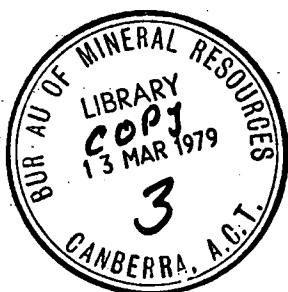


**DEPARTMENT OF
NATIONAL RESOURCES**



**BUREAU OF MINERAL RESOURCES,
GEOLOGY AND GEOPHYSICS**



Record 1978/99

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**GEOLOGICAL INVESTIGATIONS FOR THE SULLIVANS
CREEK SEWER TUNNEL, CANBERRA CITY, ACT 1978**

by

P.H. Vanden Broek, D. Ramsay, & G. Sparksman

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CONTENTS

	<u>Page</u>
SUMMARY OF CONCLUSIONS	
INTRODUCTION	1
ROCK MASS SUBSTANCE	1
Rock types	1
Weathering	2
Rock strength and hardness	3
ROCK MASS DEFECTS	3
Bedding	4
Joints	4
Faults	4
GROUNDWATER	5
Water-levels	5
Permeability and recharge	5
SEISMIC REFRACTION TRAVERSES	6
VIBRATION TESTING	7
TUNNELLING CONDITIONS	8
Rock mass quality	9
Roof support pressure	9
Support requirements	9
Estimation of standup times	11
Excavation method and potential overbreak	16
RECOMMENDATIONS	16
REFERENCES	17
APPENDICES	
1. Definition of terms	18
2. Geological logs of diamond drillholes	21
3. Drillcore photographs	22

TABLES

1. General relationship between degree of weathering and colour for calcareous mudstone
2. Correlation between seismic velocity and degree of weathering of mudstone and siltstone of City Hill Shale
3. Vibration recordings
4. Support requirements
5. Estimated rock mass quality, roof support pressures, and support requirements at tunnel level
6. Estimation of standup times at tunnel level

FIGURES

1. Location map
2. Point load strength data chart
3. Contoured stereographic projection of poles to bedding for City Hill Shale
- 4a. Water-level measurements, bores 1-6
- b. Water-level measurements, bores 6A-8A
- c. Water-level measurements, bores 9-14

PLATES

1. Geological plan and section. Chainage 0+17 -6+05 (I55-A16-2213)
2. " " " " " 6+05 -14+05 (I55-A16-2214)
3. " " " " " 14+05 -21+62 (I55-A16-2215)

SUMMARY OF CONCLUSIONS

- (1) The present tunnel alignment appears to be the best available geologically considering the constraints imposed by existing easements; however, an alternative route could be considered for the section north of manhole 11.
- (2) The rock to be excavated along the length of tunnel section is expected to include about 90% mudstone and 10% tuff and/or sandstone. About 60% of the mudstone is hard and slightly weathered to fresh. About 40% of the mudstone is soft and moderately to extremely weathered. Limestone has not been found along the alignment and is not likely to be present, but it cannot be ruled out from such a sedimentary sequence.
- (3) At least three sets of planar defects can be expected in most sections. As many as five sets may be common in some sections where the rock mass is very closely jointed or fragmented. Variably spaced bedding planes are generally expected to dip east-northeast at 25° .
- (4) The water-table is generally only slightly above tunnel crown level and only minor inflows are expected. Rock permeabilities are generally expected to be low. Only near drillhole 1 is inflow from a surface water recharge source likely; this could result in significant water inflows. Large inflows from limestone are not likely to be encountered.
- (5) Vibration test results indicate maximum charge sizes of 3.5 kg per delay close to the Lakeside Hotel, 1.5 kg per delay near Tasman House, and 0.25 kg per delay beneath the Child Care Centre. Special tunnelling methods may have to be considered near the Child Care Centre.

- (6) About 1400 m of tunnel section is expected to have rock mass quality ranging from poor to exceptionally poor, 600 m is expected to range from poor to fair and about 200 m is expected to be good.
- (7) Standup times calculated from drill cores range from less than 1 hour at drillhole 1 to 1 month at drillhole 8.
- (8) The amount of overbreak above crown will depend primarily on (a) the type of excavation methods used; (b) the speed at which support can be installed; (c) steps taken to improve the rock mass before excavation, and (d) the presence and amount of ground-water inflows.
- (9) It should be practicable to use a road-header type tunnel excavation machine from chainage 0+00 to 7+00. It may be possible to excavate other sections of the tunnel with such a machine, but harder rock between these sections may be beyond the machine's capabilities.

INTRODUCTION

This report describes the geological investigations carried out for the proposed sewer pipeline from Sullivans Creek to Commonwealth Avenue pumping station (Fig. 1). The consultant firm of Camp, Scott & Furphy Engineers Pty Ltd is designing the works for the National Capital Development Commission (NCDC). This report supplements a (previous) BMR feasibility report by Henderson (1978) which included information derived from inspections of nearby excavations by BMR geologists and from reports on drilling for foundation investigations of building sites. Subsequent data have been obtained from 21 diamond drillholes, three seismic refraction traverses along or close to the two alternative tunnel alignments, and vibration tests.

Information obtained during the investigation indicated that subsurface conditions were better for the alignment east of Commonwealth Avenue; however, information applicable to the alignment west of Commonwealth Avenue has also been included in this report.

All information obtained has been summarised on the tunnel section shown in Plates 1, 2, and 3.

ROCK MASS SUBSTANCE

ROCK TYPES

(a) City Hill Shale

The City Hill Shale is a sequence of sedimentary rocks comprising mainly calcareous mudstone, with some shale, siltstone, and fine-grained sandstone towards the top of the sequence. The possibility of the tunnel excavation intersecting a lens of limestone in such sediments cannot be ruled out, although limestone was not intersected in the drillholes or mapped in surface or subsurface exposures.

(b) Volcanic tuff member

Tuff was not intersected by the diamond drilling but has been previously mentioned in the feasibility report (Henderson, 1978). Interbeds of tuff are to be expected towards the top of the City Hill Shale, and could be in excess of 20 m thick at tunnel level.

WEATHERING

A definition of weathering categories has been included in Appendix 1.

Calcareous mudstone is particularly susceptible to leaching out of carbonate minerals, and mantles of altered rock (13 m thick in places) are preserved in areas that have not been greatly eroded or that have been downfaulted. The colour of the mudstone has been found to provide an indication of the degree of weathