.	DW	M	X		- CS, 30mm.t - XWS, 35mm.t		
		7			SW		X		- J, 80°, P, S, IS - Be, 0-5°		
		8					X				
		9					X		- J, 60°, P, S - Cr, 70mm.t - J, 70-80°, P, S		
		10					X		- J, 60-70°, HEALED - J, 80-90°, HEALED - J, 70°, HEALED		
		11					X		- XWS, 10mm.t - J, 80°, P, S - J, 80°, P, S		
							X		- J, 70-80°, P, S - J, 70-80°, P, S - FRACTURED ZONE, 240mm.t		
							X		- XWS, 15mm.t - XWS, 30mm.t		
							XW	EL			
							SW	M			
									- Cr, 50mm.t		

JOB NO: 21871 VT2

BOREHOLE NO: 201

START CORING @ 5.85<sub>m</sub>

5

5.85<sub>m</sub>

6

7

8

9

10

11

12

13

14

15

16

END OF BOREHOLE @ 16.10<sub>m</sub>



Borehole No.  
**201**  
3/3

# CORED BOREHOLE LOG

**Client:** MOORE THEOLOGICAL COLLEGE  
**Project:** PROPOSED LIBRARY BUILDING  
**Location:** 1-11 KING STREET & 2-16 CARILLON AVENUE, NEWTOWN, NSW

**Job No.** 21871VT2      **Core Size:** NMLC      **R.L. Surface:** ≈ 42.8m  
**Date:** 10-2-09      **Inclination:** VERTICAL      **Datum:** AHD  
**Drill Type:** JK300      **Bearing:** -      **Logged/Checked by:** J.P./[Signature]

Water Loss/Level	Barrel Lift	Depth (m)	Graphic Log	CORE DESCRIPTION Rock Type, grain characteristics, colour, structure, minor components.	Weathering	Strength	POINT LOAD STRENGTH INDEX I <sub>s</sub> (50)	DEFECT DETAILS											
								DEFECT SPACING (mm)		DESCRIPTION Type, inclination, thickness, planarity, roughness, coating.									
								EL	VI	L	N	H	VH	EH	500	300	100	50	20
FULL RET URN		13		SHALE: grey, with light grey laminae.	SW	M	X											- XWS, 30mm.t	
		14		X															- Cr, 70mm.t - J, 80°, P, S
		15		X															- Cr, 200mm.t
		16		X															- J, 50-60°, P, S - J, 70°, HEALED - Be, 0-5°, P, S - J, 75°, P, S
				END OF BOREHOLE AT 16.10m			X											- J, 80°, HEALED	
		17																	
		18																	



Borehole No.  
**202**  
1/3

# BOREHOLE LOG

**Client:** MOORE THEOLOGICAL COLLEGE  
**Project:** PROPOSED LIBRARY BUILDING  
**Location:** 1-11 KING STREET & 2-16 CARILLON AVENUE, NEWTOWN, NSW

**Job No.** 21871VT2      **Method:** SPIRAL AUGER JK350      **R.L. Surface:** ≈ 42.4m  
**Date:** 11-2-09      **Datum:** AHD  
**Logged/Checked by:** J.P./

Groundwater Record	SAMPLES			Field Tests	Depth (m)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition/Weathering	Strength/Rel. Density	Hand Penetrometer Readings (kPa.)	Remarks
	ES	U50	DB									
DRY ON COMPLETION OF AUGER-ING					0		-	CONCRETE: 150mm.t	-	-	-	NO OBSERVED REINFORCEMENT
							CL-CH	FILL: Sandy gravel, fine to medium grained, brown and grey.	MC > PL	St-VSt	-	RESIDUAL
				N = 33 2,5,28			-	SILTY CLAY medium to high plasticity, light grey mottled orange brown and red brown, with a trace of ironstone gravel.			180 220 220	
					1		-	SHALE: grey and brown, with iron indurated bands and clay bands.	XW DW	EL L	-	VERY LOW TO LOW 'TC' BIT RESISTANCE
					2							
					3							
					4							
					5			as above, but dark grey.		L-M		LOW TO MODERATE 'TC' BIT RESISTANCE
					6					M		
					7			REFER TO CORED BOREHOLE LOG				

▼  
ON  
COMPLETION OF  
CORING

JOB NO: 21871VTZ

BOREHOLE: 202

START CORING @ 6.40m

6

6.40m

7

8

CORE LOSS 0.2m

9

10

11

12

13

14

15

CE  
005

16

END OF BORE-HOLE AT 16.10m



Borehole No.  
**202**  
2/3

# CORED BOREHOLE LOG

**Client:** MOORE THEOLOGICAL COLLEGE  
**Project:** PROPOSED LIBRARY BUILDING  
**Location:** 1-11 KING STREET & 2-16 CARILLON AVENUE, NEWTOWN, NSW

**Job No.** 21871VT2      **Core Size:** NMLC      **R.L. Surface:** ≈ 42.4m  
**Date:** 11-2-09      **Inclination:** VERTICAL      **Datum:** AHD  
**Drill Type:** JK300      **Bearing:** -      **Logged/Checked by:** J.P./

Water Loss/Level	Barrel Lift	Depth (m)	Graphic Log	CORE DESCRIPTION Rock Type, grain characteristics, colour, structure, minor components.	Weathering	Strength	POINT LOAD STRENGTH INDEX I <sub>s</sub> (50)	DEFECT DETAILS			
								DEFECT SPACING (mm)		DESCRIPTION Type, inclination, thickness, planarity, roughness, coating.	
										Specific	General
		6		START CORING AT 6.40m							
		7		SHALE: dark grey, with light grey laminae.	DW	M	X		<ul style="list-style-type: none"> <li>- Be, 0-5°, P, S</li> <li>- J, 80°, P, S</li> <li>- J, 70°, P, S</li> <li>- J, 90°, P, S, CLAY COATED</li> </ul>		
		8		CORE LOSS 0.20m					- XWS, 10mm.t		
FULL RET URN		9		SHALE: dark grey, with light grey laminae.	SW	M	X		<ul style="list-style-type: none"> <li>- Be, 0°, P, S</li> </ul>		
		10					X		<ul style="list-style-type: none"> <li>- Cr, 10mm.t</li> <li>- J, 50-60°, P, S</li> <li>- FRACTURED 40mm.t</li> </ul>		
		11					X		<ul style="list-style-type: none"> <li>- XWS, 10mm.t</li> <li>- J, 70°, P, S</li> <li>- J, 70°, P, S</li> <li>- J, 90°, P, S</li> <li>- CS, 20mm.t</li> </ul>		
		12					X				

JOB NO: 21871VTZ

BOREHOLE: 202

START CORING @ 6.40m

6

6.40m

7

8

CORE LOSS 0.2m

9

10

11

12

13

14

15

CE  
005

16

END OF BORE-HOLE AT 16.10m



Borehole No.

**202**

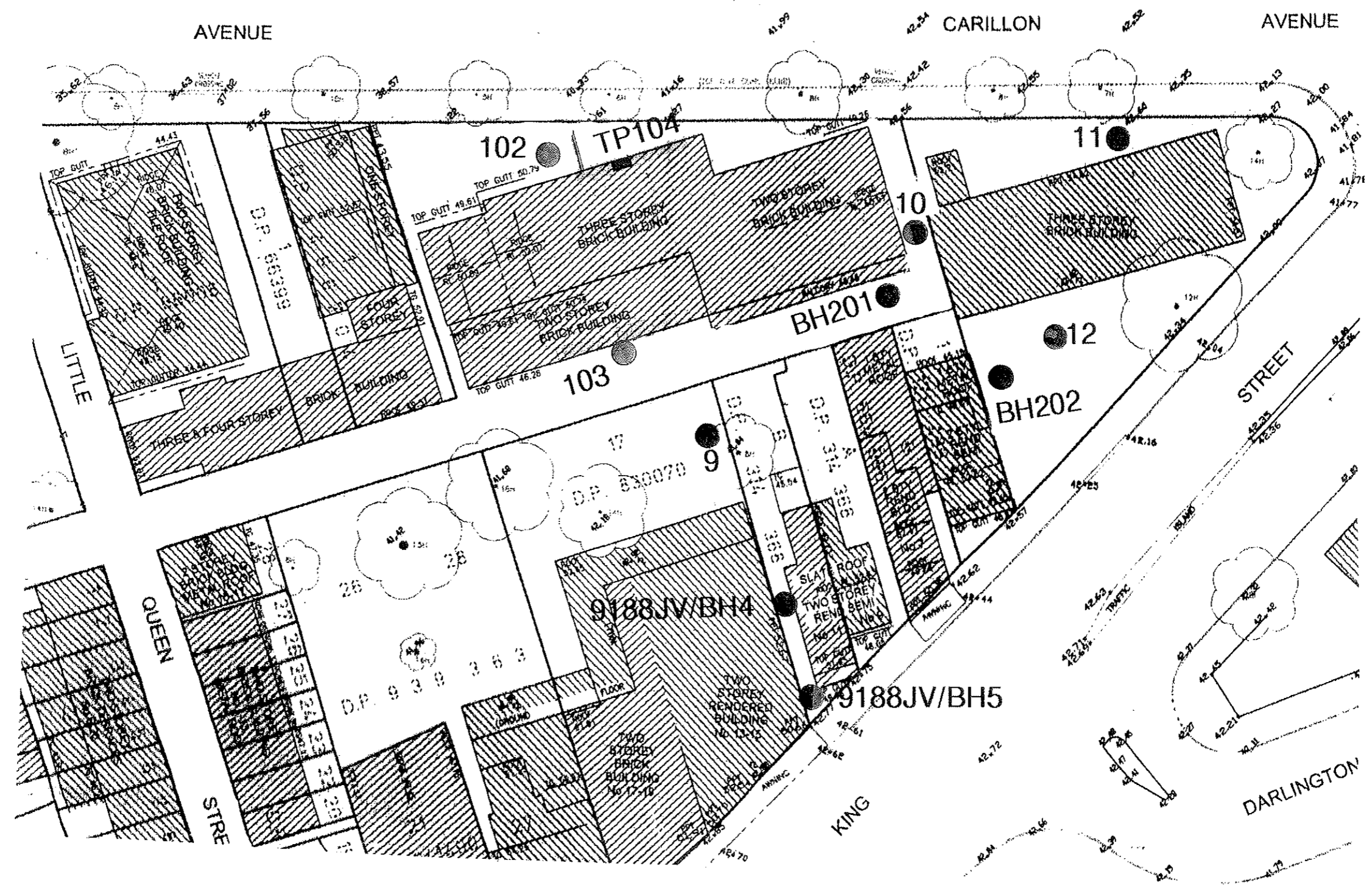
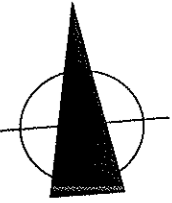
3/3

# CORED BOREHOLE LOG

<b>Client:</b>	MOORE THEOLOGICAL COLLEGE
<b>Project:</b>	PROPOSED LIBRARY BUILDING
<b>Location:</b>	1-11 KING STREET & 2-16 CARILLON AVENUE, NEWTOWN, NSW

<b>Job No.</b> 21871VT2	<b>Core Size:</b> NMLC	<b>R.L. Surface:</b> ≈ 42.4m
<b>Date:</b> 11-2-09	<b>Inclination:</b> VERTICAL	<b>Datum:</b> AHD
<b>Drill Type:</b> JK300	<b>Bearing:</b> -	<b>Logged/Checked by:</b> J.P./[Signature]

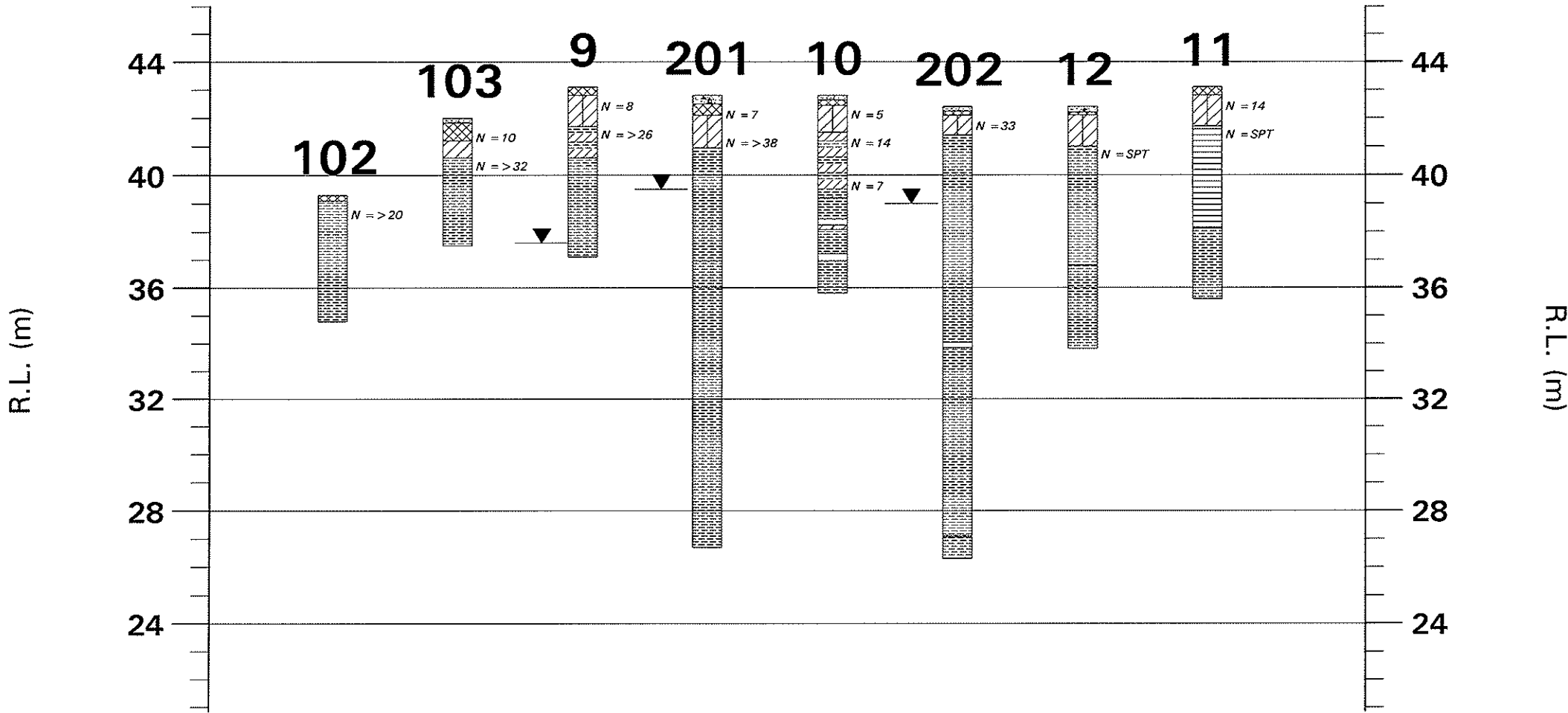
Water Loss/Level	Barrel Lift	Depth (m)	Graphic Log	CORE DESCRIPTION Rock Type, grain characteristics, colour, structure, minor components.	Weathering	Strength	POINT LOAD STRENGTH INDEX I <sub>s</sub> (50)	DEFECT DETAILS													
								DEFECT SPACING (mm)												DESCRIPTION Type, inclination, thickness, planarity, roughness, coating.	
								EL	VL	L	M	H	VH	EH	500	1000	1500	2000	2500	3000	Specific
FULL RETURN		14		SHALE: dark grey, with light grey laminae.	SW	M	X													- Cr, 10mm.t - J, 90°, P, S  - Be, 0.5°, P, S - Be, 0.5°, P, S/J, 60-70°, P, S - Be, 0.5°, P, S - XWS, 40mm.t - XWS, 20mm.t  - J, 40°, P, S - XWS, 15mm.t  - J, 50-70°, Un, R - Cr, 30mm.t - XWS, 30mm.t	
		15		CORE LOSS 0.05m	SW	M	X														
		16		SHALE: dark grey, with light grey laminae.	SW	M	X														
		17		END OF BOREHOLE AT 16.10m																	
		18																			
		19																			



INVESTIGATION LOCATION PLAN



# GRAPHICAL BOREHOLE SUMMARY



	Fill		Clay		Core Loss/Empty	N	SPT "N" VALUE
	Shale		Silty Clay		Siltstone, Mudstone, Claystone	Nc	SOLID CONE BLOW COUNTS PER 150mm
	Concrete		Interbedded Shale and Clay		Observed water level		

NOTE: REFER TO BOREHOLE LOGS

Scale: 1 : 200 (vert) ; NTS (horiz)

Jeffery and Katauskas Pty Ltd

Job No.: 21871VT2

Figure No.: 2



## VIBRATION EMISSION DESIGN GOALS

German Standard DIN 4150 – Part 3: 1986 provides guideline levels of vibration velocity for evaluating the effects of vibration in structures. The limits presented in this standard are generally recognised to be conservative.

The DIN 4150 values (maximum levels measured in any direction at the foundation, OR, maximum levels measured in (x) or (y) horizontal directions, in the plane of the uppermost floor), are summarised in Table 1 below.

It should be noted that peak vibration velocities higher than the minimum figures in Table 1 for low frequencies may be quite “safe”, depending on the frequency content of the vibration and the actual condition of the structure.

It should also be noted that these levels are “safe limits”, up to which no damage due to vibration effects has been observed for the particular class of building. “Damage” is defined by DIN 4150 to include even minor non-structural effects such as superficial cracking in cement render, the enlargement of cracks already present, and the separation of partitions or intermediate walls from load bearing walls. Should damage be observed at vibration levels lower than the “safe limits” then it may be attributed to other causes. DIN 4150 also states that when vibration levels higher than the “safe limits” are present, it does not necessarily follow that damage will occur. Values given are only a broad guide.

**Table 1 DIN 4150 – Structural Damage – Safe Limits for Building Vibration**

Group	Type of Structure	Peak Vibration Velocity in mm/s			
		At Foundation Level At a Frequency of			Plane of Floor of Uppermost Storey
		Less than 10 Hz	10 Hz to 50 Hz	50 Hz to 100 Hz	All Frequencies
1	Buildings used for commercial purposes, industrial buildings and buildings of similar design.	20	20 to 40	40 to 50	40
2	Dwellings and buildings of similar design and/or use.	5	5 to 15	15 to 20	15
3	Structures that because of their particular sensitivity to vibration, do not correspond to those listed in Group 1 and 2 and have intrinsic value (eg buildings that are under a preservation order).	3	3 to 8	8 to 10	8

**Note:** For frequencies above 100 Hz, the higher values in the 50 Hz to 100 Hz column should be used.



## REPORT EXPLANATION NOTES

### INTRODUCTION

These notes have been provided to amplify the geotechnical report in regard to classification methods, field procedures and certain matters relating to the Comments and Recommendations section. Not all notes are necessarily relevant to all reports.

The ground is a product of continuing natural and man-made processes and therefore exhibits a variety of characteristics and properties which vary from place to place and can change with time. Geotechnical engineering involves gathering and assimilating limited facts about these characteristics and properties in order to understand or predict the behaviour of the ground on a particular site under certain conditions. This report may contain such facts obtained by inspection, excavation, probing, sampling, testing or other means of investigation. If so, they are directly relevant only to the ground at the place where and time when the investigation was carried out.

### DESCRIPTION AND CLASSIFICATION METHODS

The methods of description and classification of soils and rocks used in this report are based on Australian Standard 1726, the SAA Site Investigation Code. In general, descriptions cover the following properties – soil or rock type, colour, structure, strength or density, and inclusions. Identification and classification of soil and rock involves judgement and the Company infers accuracy only to the extent that is common in current geotechnical practice.

Soil types are described according to the predominating particle size and behaviour as set out in the attached Unified Soil Classification Table qualified by the grading of other particles present (eg sandy clay) as set out below:

Soil Classification	Particle Size
Clay	less than 0.002mm
Silt	0.002 to 0.06mm
Sand	0.06 to 2mm
Gravel	2 to 60mm

Non-cohesive soils are classified on the basis of relative density, generally from the results of Standard Penetration Test (SPT) as below:

Relative Density	SPT 'N' Value (blows/300mm)
Very loose	less than 4
Loose	4 – 10
Medium dense	10 – 30
Dense	30 – 50
Very Dense	greater than 50

Cohesive soils are classified on the basis of strength (consistency) either by use of hand penetrometer, laboratory testing or engineering examination. The strength terms are defined as follows.

Classification	Unconfined Compressive Strength kPa
Very Soft	less than 25
Soft	25 – 50
Firm	50 – 100
Stiff	100 – 200
Very Stiff	200 – 400
Hard	Greater than 400
Friable	Strength not attainable – soil crumbles

Rock types are classified by their geological names, together with descriptive terms regarding weathering, strength, defects, etc. Where relevant, further information regarding rock classification is given in the text of the report. In the Sydney Basin, 'Shale' is used to describe thinly bedded to laminated siltstone.

### SAMPLING

Sampling is carried out during drilling or from other excavations to allow engineering examination (and laboratory testing where required) of the soil or rock.

Disturbed samples taken during drilling provide information on plasticity, grain size, colour, moisture content, minor constituents and, depending upon the degree of disturbance, some information on strength and structure. Bulk samples are similar but of greater volume required for some test procedures.

Undisturbed samples are taken by pushing a thin-walled sample tube, usually 50mm diameter (known as a U50), into the soil and withdrawing it with a sample of the soil contained in a relatively undisturbed state. Such samples yield information on structure and strength, and are necessary for laboratory determination of shear strength and compressibility. Undisturbed sampling is generally effective only in cohesive soils.

Details of the type and method of sampling used are given on the attached logs.

### INVESTIGATION METHODS

The following is a brief summary of investigation methods currently adopted by the Company and some comments on their use and application. All except test pits, hand auger drilling and portable dynamic cone penetrometers require the use of a mechanical drilling rig which is commonly mounted on a truck chassis.



**Test Pits:** These are normally excavated with a backhoe or a tracked excavator, allowing close examination of the insitu soils if it is safe to descend into the pit. The depth of penetration is limited to about 3m for a backhoe and up to 6m for an excavator. Limitations of test pits are the problems associated with disturbance and difficulty of reinstatement and the consequent effects on close-by structures. Care must be taken if construction is to be carried out near test pit locations to either properly recompact the backfill during construction or to design and construct the structure so as not to be adversely affected by poorly compacted backfill at the test pit location.

**Hand Auger Drilling:** A borehole of 50mm to 100mm diameter is advanced by manually operated equipment. Premature refusal of the hand augers can occur on a variety of materials such as hard clay, gravel or ironstone, and does not necessarily indicate rock level.

**Continuous Spiral Flight Augers:** The borehole is advanced using 75mm to 115mm diameter continuous spiral flight augers, which are withdrawn at intervals to allow sampling and insitu testing. This is a relatively economical means of drilling in clays and in sands above the water table. Samples are returned to the surface by the flights or may be collected after withdrawal of the auger flights, but they can be very disturbed and layers may become mixed. Information from the auger sampling (as distinct from specific sampling by SPTs or undisturbed samples) is of relatively lower reliability due to mixing or softening of samples by groundwater, or uncertainties as to the original depth of the samples. Augering below the groundwater table is of even lesser reliability than augering above the water table.

**Rock Augering:** Use can be made of a Tungsten Carbide (TC) bit for auger drilling into rock to indicate rock quality and continuity by variation in drilling resistance and from examination of recovered rock fragments. This method of investigation is quick and relatively inexpensive but provides only an indication of the likely rock strength and predicted values may be in error by a strength order. Where rock strengths may have a significant impact on construction feasibility or costs, then further investigation by means of cored boreholes may be warranted.

**Wash Boring:** The borehole is usually advanced by a rotary bit, with water being pumped down the drill rods and returned up the annulus, carrying the drill cuttings. Only major changes in stratification can be determined from the cuttings, together with some information from "feel" and rate of penetration.

**Mud Stabilised Drilling:** Either Wash Boring or Continuous Core Drilling can use drilling mud as a circulating fluid to stabilise the borehole. The term 'mud' encompasses a range of products ranging from bentonite to polymers such as Revert or Biogel. The mud tends to mask the cuttings and reliable identification is only possible from intermittent intact sampling (eg from SPT and U50 samples) or from rock coring, etc.

**Continuous Core Drilling:** A continuous core sample is obtained using a diamond tipped core barrel. Provided full core recovery is achieved (which is not always possible in very low strength rocks and granular soils), this technique provides a very reliable (but relatively expensive) method of investigation. In rocks, an NMLC triple tube core barrel, which gives a core of about 50mm diameter, is usually used with water flush. The length of core recovered is compared to the length drilled and any length not recovered is shown as CORE LOSS. The location of losses are determined on site by the supervising engineer; where the location is uncertain, the loss is placed at the top end of the drill run.

**Standard Penetration Tests:** Standard Penetration Tests (SPT) are used mainly in non-cohesive soils, but can also be used in cohesive soils as a means of indicating density or strength and also of obtaining a relatively undisturbed sample. The test procedure is described in Australian Standard 1289, "Methods of Testing Soils for Engineering Purposes" – Test F3.1.

The test is carried out in a borehole by driving a 50mm diameter split sample tube with a tapered shoe, under the impact of a 63kg hammer with a free fall of 760mm. It is normal for the tube to be driven in three successive 150mm increments and the 'N' value is taken as the number of blows for the last 300mm. In dense sands, very hard clays or weak rock, the full 450mm penetration may not be practicable and the test is discontinued.

The test results are reported in the following form:

- In the case where full penetration is obtained with successive blow counts for each 150mm of, say, 4, 6 and 7 blows, as

$$N = 13$$

$$4, 6, 7$$

- In a case where the test is discontinued short of full penetration, say after 15 blows for the first 150mm and 30 blows for the next 40mm, as

$$N > 30$$

$$15, 30/40\text{mm}$$

The results of the test can be related empirically to the engineering properties of the soil.

Occasionally, the drop hammer is used to drive 50mm diameter thin walled sample tubes (U50) in clays. In such circumstances, the test results are shown on the borehole logs in brackets.

A modification to the SPT test is where the same driving system is used with a solid 60° tipped steel cone of the same diameter as the SPT hollow sampler. The solid cone can be continuously driven for some distance in soft clays or loose sands, or may be used where damage would otherwise occur to the SPT. The results of this Solid Cone Penetration Test (SCPT) are shown as "N<sub>c</sub>" on the borehole logs, together with the number of blows per 150mm penetration.



**Static Cone Penetrometer Testing and Interpretation:** Cone penetrometer testing (sometimes referred to as a Dutch Cone) described in this report has been carried out using an Electronic Friction Cone Penetrometer (EFCP). The test is described in Australian Standard 1289, Test F5.1.

In the tests, a 35mm diameter rod with a conical tip is pushed continuously into the soil, the reaction being provided by a specially designed truck or rig which is fitted with an hydraulic ram system. Measurements are made of the end bearing resistance on the cone and the frictional resistance on a separate 134mm long sleeve, immediately behind the cone. Transducers in the tip of the assembly are electrically connected by wires passing through the centre of the push rods to an amplifier and recorder unit mounted on the control truck.

As penetration occurs (at a rate of approximately 20mm per second) the information is output as incremental digital records every 10mm. The results given in this report have been plotted from the digital data.

The information provided on the charts comprise:

- Cone resistance – the actual end bearing force divided by the cross sectional area of the cone – expressed in MPa.
- Sleeve friction – the frictional force on the sleeve divided by the surface area – expressed in kPa.
- Friction ratio – the ratio of sleeve friction to cone resistance, expressed as a percentage.

The ratios of the sleeve resistance to cone resistance will vary with the type of soil encountered, with higher relative friction in clays than in sands. Friction ratios of 1% to 2% are commonly encountered in sands and occasionally very soft clays, rising to 4% to 10% in stiff clays and peats. Soil descriptions based on cone resistance and friction ratios are only inferred and must not be considered as exact.

Correlations between EFCP and SPT values can be developed for both sands and clays but may be site specific.

Interpretation of EFCP values can be made to empirically derive modulus or compressibility values to allow calculation of foundation settlements.

Stratification can be inferred from the cone and friction traces and from experience and information from nearby boreholes etc. Where shown, this information is presented for general guidance, but must be regarded as interpretive. The test method provides a continuous profile of engineering properties but, where precise information on soil classification is required, direct drilling and sampling may be preferable.

**Portable Dynamic Cone Penetrometers:** Portable Dynamic Cone Penetrometer (DCP) tests are carried out by driving a rod into the ground with a sliding hammer and counting the blows for successive 100mm increments of penetration.

Two relatively similar tests are used:

- Cone penetrometer (commonly known as the Scala Penetrometer) – a 16mm rod with a 20mm diameter cone end is driven with a 9kg hammer dropping 510mm (AS1289, Test F3.2). The test was developed initially for pavement subgrade investigations, and correlations of the test results with California Bearing Ratio have been published by various Road Authorities.
- Perth sand penetrometer – a 16mm diameter flat ended rod is driven with a 9kg hammer, dropping 600mm (AS1289, Test F3.3). This test was developed for testing the density of sands (originating in Perth) and is mainly used in granular soils and filling.

## LOGS

The borehole or test pit logs presented herein are an engineering and/or geological interpretation of the subsurface conditions, and their reliability will depend to some extent on the frequency of sampling and the method of drilling or excavation. Ideally, continuous undisturbed sampling or core drilling will enable the most reliable assessment, but is not always practicable or possible to justify on economic grounds. In any case, the boreholes or test pits represent only a very small sample of the total subsurface conditions.

The attached explanatory notes define the terms and symbols used in preparation of the logs.

Interpretation of the information shown on the logs, and its application to design and construction, should therefore take into account the spacing of boreholes or test pits, the method of drilling or excavation, the frequency of sampling and testing and the possibility of other than "straight line" variations between the boreholes or test pits. Subsurface conditions between boreholes or test pits may vary significantly from conditions encountered at the borehole or test pit locations.

## GROUNDWATER

Where groundwater levels are measured in boreholes, there are several potential problems:

- Although groundwater may be present, in low permeability soils it may enter the hole slowly or perhaps not at all during the time it is left open.
- A localised perched water table may lead to an erroneous indication of the true water table.
- Water table levels will vary from time to time with seasons or recent weather changes and may not be the same at the time of construction.
- The use of water or mud as a drilling fluid will mask any groundwater inflow. Water has to be blown out of the hole and drilling mud must be washed out of the hole or 'reverted' chemically if water observations are to be made.



More reliable measurements can be made by installing standpipes which are read after stabilising at intervals ranging from several days to perhaps weeks for low permeability soils. Piezometers, sealed in a particular stratum, may be advisable in low permeability soils or where there may be interference from perched water tables or surface water.

#### **FILL**

The presence of fill materials can often be determined only by the inclusion of foreign objects (eg bricks, steel etc) or by distinctly unusual colour, texture or fabric. Identification of the extent of fill materials will also depend on investigation methods and frequency. Where natural soils similar to those at the site are used for fill, it may be difficult with limited testing and sampling to reliably determine the extent of the fill.

The presence of fill materials is usually regarded with caution as the possible variation in density, strength and material type is much greater than with natural soil deposits. Consequently, there is an increased risk of adverse engineering characteristics or behaviour. If the volume and quality of fill is of importance to a project, then frequent test pit excavations are preferable to boreholes.

#### **LABORATORY TESTING**

Laboratory testing is normally carried out in accordance with Australian Standard 1289 *'Methods of Testing Soil for Engineering Purposes'*. Details of the test procedure used are given on the individual report forms.

#### **ENGINEERING REPORTS**

Engineering reports are prepared by qualified personnel and are based on the information obtained and on current engineering standards of interpretation and analysis. Where the report has been prepared for a specific design proposal (eg. a three storey building) the information and interpretation may not be relevant if the design proposal is changed (eg to a twenty storey building). If this happens, the company will be pleased to review the report and the sufficiency of the investigation work.

Every care is taken with the report as it relates to interpretation of subsurface conditions, discussion of geotechnical aspects and recommendations or suggestions for design and construction. However, the Company cannot always anticipate or assume responsibility for:

- Unexpected variations in ground conditions – the potential for this will be partially dependent on borehole spacing and sampling frequency as well as investigation technique.
- Changes in policy or interpretation of policy by statutory authorities.
- The actions of persons or contractors responding to commercial pressures.

If these occur, the company will be pleased to assist with investigation or advice to resolve any problems occurring.

#### **SITE ANOMALIES**

In the event that conditions encountered on site during construction appear to vary from those which were expected from the information contained in the report, the company requests that it immediately be notified. Most problems are much more readily resolved when conditions are exposed that at some later stage, well after the event.

#### **REPRODUCTION OF INFORMATION FOR CONTRACTUAL PURPOSES**

Attention is drawn to the document *'Guidelines for the Provision of Geotechnical Information in Tender Documents'*, published by the Institution of Engineers, Australia. Where information obtained from this investigation is provided for tendering purposes, it is recommended that all information, including the written report and discussion, be made available. In circumstances where the discussion or comments section is not relevant to the contractual situation, it may be appropriate to prepare a specially edited document. The company would be pleased to assist in this regard and/or to make additional report copies available for contract purposes at a nominal charge.

Copyright in all documents (such as drawings, borehole or test pit logs, reports and specifications) provided by the Company shall remain the property of Jeffery and Katauskas Pty Ltd. Subject to the payment of all fees due, the Client alone shall have a licence to use the documents provided for the sole purpose of completing the project to which they relate. License to use the documents may be revoked without notice if the Client is in breach of any objection to make a payment to us.

#### **REVIEW OF DESIGN**

Where major civil or structural developments are proposed or where only a limited investigation has been completed or where the geotechnical conditions/ constraints are quite complex, it is prudent to have a joint design review which involves a senior geotechnical engineer.


































#### **SITE INSPECTION**

The company will always be pleased to provide engineering inspection services for geotechnical aspects of work to which this report is related.

Requirements could range from:

- i) a site visit to confirm that conditions exposed are no worse than those interpreted, to
- ii) a visit to assist the contractor or other site personnel in identifying various soil/rock types such as appropriate footing or pier founding depths, or
- iii) full time engineering presence on site.

# GRAPHIC LOG SYMBOLS FOR SOILS AND ROCKS

SOIL		ROCK		DEFECTS AND INCLUSIONS	
	FILL		CONGLOMERATE		CLAY SEAM
	TOPSOIL		SANDSTONE		SHEARED OR CRUSHED SEAM
	CLAY (CL, CH)		SHALE		BRECCIATED OR SHATTERED SEAM/ZONE
	SILT (ML, MH)		SILTSTONE, MUDSTONE, CLAYSTONE		IRONSTONE GRAVEL
	SAND (SP, SW)		LIMESTONE		ORGANIC MATERIAL
	GRAVEL (GP, GW)		PHYLLITE, SCHIST		
	SANDY CLAY (CL, CH)		TUFF		
	SILTY CLAY (CL, CH)		GRANITE, GABBRO		
	CLAYEY SAND (SC)		DOLERITE, DIORITE		
	SILTY SAND (SM)		BASALT, ANDESITE		
	GRAVELLY CLAY (CL, CH)		QUARTZITE		
	CLAYEY GRAVEL (GC)				
	SANDY SILT (ML)				
	PEAT AND ORGANIC SOILS				
					<b>OTHER MATERIALS</b>
					CONCRETE
					BITUMINOUS CONCRETE, COAL
					COLLUVIUM





## LOG SYMBOLS

LOG COLUMN	SYMBOL	DEFINITION
Groundwater Record		Standing water level. Time delay following completion of drilling may be shown.
		Extent of borehole collapse shortly after drilling.
		Groundwater seepage into borehole or excavation noted during drilling or excavation.
Samples	ES	Soil sample taken over depth indicated, for environmental analysis.
	U50	Undisturbed 50mm diameter tube sample taken over depth indicated.
	DB	Bulk disturbed sample taken over depth indicated.
	DS	Small disturbed bag sample taken over depth indicated.
	ASB	Soil sample taken over depth indicated, for asbestos screening.
	ASS SAL	Soil sample taken over depth indicated, for acid sulfate soil analysis. Soil sample taken over depth indicated, for salinity analysis.
Field Tests	N = 17 4, 7, 10	Standard Penetration Test (SPT) performed between depths indicated by lines. Individual figures show blows per 150mm penetration. 'R' as noted below.
	N <sub>c</sub> =	Solid Cone Penetration Test (SCPT) performed between depths indicated by lines. Individual figures show blows per 150mm penetration for 60 degree solid cone driven by SPT hammer. 'R' refers to apparent hammer refusal within the corresponding 150mm depth increment.
	5	
	7 3R	
VNS = 25 PID = 100	Vane shear reading in kPa of Undrained Shear Strength. Photoionisation detector reading in ppm (Soil sample headspace test).	
Moisture Condition (Cohesive Soils)  (Cohesionless Soils)	MC > PL	Moisture content estimated to be greater than plastic limit.
	MC ≈ PL	Moisture content estimated to be approximately equal to plastic limit.
	MC < PL	Moisture content estimated to be less than plastic limit.
	D	DRY - runs freely through fingers.
	M	MOIST - does not run freely but no free water visible on soil surface.
	W	WET - free water visible on soil surface.
Strength (Consistency) Cohesive Soils	VS	VERY SOFT - Unconfined compressive strength less than 25kPa
	S	SOFT - Unconfined compressive strength 25-50kPa
	F	FIRM - Unconfined compressive strength 50-100kPa
	St	STIFF - Unconfined compressive strength 100-200kPa
	VSt	VERY STIFF - Unconfined compressive strength 200-400kPa
	H	HARD - Unconfined compressive strength greater than 400kPa
	( )	Bracketed symbol indicates estimated consistency based on tactile examination or other tests.
Density Index/ Relative Density (Cohesionless Soils)	VL	<b>Density Index (I<sub>d</sub>) Range (%)</b> Very Loose < 15
	L	Loose 15-35
	MD	Medium Dense 35-65
	D	Dense 65-85
	VD	Very Dense > 85
	( )	Bracketed symbol indicates estimated density based on ease of drilling or other tests.
Hand Penetrometer Readings	300	Numbers indicate individual test results in kPa on representative undisturbed material unless noted otherwise.
	250	
Remarks	'V' bit	Hardened steel 'V' shaped bit.
	'TC' bit	Tungsten carbide wing bit.
	T 60	Penetration of auger string in mm under static load of rig applied by drill head hydraulics without rotation of augers.



## LOG SYMBOLS

### ROCK MATERIAL WEATHERING CLASSIFICATION

TERM	SYMBOL	DEFINITION
Residual Soil	RS	Soil developed on extremely weathered rock; the mass structure and substance fabric are no longer evident; there is a large change in volume but the soil has not been significantly transported.
Extremely weathered rock	XW	Rock is weathered to such an extent that it has "soil" properties, ie it either disintegrates or can be remoulded, in water.
Distinctly weathered rock	DW	Rock strength usually changed by weathering. The rock may be highly discoloured, usually by ironstaining. Porosity may be increased by leaching, or may be decreased due to deposition of weathering products in pores.
Slightly weathered rock	SW	Rock is slightly discoloured but shows little or no change of strength from fresh rock.
Fresh rock	FR	Rock shows no sign of decomposition or staining.

### ROCK STRENGTH

Rock strength is defined by the Point Load Strength Index ( $I_s$  50) and refers to the strength of the rock substance in the direction normal to the bedding. The test procedure is described by the International Journal of Rock Mechanics, Mining, Science and Geomechanics, Abstract Volume 22, No 2, 1985.

TERM	SYMBOL	$I_s$ (50) MPa	FIELD GUIDE
Extremely Low:	EL	0.03	Easily remoulded by hand to a material with soil properties.
Very Low:	VL	0.1	May be crumbled in the hand. Sandstone is "sugary" and friable.
Low:	L	0.3	A piece of core 150mm long x 50mm dia. may be broken by hand and easily scored with a knife. Sharp edges of core may be friable and break during handling.
Medium Strength:	M	1	A piece of core 150mm long x 50mm dia. can be broken by hand with difficulty. Readily scored with knife.
High:	H	3	A piece of core 150mm long x 50mm dia. core cannot be broken by hand, can be slightly scratched or scored with knife; rock rings under hammer.
Very High:	VH	10	A piece of core 150mm long x 50mm dia. may be broken with hand-held pick after more than one blow. Cannot be scratched with pen knife; rock rings under hammer.
Extremely High:	EH		A piece of core 150mm long x 50mm dia. is very difficult to break with hand-held hammer. Rings when struck with a hammer.

### ABBREVIATIONS USED IN DEFECT DESCRIPTION

ABBREVIATION	DESCRIPTION	NOTES
Be	Bedding Plane Parting	Defect orientations measured relative to the normal to the long core axis (ie relative to horizontal for vertical holes)
CS	Clay Seam	
J	Joint	
P	Planar	
Un	Undulating	
S	Smooth	
R	Rough	
IS	Ironstained	
XWS	Extremely Weathered Seam	
Cr	Crushed Seam	
60t	Thickness of defect in millimetres	

# APPENDIX



Borehole No.

9

1/1

## BOREHOLE LOG

Client: MOORE THEOLOGICAL COLLEGE  
 Project: PROPOSED REDEVELOPMENT  
 Location: KING STREET AND CARILLON AVENUE, NEWTOWN, NSW

Job No. 21871VT      Method: SPIRAL AUGER      R.L. Surface: ≈ 43.1m  
 Date: 30-1-08      JK250      Datum: AHD  
 Logged/Checked by: W.W./G.F./

Groundwater Record	SAMPLES				Field Tests	Depth (m)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition/ Weathering	Strength/ Rel. Density	Hand Penetrometer Readings (kPa.)	Remarks
	FS	USO	DB	DS									
						0			FILL: Silty clay, medium to high plasticity, red brown and dark grey, with root fibres, fine to medium grained gravel and a trace of fine to medium grained sand.	MC > PL			GRASS COVER
				N = 8 4,4,4		1		CL-CH	SILTY CLAY: medium to high plasticity, orange brown and light grey, with fine to medium ironstone gravel.	MC > PL	VSt	270 390 260	
				N > 26 19,7/ 10mm REFUSAL		2		-	INTERBEDDED SILTY CLAY: medium plasticity, light grey mottled orange brown, with fine to medium grained ironstone gravel and SHALE: grey, with iron indurated bands.	MC > PL/ XW	VSt/ EL		LOW TO MODERATE 'TC' BIT RESISTANCE
						3		-	SHALE: dark grey, with iron indurated band.	DW	L		MODERATE RESISTANCE
						4							
						5							
						6							
						6.0			END OF BOREHOLE AT 6.0m				
						7							

ON COMPLETION



Borehole No.

**10**

1/2

## BOREHOLE LOG

**Client:** MOORE THEOLOGICAL COLLEGE  
**Project:** PROPOSED REDEVELOPMENT  
**Location:** KING STREET AND CARILLON AVENUE, NEWTOWN, NSW

**Job No.** 21871VT      **Method:** SPIRAL AUGER      **R.L. Surface:** ≈ 42.8m  
**Date:** 31-1-08      **JK250**      **Datum:** AHD  
**Logged/Checked by:** G.F./

Groundwater Record	SAMPLES			Field Tests	Depth (m)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition/ Weathering	Strength/ Ref. Density	Hand Penetrometer Readings (kPa.)	Remarks
	FS	USO	DB									
DRY ON COMPLETION OF AUGER DRILLING					0			CONCRETE: 160mm.t				
					0.5		CL-CH	FILL: Sandy gravel, fine to medium grained, brown. SILTY CLAY: medium to high plasticity, light grey mottled brown and orange brown, with fine to medium grained ironstone gravel.	D MC > PL	- VSt	- 360 250 290	
				N = 5 3,2,3	1							
				N = 14 5,7,7	2			INTERBEDDED SILTY CLAY: low to medium plasticity, light grey, with fine to medium grained ironstone gravel and SHALE: light grey, with iron indurated bands.	MC < PL/ XW	(St) /EL	-	VERY LOW 'TC' BIT RESISTANCE WITH LOW BANDS
			N = 7 2,2,5	3								
				4				SHALE: grey and brown, with iron indurated bands.	DW	L-M H	-	MODERATE RESISTANCE MODERATE TO HIGH RESISTANCE
					4.1			REFER TO CORED BOREHOLE LOG				CASING REAMED INTO SHALE FROM 4.1m TO 4.35m
					5							
					6							
					7							

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JOB NO 21871VT BH 10 START CORING AT 4.35m

4

CORE LOSS 0.21m

5

CORE LOSS 0.26m

6

7

END AT 7.00m



Borehole No.  
**10**  
2/2

# CORED BOREHOLE LOG

**Client:** MOORE THEOLOGICAL COLLEGE  
**Project:** PROPOSED REDEVELOPMENT  
**Location:** KING STREET AND CARILLON AVENUE, NEWTOWN, NSW

**Job No.** 21871VT      **Core Size:** NMLC      **R.L. Surface:** ≈ 42.8m  
**Date:** 31-1-08      **Inclination:** VERTICAL      **Datum:** AHD  
**Drill Type:** JK250      **Bearing:** -      **Logged/Checked by:** G.F./

Core Loss/Level	Barrel Lift	Depth (m)	Graphic Log	CORE DESCRIPTION Rock Type, grain characteristics, colour, structure, minor components.	Weathering	Strength	POINT LOAD STRENGTH INDEX I <sub>s</sub> (50)	DEFECT DETAILS												
								DEFECT SPACING (mm)										DESCRIPTION Type, inclination, thickness, planarity, roughness, coating.		
								500	300	100	50	30	10	Specific	General					
		3																		
		4		START CORING AT 4.35m																
		5		SILTY CLAY: medium plasticity, light grey. SHALE: grey and brown, with iron indurated bands.	MC>PL DW	VSt L	X													
FULL RETURN		6		SHALE: grey and brown, with iron indurated bands. SHALE: dark grey, with iron indurated bands.	DW	L L-M	X X X													
		7		END OF BOREHOLE AT 7.0m																
		8																		
		9																		



Borehole No.

**11**

1/2

## BOREHOLE LOG

**Client:** MOORE THEOLOGICAL COLLEGE  
**Project:** PROPOSED REDEVELOPMENT  
**Location:** KING STREET AND CARILLON AVENUE, NEWTOWN, NSW

**Job No.** 21871VT      **Method:** SPIRAL AUGER      **R.L. Surface:** ≈ 43.1m  
**Date:** 30-1-08      **JK250**      **Datum:** AHD  
**Logged/Checked by:** G.F./

Groundwater Record	SAMPLES			Field Tests	Depth (m)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition/Weathering	Strength/Rel. Density	Hand Penetrometer Readings (kPa.)	Remarks
	ES	USO	DB									
DRY ON COMPLETION					0			TOPSOIL/FILL: Clayey silt, low plasticity, light brown.				GRASS COVER
				N = 14 5,7,7	1		CH	SILTY CLAY: high plasticity, light grey mottled orange brown, with root fibres and fine to medium grained ironstone gravel.	MC < PL	H	> 600	
				SPT 10/20mm REFUSAL	2		-	SILTSTONE: light grey, with iron indurated bands.	XW	EL	-	MODERATE RESISTANCE WITH HIGH BANDS
					3				DW	M		
					4							
					5		-	SHALE: grey, with iron indurated and L strength bands.	XW-DW	EL-VL	-	VERY LOW 'TC' BIT RESISTANCE WITH LOW BANDS
					6							
					7			SHALE: dark grey.	DW	M		MODERATE RESISTANCE



Borehole No.

11

2/2

## BOREHOLE LOG

Client: MOORE THEOLOGICAL COLLEGE  
 Project: PROPOSED REDEVELOPMENT  
 Location: KING STREET AND CARILLON AVENUE, NEWTOWN, NSW

Job No. 21871VT      Method: SPIRAL AUGER      R.L. Surface: ≈ 43.1m  
 Date: 30-1-08      JK250      Datum: AHD  
 Logged/Checked by: G.F./

Groundwater Record	SAMPLES				Field Tests	Depth (m)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition/ Weathering	Strength/ Rel. Density	Hand Penetrometer Readings (kPa.)	Remarks
	FS	USO	DB	DS									
									SHALE: dark grey.	DW	M		MODERATE RESISTANCE
											H		HIGH RESISTANCE
						8			END OF BOREHOLE AT 7.5m				
						9							
						10							
						11							
						12							
						13							
						14							



Borehole No.

12

1/2

## BOREHOLE LOG

Client:		MOORE THEOLOGICAL COLLEGE											
Project:		PROPOSED REDEVELOPMENT											
Location:		KING STREET AND CARILLON AVENUE, NEWTOWN, NSW											
Job No. 21871VT		Method: SPIRAL AUGER				R.L. Surface: ≈ 42.4m							
Date: 30-1-08		JK350				Datum: AHD							
Logged/Checked by: G.F./													
Underwater Record	SAMPLES				Field Tests	Depth (m)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition/Weathering	Strength/Rel. Density	Hand Penetrometer Readings (kPa.)	Remarks
	ES	USO	DB	DS									
DRY ON COMPLETION OF AUGER - ING						0		-	CONCRETE: 200mm.t				
						0.5		CL	FILL: Sandy gravel, fine to medium grained, brown and grey. SILTY CLAY: medium plasticity, light grey mottled orange brown and red brown, with fine to medium grained ironstone gravel.	MC > PL	(VSt)		
					SPT 15/20mm REFUSAL	2		-	SHALE: grey and brown, with iron indurated bands and XW bands.	XW DW	EL L		LOW 'TC' BIT RESISTANCE WITH VERY LOW BANDS
						5			as above, but without XW bands.		M		MODERATE RESISTANCE
						6			REFER TO CORED BOREHOLE LOG				
						7							

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JOB NO: 21871YT , BH 12 ; START CORING AT 5.62m

5

6

7

8

END AT 8.59m



Borehole No.  
**12**  
2/2

# CORED BOREHOLE LOG

**Client:** MOORE THEOLOGICAL COLLEGE  
**Project:** PROPOSED REDEVELOPMENT  
**Location:** KING STREET AND CARILLON AVENUE, NEWTOWN, NSW

**Job No.** 21871VT      **Core Size:** NMLC      **R.L. Surface:** ≈ 42.4m  
**Date:** 30-1-08      **Inclination:** VERTICAL      **Datum:** AHD  
**Drill Type:** JK250      **Bearing:** -      **Logged/Checked by:** G.F./

Core Loss/Level	Barrel Lift	Depth (m)	Graphic Log	CORE DESCRIPTION Rock Type, grain characteristics, colour, structure, minor components.	Weathering	Strength	POINT LOAD STRENGTH INDEX I <sub>s</sub> (50)	DEFECT DETAILS										
								DEFECT SPACING (mm)		DESCRIPTION Type, inclination, thickness, planarity, roughness, coating.								
								500	300	100	50	30	10	Specific	General			
		5		START CORING AT 5.62m														
		6		SHALE: dark grey, with light grey laminae, bedded at 0-5°.	DW	M	VL L M H VH EL	500 300 100 50 30 10	<ul style="list-style-type: none"> <li>- XWS, 5mm.t</li> <li>- XWS, 5mm.t</li> <li>- J, 40-50°, P, S</li> </ul>									
		7							<ul style="list-style-type: none"> <li>- CS, 40mm.t</li> <li>- XWS, 5mm.t</li> <li>- XWS, 5mm.t</li> <li>- XWS, 40mm.t</li> </ul>									
		8							<ul style="list-style-type: none"> <li>- XWS, 5mm.t</li> </ul>									
		9		END OF BOREHOLE AT 8.59m					<ul style="list-style-type: none"> <li>- J, 85-90°, P, S</li> <li>- J, 30-40°, P, S</li> <li>- HEALED JOINT</li> </ul>									
		10							<ul style="list-style-type: none"> <li>- J, 40-50°, P, S</li> </ul>									
		11																



Borehole No.

**102**

1/1

## BOREHOLE LOG

Client: MOORE THEOLOGICAL COLLEGE  
 Project: PROPOSED ADDITIONS  
 Location: 2-16 CARILLON AVENUE, NEWTOWN, NSW

Job No. 21871VT      Method: SPIRAL AUGER      R.L. Surface: ≈ 39.3m  
 Date: 30-7-08      JK250      Datum: AHD  
 Logged/Checked by: J.P.

Underwater Record	SAMPLES				Field Tests	Depth (m)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition/ Weathering	Strength/ Rel. Density	Hand Penetrometer Readings (kPa.)	Remarks
	ES	USO	DB	DS									
DRY ON COMPLETION						0			TOPSOIL/FILL: Silty clay, with roots.				GRASS COVER
					N > 20 20/150mm REFUSAL	1			SHALE: grey mottled dark brown.	XW-DW	EL-VL		
						1			SHALE: grey brown and orange brown, with iron indurated bands, extremely weathered bands and clay bands.	DW	L		LOW 'TC' BIT RESISTANCE
						4			SHALE: dark brown to dark grey, with iron indurated bands.				
						4.5		END OF BOREHOLE AT 4.5m					
						5							
						6							
						7							

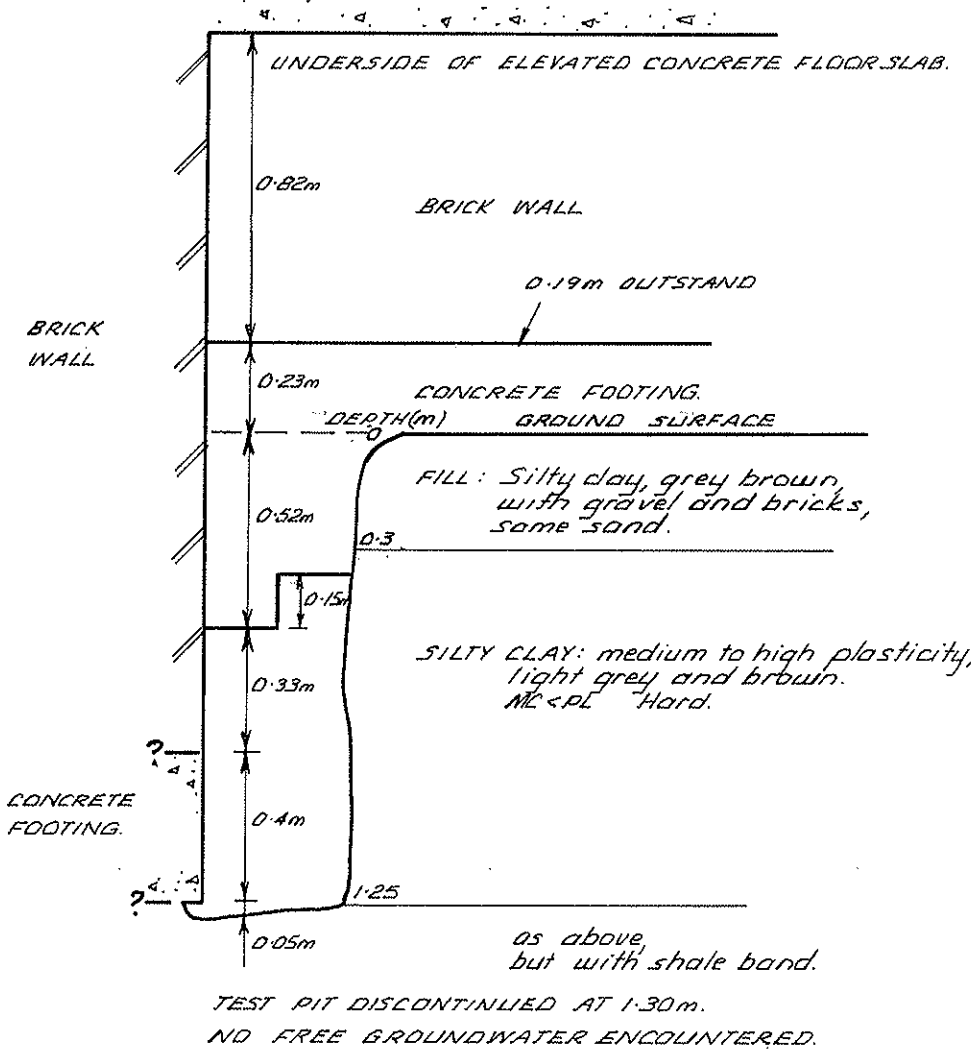


Borehole No.  
**103**  
1/1

# BOREHOLE LOG

<b>Client:</b>	MOORE THEOLOGICAL COLLEGE		
<b>Project:</b>	PROPOSED ADDITIONS		
<b>Location:</b>	2-16 CARILLON AVENUE, NEWTOWN, NSW		
<b>Job No.</b> 21871VT	<b>Method:</b> SPIRAL AUGER JK250	<b>R.L. Surface:</b> ≈ 42.0m	
<b>Date:</b> 30-7-08		<b>Datum:</b> AHD	
<b>Logged/Checked by:</b> J.P./s			

Groundwater Record	SAMPLES				Field Tests	Depth (m)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition/ Weathering	Strength/ Rel. Density	Hand Penetrometer Readings (kPa.)	Remarks
	ES	USO	DB	DS									
DRY ON COMPLETION						0		-	CONCRETE: 160mm.t FILL: Sand, medium to coarse grained, dark grey, with gravel.	-	-	-	NO OBSERVED REINFORCEMENT
					N = 10 3,4,6	1		CL-CH	SILTY CLAY: medium to high plasticity, grey, light orange brown and red brown.	MC≈PL	VSt	310 300 290	
					N > 32 12,20, 20/75mm REFUSAL	2		-	SHALE: light grey mottled dark brown, with clay and occasional VL-L strength iron indurated bands.	XW	EL	-	LOW 'TC' BIT RESISTANCE
									SHALE: orange brown and grey, with iron indurated bands, extremely weathered bands and clay bands.				
						4		as above, but dark grey.	DW	L-M		LOW RESISTANCE WITH MODERATE BANDS	
						5			END OF BOREHOLE AT 4.5m				
						6							
						7							



**TEST PIT 104**  
**CROSS SECTIONAL SKETCH**  
**LOOKING EAST**

SCALE (m)



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Report No. 21871VT

Figure No. 3



Borehole No.

4

## BOREHOLE LOG

Client: *TAYLOR THOMSON WHITTING PTY. LTD.*  
 Project: *LECTURE HALL*  
 Location: *MOORE THEOLOGICAL COLLEGE, 13-19 KING STREET, NEWTOWN, N.S.W.*  
 Job No. *9188JV* Method: *HAND AUGER*  
 Date: *6-11-92*

roundwater record	Samples	Field Tests	Depth (m.)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition	Consistency/ Rel. Density	Hand Penetrometer KPa. Readings	Remarks
TRY DN COMPL- ETION 2 AFTER 2 HRS	DS	REFER TO SCALA SHEET	1		CH	CONCRETE SLAB: 120mm. T. over FILL: Clay, high plasticity, brown & dark brown	MC7PL	(ST.)  Y. ST - H		APPEARS POORLY COMPACTED
	DS					CLAY: high plasticity, grey brown with a trace of ironstone gravel.	MC7PL			RESIDUAL
	DS					END OF BOREHOLE AT 1.8m.	MC<PL			HAND AUGER REFUSAL
			2							
			3							
			4							
			5							
			6							

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Logged by : *P. Wright.* Chkd. by : *Y.P.*



Borehole No.

5

## BOREHOLE LOG

Client: TAYLOR THOMSON WHITTING PTY. LTD.  
 Project: LECTURE HALL  
 Location: MOORE THEOLOGICAL COLLEGE, 13-19 KING STREET, NEWTOWN, N.S.W.  
 Job No. 9188JV Method: SPIRAL AUGER  
 Date: 6-11-92 JACRO RIG

Groundwater record	Samples	Field Tests	Depth (m.)	Graphic Log	Unified Classification	DESCRIPTION	Moisture Condition	Consistency/Rel. Density	Hand Penetrometer Readings kPa.	Remarks
ON COMPLETION          AFTER 4 HRS.		REFER TO SCALA SHEET				CONCRETE SLAB: 40mm. T. over	MC > PL			APPEARS POORLY COMPACTED
	DS				CH	FILL: Clay, high plasticity, brown & grey, with traces of crushed rock & shale gravel & brick fragments.	MC > PL	V. ST.		
	DS	N=26 3, 7, 19	1			CLAY: high plasticity, grey brown, with some ironstone gravel. as above, but with bands of shale, light grey, moderately weathered, weak to medium strong.		(H)	200 230 280	RESIDUAL
	DS					SHALE: light grey, with red bands, highly weathered, weak to medium strong, with iron-infiltrated bands.				LOW TO MODERATE "TC" BIT RESISTANCE
	DS					as above, but grey brown, highly to extremely weathered, very to extremely weak.				LOW RESISTANCE
	DS	N=30 13, 12/50mm. REFUSAL	3			as above, but extremely weathered, extremely weak.				LOW TO MODERATE RESISTANCE
	DS					SHALE: grey, moderately weathered, weak to medium strong.				MODERATE RESISTANCE
	DS					as above, but dark grey, moderately weathered, medium strong.				
				5		END OF BOREHOLE AT 4.5m.				
				6						

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Logged by : P. Wright. Chkd. by : *[Signature]*



## SCALA PENETRATION TEST RESULTS

Client: *TAYLOR THOMSON WHITTING PTY. LTD.*  
 Project: *LECTURE HALL*  
 Location: *MOORE THEOLOGICAL COLLEGE, 13-19 KING STREET, NEWTOWN.*

Job No.: *918BJV* Hammer Weight & Drop: 9kg/510mm  
 Date: *6-11-92* Rod Diameter: 16mm  
 Tested By: *P.W.* Point Diameter: 20mm

Number of Blows per 100mm Penetration

Depth (mm) \ Test Location	1	2	3	4	5		
0 - 100	1	1	EXCAVATED		EXCAVATED		
100 - 200	2	}	1	EXCAVATED			
200 - 300	1		2		1		
300 - 400	1	3	5	1	1		
400 - 500	2	4	4	1	1		
500 - 600	3	5	4	1	1		
600 - 700	1	5	4	1	2		
700 - 800	4	5	3	3	2		
800 - 900	3	7	4	5	4		
900 - 1000	4	5	5	5	6		
1000 - 1100	5	5	4	9	26		
1100 - 1200	7	6	4	12	28		
1200 - 1300	8	<del>30/70mm</del> REFUSAL	9	15	END		
1300 - 1400	23		<del>30/60mm</del> REFUSAL	22			
1400 - 1500	REFUSAL			<del>25/60mm</del>			
1500 - 1600				EXCAVATED			
1600 - 1700							
1700 - 1800				32			
1800 - 1900				END			
1900 - 2000							
2000 - 2100							
2100 - 2200							
2200 - 2300							
2300 - 2400							
2400 - 2500							
2500 - 2600							
2600 - 2700							
2700 - 2800							
2800 - 2900							
2900 - 3000							

Remarks:



## LOG SYMBOLS – CORED BOREHOLE

### DEGREE OF WEATHERING

TERM	SYMBOL	DEFINITION
Extremely Weathered	EW	Rock substance affected by weathering to the extent that the rock exhibits soil properties - i.e. it can be remoulded and can be classified according to the Unified Classification System, but the texture of the original rock is still evident.
Highly Weathered	HW	Rock substance affected by weathering to the extent that limonite staining or bleaching affects the whole of the rock substance and other signs of chemical or physical decomposition are evident. Porosity and strength may be increased or decrease compared to the fresh rock usually as a result of iron leaching or deposition. The colour and strength of the original fresh rock substance is no longer recognisable.
Moderately Weathered	MW	Rock substance affected by weathering to the extent that staining extends throughout the whole of the rock substance and the original colour of the fresh rock is no longer recognisable.
Slightly Weathered	SW	Rock substance affected by weathering to the extent that partial staining or discolouration of the rock substance usually by limonite has taken place. The colour and texture of the fresh rock is recognisable.
Fresh	Fr	Rock substance unaffected by weathering.

### ROCK STRENGTH

Rock strength is defined by the Point Load Strength Index ( $I_s(50)$ ) and refers to the strength of the rock substance in the direction normal to the bedding. The test procedure is described by the International Society of Rock Mechanics (Reference).

TERM	$I_s(50)$ MPa	FIELD GUIDE	SYMBOL
Extremely Weak:	0.03	Easily remoulded by hand to a material with soil properties.	EW
Very Weak:		May be crumbled in the hand. Sandstone is "sugary" and friable.	VW
Weak:	0.1	A piece of core 150 mm long x 50 mm dia. may be broken by hand and easily scored with a knife. Sharp edges of core may be friable and break during handling.	W
	0.3		
Medium Strong:	1	A piece of core 150 mm long x 50 mm dia. can be broken by hand with considerable difficulty. Readily scored with knife.	MS
Strong:	3	A piece of core 150 mm long x 50 mm dia. core cannot be broken by unaided hands, can be slightly scratched or scored with knife.	S
	10		
Very Strong:	10	A piece of core 150 mm long x 50 mm dia. may be broken readily with hand held hammer. Cannot be scratched with pen knife.	VS
Extremely Strong:		A piece of core 150 mm long x 50 mm dia. is difficult to break with hand held hammer. Rings when struck with a hammer.	ES

### DEGREE OF FRACTURING

This classification applies to diamond drill cores and refers to the spacing of all types of natural fractures along which the core is discontinuous. These include bedding plane partings, joints and other rock defects, but exclude known artificial fractures such as drilling breaks.

TERM	DESCRIPTION
Fragmented	The core is comprised primarily of fragments of length less than 20 mm, and mostly of width less than the core diameter.
Highly Fractured:	Core lengths are generally less than 20 mm–40 mm with occasional fragments.
Fractured:	Core lengths are mainly 30 mm–100 mm with occasional shorter and longer section.
Slightly Fractured:	Core lengths are generally 300 mm–1000 mm with occasional longer sections and occasional sections of 100 mm–300 mm.
Unbroken:	The core does not contain any fracture.



## REPORT EXPLANATION NOTES

### INTRODUCTION

These notes have been provided to amplify the geotechnical report in regard to classification methods, field procedures and certain matters relating to the Comments and Recommendations section. Not all notes are necessarily relevant to all reports.

The ground is a product of continuing natural and man-made processes and therefore exhibits a variety of characteristics and properties which vary from place to place and can change with time. Geotechnical engineering involves gathering and assimilating limited facts about these characteristics and properties in order to understand or predict the behaviour of the ground on a particular site under certain conditions. This report may contain such facts obtained by inspection, excavation, probing, sampling, testing or other means of investigation. If so, they are directly relevant only to the ground at the place where and time when the investigation was carried out.

### DESCRIPTION AND CLASSIFICATION METHODS

The methods of description and classification of soils and rocks used in this report are based on Australian Standard 1726, the SAA Site Investigation Code. In general, descriptions cover the following properties – soil or rock type, colour, structure, strength or density, and inclusions. Identification and classification of soil and rock involves judgement and the Company infers accuracy only to the extent that is common in current geotechnical practice.

Soil types are described according to the predominating particle size and behaviour as set out in the attached Unified Soil Classification Table qualified by the grading of other particles present (eg sandy clay) as set out below:

Soil Classification	Particle Size
Clay	less than 0.002mm
Silt	0.002 to 0.06mm
Sand	0.06 to 2mm
Gravel	2 to 60mm

Non-cohesive soils are classified on the basis of relative density, generally from the results of Standard Penetration Test (SPT) as below:

Relative Density	SPT 'N' Value (blows/300mm)
Very loose	less than 4
Loose	4 – 10
Medium dense	10 – 30
Dense	30 – 50
Very Dense	greater than 50

Cohesive soils are classified on the basis of strength (consistency) either by use of hand penetrometer, laboratory testing or engineering examination. The strength terms are defined as follows.

Classification	Unconfined Compressive Strength kPa
Very Soft	less than 25
Soft	25 – 50
Firm	50 – 100
Stiff	100 – 200
Very Stiff	200 – 400
Hard	Greater than 400
Friable	Strength not attainable – soil crumbles

Rock types are classified by their geological names, together with descriptive terms regarding weathering, strength, defects, etc. Where relevant, further information regarding rock classification is given in the text of the report. In the Sydney Basin, "Shale" is used to describe thinly bedded to laminated siltstone.

### SAMPLING

Sampling is carried out during drilling or from other excavations to allow engineering examination (and laboratory testing where required) of the soil or rock.

Disturbed samples taken during drilling provide information on plasticity, grain size, colour, moisture content, minor constituents and, depending upon the degree of disturbance, some information on strength and structure. Bulk samples are similar but of greater volume required for some test procedures.

Undisturbed samples are taken by pushing a thin-walled sample tube, usually 50mm diameter (known as a U50), into the soil and withdrawing it with a sample of the soil contained in a relatively undisturbed state. Such samples yield information on structure and strength, and are necessary for laboratory determination of shear strength and compressibility. Undisturbed sampling is generally effective only in cohesive soils.

Details of the type and method of sampling used are given on the attached logs.

### INVESTIGATION METHODS

The following is a brief summary of investigation methods currently adopted by the Company and some comments on their use and application. All except test pits, hand auger drilling and portable dynamic cone penetrometers require the use of a mechanical drilling rig which is commonly mounted on a truck chassis.



**Test Pits:** These are normally excavated with a backhoe or a tracked excavator, allowing close examination of the insitu soils if it is safe to descend into the pit. The depth of penetration is limited to about 3m for a backhoe and up to 6m for an excavator. Limitations of test pits are the problems associated with disturbance and difficulty of reinstatement and the consequent effects on close-by structures. Care must be taken if construction is to be carried out near test pit locations to either properly recompact the backfill during construction or to design and construct the structure so as not to be adversely affected by poorly compacted backfill at the test pit location.

**Hand Auger Drilling:** A borehole of 50mm to 100mm diameter is advanced by manually operated equipment. Premature refusal of the hand augers can occur on a variety of materials such as hard clay, gravel or ironstone, and does not necessarily indicate rock level.

**Continuous Spiral Flight Augers:** The borehole is advanced using 75mm to 115mm diameter continuous spiral flight augers, which are withdrawn at intervals to allow sampling and insitu testing. This is a relatively economical means of drilling in clays and in sands above the water table. Samples are returned to the surface by the flights or may be collected after withdrawal of the auger flights, but they can be very disturbed and layers may become mixed. Information from the auger sampling (as distinct from specific sampling by SPTs or undisturbed samples) is of relatively lower reliability due to mixing or softening of samples by groundwater, or uncertainties as to the original depth of the samples. Augering below the groundwater table is of even lesser reliability than augering above the water table. Use can be made of a Tungsten Carbide (TC) bit for auger drilling into rock to indicate rock quality and continuity by variation in drilling resistance and from examination of recovered rock fragments.

**Wash Boring:** The borehole is usually advanced by a rotary bit, with water being pumped down the drill rods and returned up the annulus, carrying the drill cuttings. Only major changes in stratification can be determined from the cuttings, together with some information from "feel" and rate of penetration.

**Mud Stabilised Drilling:** Either Wash Boring or Continuous Core Drilling can use drilling mud as a circulating fluid to stabilise the borehole. The term "mud" encompasses a range of products ranging from bentonite to polymers such as Revert or Biogel. The mud tends to mask the cuttings and reliable identification is only possible from intermittent intact sampling (eg from SPT and U50 samples) or from rock coring, etc.

**Continuous Core Drilling:** A continuous core sample is obtained using a diamond tipped core barrel. Provided full core recovery is achieved (which is not always possible in very low strength rocks and granular soils), this technique provides a very reliable (but relatively expensive) method of investigation. In rocks, an NMLC triple tube core barrel, which gives a core of about 50mm diameter, is usually used with water flush. The length of core recovered is compared to the length drilled and any length not recovered is shown as CORE LOSS. The location of losses are determined on site by the supervising engineer; where the location is uncertain, the loss is placed at the top end of the drill run.

**Standard Penetration Tests:** Standard Penetration Tests (SPT) are used mainly in non-cohesive soils, but can also be used in cohesive soils as a means of indicating density or strength and also of obtaining a relatively undisturbed sample. The test procedure is described in Australian Standard 1289, "Methods of Testing Soils for Engineering Purposes" – Test F3.1.

The test is carried out in a borehole by driving a 50mm diameter split sample tube with a tapered shoe, under the impact of a 63kg hammer with a free fall of 760mm. It is normal for the tube to be driven in three successive 150mm increments and the 'N' value is taken as the number of blows for the last 300mm. In dense sands, very hard clays or weak rock, the full 450mm penetration may not be practicable and the test is discontinued.

The test results are reported in the following form:

- In the case where full penetration is obtained with successive blow counts for each 150mm of, say, 4, 6 and 7 blows, as

$$N = 13$$

$$4, 6, 7$$

- In a case where the test is discontinued short of full penetration, say after 15 blows for the first 150mm and 30 blows for the next 40mm, as

$$N > 30$$

$$15, 30/40\text{mm}$$

The results of the test can be related empirically to the engineering properties of the soil.

Occasionally, the drop hammer is used to drive 50mm diameter thin walled sample tubes (U50) in clays. In such circumstances, the test results are shown on the borehole logs in brackets.

A modification to the SPT test is where the same driving system is used with a solid 60° tipped steel cone of the same diameter as the SPT hollow sampler. The solid cone can be continuously driven for some distance in soft clays or loose sands, or may be used where damage would otherwise occur to the SPT. The results of this Solid Cone Penetration Test (SCPT) are shown as "N<sub>c</sub>" on the borehole logs, together with the number of blows per 150mm penetration.



### Static Cone Penetrometer Testing and Interpretation:

Cone penetrometer testing (sometimes referred to as a Dutch Cone) described in this report has been carried out using an Electronic Friction Cone Penetrometer (EFCP). The test is described in Australian Standard 1289, Test F5.1.

In the tests, a 35mm diameter rod with a conical tip is pushed continuously into the soil, the reaction being provided by a specially designed truck or rig which is fitted with an hydraulic ram system. Measurements are made of the end bearing resistance on the cone and the frictional resistance on a separate 134mm long sleeve, immediately behind the cone. Transducers in the tip of the assembly are electrically connected by wires passing through the centre of the push rods to an amplifier and recorder unit mounted on the control truck.

As penetration occurs (at a rate of approximately 20mm per second) the information is output as incremental digital records every 10mm. The results given in this report have been plotted from the digital data.

The information provided on the charts comprise:

- Cone resistance – the actual end bearing force divided by the cross sectional area of the cone – expressed in MPa.
- Sleeve friction – the frictional force on the sleeve divided by the surface area – expressed in kPa.
- Friction ratio – the ratio of sleeve friction to cone resistance, expressed as a percentage.

The ratios of the sleeve resistance to cone resistance will vary with the type of soil encountered, with higher relative friction in clays than in sands. Friction ratios of 1% to 2% are commonly encountered in sands and occasionally very soft clays, rising to 4% to 10% in stiff clays and peats. Soil descriptions based on cone resistance and friction ratios are only inferred and must not be considered as exact.

Correlations between EFCP and SPT values can be developed for both sands and clays but may be site specific.

Interpretation of EFCP values can be made to empirically derive modulus or compressibility values to allow calculation of foundation settlements.

Stratification can be inferred from the cone and friction traces and from experience and information from nearby boreholes etc. Where shown, this information is presented for general guidance, but must be regarded as interpretive. The test method provides a continuous profile of engineering properties but, where precise information on soil classification is required, direct drilling and sampling may be preferable.

**Portable Dynamic Cone Penetrometers:** Portable Dynamic Cone Penetrometer (DCP) tests are carried out by driving a rod into the ground with a sliding hammer and counting the blows for successive 100mm increments of penetration.

Two relatively similar tests are used:

- Cone penetrometer (commonly known as the Scala Penetrometer) – a 16mm rod with a 20mm diameter cone end is driven with a 9kg hammer dropping 510mm (AS1289, Test F3.2). The test was developed initially for pavement subgrade investigations, and correlations of the test results with California Bearing Ratio have been published by various Road Authorities.
- Perth sand penetrometer – a 16mm diameter flat ended rod is driven with a 9kg hammer, dropping 600mm (AS1289, Test F3.3). This test was developed for testing the density of sands (originating in Perth) and is mainly used in granular soils and filling.

### LOGS

The borehole or test pit logs presented herein are an engineering and/or geological interpretation of the sub-surface conditions, and their reliability will depend to some extent on the frequency of sampling and the method of drilling or excavation. Ideally, continuous undisturbed sampling or core drilling will enable the most reliable assessment, but is not always practicable or possible to justify on economic grounds. In any case, the boreholes or test pits represent only a very small sample of the total subsurface conditions.

The attached explanatory notes define the terms and symbols used in preparation of the logs.

Interpretation of the information shown on the logs, and its application to design and construction, should therefore take into account the spacing of boreholes or test pits, the method of drilling or excavation, the frequency of sampling and testing and the possibility of other than "straight line" variations between the boreholes or test pits. Subsurface conditions between boreholes or test pits may vary significantly from conditions encountered at the borehole or test pit locations.

### GROUNDWATER

Where groundwater levels are measured in boreholes, there are several potential problems:

- Although groundwater may be present, in low permeability soils it may enter the hole slowly or perhaps not at all during the time it is left open.
- A localised perched water table may lead to an erroneous indication of the true water table.
- Water table levels will vary from time to time with seasons or recent weather changes and may not be the same at the time of construction.
- The use of water or mud as a drilling fluid will mask any groundwater inflow. Water has to be blown out of the hole and drilling mud must be washed out of the hole or "reverted" chemically if water observations are to be made.

More reliable measurements can be made by installing standpipes which are read after stabilising at intervals ranging from several days to perhaps weeks



for low permeability soils. Piezometers, sealed in a particular stratum, may be advisable in low permeability soils or where there may be interference from perched water tables or surface water.

#### **FILL**

The presence of fill materials can often be determined only by the inclusion of foreign objects (eg bricks, steel etc) or by distinctly unusual colour, texture or fabric. Identification of the extent of fill materials will also depend on investigation methods and frequency. Where natural soils similar to those at the site are used for fill, it may be difficult with limited testing and sampling to reliably determine the extent of the fill.

The presence of fill materials is usually regarded with caution as the possible variation in density, strength and material type is much greater than with natural soil deposits. Consequently, there is an increased risk of adverse engineering characteristics or behaviour. If the volume and quality of fill is of importance to a project, then frequent test pit excavations are preferable to boreholes.

#### **LABORATORY TESTING**

Laboratory testing is normally carried out in accordance with Australian Standard 1289 "Methods of Testing Soil for Engineering Purposes". Details of the test procedure used are given on the individual report forms.

#### **ENGINEERING REPORTS**

Engineering reports are prepared by qualified personnel and are based on the information obtained and on current engineering standards of interpretation and analysis. Where the report has been prepared for a specific design proposal (eg a three storey building) the information and interpretation may not be relevant if the design proposal is changed (eg to a twenty storey building). If this happens, the company will be pleased to review the report and the sufficiency of the investigation work.

Every care is taken with the report as it relates to interpretation of subsurface conditions, discussion of geotechnical aspects and recommendations or suggestions for design and construction. However, the Company cannot always anticipate or assume responsibility for:

- Unexpected variations in ground conditions – the potential for this will be partially dependent on borehole spacing and sampling frequency as well as investigation technique.
- Changes in policy or interpretation of policy by statutory authorities.
- The actions of persons or contractors responding to commercial pressures.

If these occur, the company will be pleased to assist with investigation or advice to resolve any problems occurring.

#### **SITE ANOMALIES**

In the event that conditions encountered on site during construction appear to vary from those which were expected from the information contained in the report, the company requests that it immediately be notified. Most problems are much more readily resolved when conditions are exposed that at some later stage, well after the event.

#### **REPRODUCTION OF INFORMATION FOR CONTRACTUAL PURPOSES**

Attention is drawn to the document "Guidelines for the Provision of Geotechnical Information in Tender Documents", published by the Institution of Engineers, Australia. Where information obtained from this investigation is provided for tendering purposes, it is recommended that all information, including the written report and discussion, be made available. In circumstances where the discussion or comments section is not relevant to the contractual situation, it may be appropriate to prepare a specially edited document. The company would be pleased to assist in this regard and/or to make additional report copies available for contract purposes at a nominal charge.

Copyright in all documents (such as drawings, borehole or test pit logs, reports and specifications) provided by the Company shall remain the property of Jeffery and Katauskas Pty Ltd. Subject to the payment of all fees due, the Client alone shall have a licence to use the documents provided for the sole purpose of completing the project to which they relate. License to use the documents may be revoked without notice if the Client is in breach of any objection to make a payment to us.

#### **REVIEW OF DESIGN**

Where major civil or structural developments are proposed or where only a limited investigation has been completed or where the geotechnical conditions/constraints are quite complex, it is prudent to have a joint design review which involves a senior geotechnical engineer.

#### **SITE INSPECTION**

The company will always be pleased to provide engineering inspection services for geotechnical aspects of work to which this report is related.

- Requirements could range from:
- i) a site visit to confirm that conditions exposed are no worse than those interpreted, to
  - ii) a visit to assist the contractor or other site personnel in identifying various soil/rock types such as appropriate footing or pier founding depths, or
  - iii) full time engineering presence on site.

# GRAPHIC LOG SYMBOLS FOR SOILS AND ROCKS

SOIL		ROCK		DEFECTS AND INCLUSIONS	
	FILL		CONGLOMERATE		CLAY SEAM
	TOPSOIL		SANDSTONE		SHEARED OR CRUSHED SEAM
	CLAY (CL, CH)		SHALE		BRECCIATED OR SHATTERED SEAM/ZONE
	SILT (ML, MH)		SILTSTONE, MUDSTONE, CLAYSTONE		IRONSTONE GRAVEL
	SAND (SP, SW)		LIMESTONE		ORGANIC MATERIAL
	GRAVEL (GP, GW)		PHYLLITE, SCHIST	<b>OTHER MATERIALS</b>	
	SANDY CLAY (CL, CH)		TUFF		CONCRETE
	SILTY CLAY (CL, CH)		GRANITE, GABBRO		BITUMINOUS CONCRETE, COAL
	CLAYEY SAND (SC)		DOLERITE, DIORITE		COLLUVIUM
	SILTY SAND (SM)		BASALT, ANDESITE		
	GRAVELLY CLAY (CL, CH)		QUARTZITE		
	CLAYEY GRAVEL (GC)				
	SANDY SILT (ML)				
	PEAT AND ORGANIC SOILS				

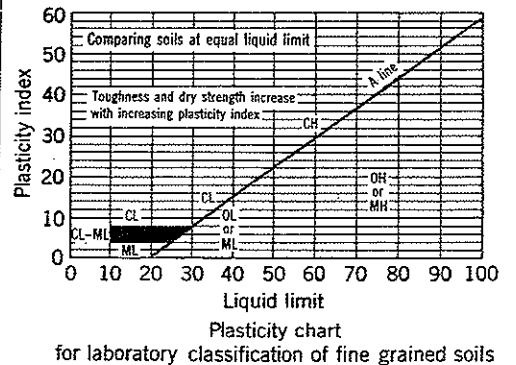


# UNIFIED SOIL CLASSIFICATION TABLE

Field Identification Procedures (Excluding particles larger than 75 µm and basing fractions on estimated weights)				Group Symbols	Typical Names	Information Required for Describing Soils	Laboratory Classification Criteria	
Coarse-grained soils More than half of material is larger than 75 µm sieve size	Gravels More than half of coarse fraction is larger than 4 mm sieve size	Clean gravels (little or no fines)	Wide range in grain size and substantial amounts of all intermediate particle sizes	GW	Well graded gravels, gravel-sand mixtures, little or no fines	Give typical name; indicate approximate percentages of sand and gravel; maximum size; angularity, surface condition, and hardness of the coarse grains; local or geologic name and other pertinent descriptive information; and symbols in parentheses  For undisturbed soils add information on stratification, degree of compactness, cementation, moisture conditions and drainage characteristics  Example: <i>Silty sand, gravelly</i> : about 20% hard, angular gravel particles 12 mm maximum size; rounded and subangular sand grains coarse to fine, about 15% non-plastic fines with low dry strength; well compacted and moist in place; alluvial sand; ( <i>SM</i> )	$C_U = \frac{D_{60}}{D_{10}} \text{ Greater than 4}$ $C_C = \frac{(D_{30})^2}{D_{10} \times D_{60}} \text{ Between 1 and 3}$ Not meeting all gradation requirements for <i>GW</i>	
			Predominantly one size or a range of sizes with some intermediate sizes missing	GP	Poorly graded gravels, gravel-sand mixtures, little or no fines			
		Gravels with fines (appreciable amount of fines)	Nonplastic fines (for identification procedures see <i>ML</i> below)	GM	Silty gravels, poorly graded gravel-sand-silt mixtures			Atterberg limits below "A" line, or <i>PI</i> less than 4  Above "A" line with <i>PI</i> between 4 and 7 are <i>borderline</i> cases requiring use of dual symbols
			Plastic fines (for identification procedures, see <i>CL</i> below)	GC	Clayey gravels, poorly graded gravel-sand-clay mixtures			
	Sands More than half of coarse fraction is smaller than 4 mm sieve size	Clean sands (little or no fines)	Wide range in grain sizes and substantial amounts of all intermediate particle sizes	SW	Well graded sands, gravelly sands, little or no fines		$C_U = \frac{D_{60}}{D_{10}} \text{ Greater than 6}$ $C_C = \frac{(D_{30})^2}{D_{10} \times D_{60}} \text{ Between 1 and 3}$ Not meeting all gradation requirements for <i>SW</i>	
			Predominantly one size or a range of sizes with some intermediate sizes missing	SP	Poorly graded sands, gravelly sands, little or no fines			
		Sands with fines (appreciable amount of fines)	Nonplastic fines (for identification procedures, see <i>ML</i> below)	SM	Silty sands, poorly graded sand-silt mixtures			Atterberg limits below "A" line or <i>PI</i> less than 5  Above "A" line with <i>PI</i> between 4 and 7 are <i>borderline</i> cases requiring use of dual symbols
			Plastic fines (for identification procedures, see <i>CL</i> below)	SC	Clayey sands, poorly graded sand-clay mixtures			
Fine-grained soils More than half of material is smaller than 75 µm sieve size (The 75 µm sieve size is about the smallest particle visible to naked eye)	Identification Procedures on Fraction Smaller than 380 µm Sieve Size							
	Silt and clays liquid limit less than 50	Dry Strength (crushing characteristics)	Dilatancy (reaction to shaking)	Toughness (consistency near plastic limit)	<i>ML</i> Inorganic silts and very fine sands, rock flour, silty or clayey fine sands with slight plasticity	Give typical name; indicate degree and character of plasticity, amount and maximum size of coarse grains; colour in wet condition, odour if any, local or geologic name, and other pertinent descriptive information, and symbol in parentheses		
		None to slight	Quick to slow	None				
		Medium to high	None to very slow	Medium				
		Slight to medium	Slow	Slight				
	Silt and clays liquid limit greater than 50	Slight to medium	Slow to none	Slight to medium	<i>MH</i> Inorganic silts, micaceous or diatomaceous fine sandy or silty soils, elastic silts	For undisturbed soils add information on structure, stratification, consistency in undisturbed and remoulded states, moisture and drainage conditions  Example: <i>Clayey silt, brown</i> : slightly plastic; small percentage of fine sand; numerous vertical root holes; firm and dry in place; loess; ( <i>ML</i> )		
		High to very high	None	High				
		Medium to high	None to very slow	Slight to medium				
		Highly Organic Soils	Readily identified by colour, odour, spongy feel and frequently by fibrous texture				<i>Pt</i>	Peat and other highly organic soils

Determine percentages of gravel and sand from grain size curve  
Depending on percentage of fines (fraction smaller than 75 µm sieve size) coarse grained soils are classified as follows:  
Less than 5% GW, GP, SW, SP  
More than 5% GM, GC, SM, SC  
5% to 12% *Borderline* cases requiring use of dual symbols

Use grain size curve in identifying the fractions as given under field identification



NOTE: 1) Soils possessing characteristics of two groups are designated by combinations of group symbols (e.g. GW-GC, well graded gravel-sand mixture with clay fines).

2) Soils with liquid limits of the order of 35 to 50 may be visually classified as being of medium plasticity.

# Jeffery and Katauskas Pty Ltd

CONSULTING GEOTECHNICAL AND ENVIRONMENTAL ENGINEERS  
A.B.N. 17 003 550 801 A.C.N. 003 550 801



## LOG SYMBOLS

### ROCK MATERIAL WEATHERING CLASSIFICATION

TERM	SYMBOL	DEFINITION
Residual Soil	RS	Soil developed on extremely weathered rock; the mass structure and substance fabric are no longer evident; there is a large change in volume but the soil has not been significantly transported.
Extremely weathered rock	XW	Rock is weathered to such an extent that it has "soil" properties, ie it either disintegrates or can be remoulded, in water.
Distinctly weathered rock	DW	Rock strength usually changed by weathering. The rock may be highly discoloured, usually by ironstaining. Porosity may be increased by leaching, or may be decreased due to deposition of weathering products in pores.
Slightly weathered rock	SW	Rock is slightly discoloured but shows little or no change of strength from fresh rock.
Fresh rock	FR	Rock shows no sign of decomposition or staining.

### ROCK STRENGTH

Rock strength is defined by the Point Load Strength Index ( $I_s$  50) and refers to the strength of the rock substance in the direction normal to the bedding. The test procedure is described by the International Journal of Rock Mechanics, Mining, Science and Geomechanics, Abstract Volume 22, No 2, 1985.

TERM	SYMBOL	$I_s$ (50) MPa	FIELD GUIDE
Extremely Low:	EL	0.03	Easily remoulded by hand to a material with soil properties.
Very Low:	VL	0.1	May be crumbled in the hand. Sandstone is "sugary" and friable.
Low:	L	0.3	A piece of core 150mm long x 50mm dia. may be broken by hand and easily scored with a knife. Sharp edges of core may be friable and break during handling.
Medium Strength:	M	1	A piece of core 150mm long x 50mm dia. can be broken by hand with difficulty. Readily scored with knife.
High:	H	3	A piece of core 150mm long x 50mm dia. core cannot be broken by hand, can be slightly scratched or scored with knife; rock rings under hammer.
Very High:	VH	10	A piece of core 150mm long x 50mm dia. may be broken with hand-held pick after more than one blow. Cannot be scratched with pen knife; rock rings under hammer.
Extremely High:	EH		A piece of core 150mm long x 50mm dia. is very difficult to break with hand-held hammer. Rings when struck with a hammer.

### ABBREVIATIONS USED IN DEFECT DESCRIPTION

ABBREVIATION	DESCRIPTION	NOTES
Be	Bedding Plane Parting	Defect orientations measured relative to the normal to the long core axis (ie relative to horizontal for vertical holes)
CS	Clay Seam	
J	Joint	
P	Planar	
Un	Undulating	
S	Smooth	
R	Rough	
IS	Ironstained	
XWS	Extremely Weathered Seam	
Cr	Crushed Seam	
60t	Thickness of defect in millimetres	

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A.B.N. 17 003 550 801

A.C.N. 003 550 801



## LOG SYMBOLS

LOG COLUMN	SYMBOL	DEFINITION
Groundwater Record		Standing water level. Time delay following completion of drilling may be shown.
		Extent of borehole collapse shortly after drilling.
		Groundwater seepage into borehole or excavation noted during drilling or excavation.
Samples	ES	Soil sample taken over depth indicated, for environmental analysis.
	U50	Undisturbed 50mm diameter tube sample taken over depth indicated.
	DB	Bulk disturbed sample taken over depth indicated.
	DS	Small disturbed bag sample taken over depth indicated.
Field Tests	N = 17 4, 7, 10	Standard Penetration Test (SPT) performed between depths indicated by lines. Individual figures show blows per 150mm penetration. 'R' as noted below.
	N <sub>c</sub> = 5 7 3R	Solid Cone Penetration Test (SCPT) performed between depths indicated by lines. Individual figures show blows per 150mm penetration for 60 degree solid cone driven by SPT hammer. 'R' refers to apparent hammer refusal within the corresponding 150mm depth increment.
	VNS = 25	Vane shear reading in kPa of Undrained Shear Strength.
	PID = 100	Photoionisation detector reading in ppm (Soil sample headspace test).
Moisture Condition (Cohesive Soils)  (Cohesionless Soils)	MC > PL	Moisture content estimated to be greater than plastic limit.
	MC ≈ PL	Moisture content estimated to be approximately equal to plastic limit.
	MC < PL	Moisture content estimated to be less than plastic limit.
	D	DRY - runs freely through fingers.
	M	MOIST - does not run freely but no free water visible on soil surface.
	W	WET - free water visible on soil surface.
	( )	Bracketed symbol indicates estimated consistency based on tactile examination or other tests.
Strength (Consistency) Cohesive Soils	VS	VERY SOFT - Unconfined compressive strength less than 25kPa
	S	SOFT - Unconfined compressive strength 25-50kPa
	F	FIRM - Unconfined compressive strength 50-100kPa
	St	STIFF - Unconfined compressive strength 100-200kPa
	VSt	VERY STIFF - Unconfined compressive strength 200-400kPa
	H	HARD - Unconfined compressive strength greater than 400kPa
	( )	Bracketed symbol indicates estimated consistency based on tactile examination or other tests.
Density Index/ Relative Density (Cohesionless Soils)	VL	Very Loose < 15
	L	Loose 15-35
	MD	Medium Dense 35-65
	D	Dense 65-85
	VD	Very Dense > 85
	( )	Bracketed symbol indicates estimated density based on ease of drilling or other tests.
Hand Penetrometer Readings	300	Numbers indicate individual test results in kPa on representative undisturbed material unless noted otherwise.
	250	
Remarks	'V' bit	Hardened steel 'V' shaped bit.
	'TC' bit	Tungsten carbide wing bit.
	T 60	Penetration of auger string in mm under static load of rig applied by drill head hydraulics without rotation of augers.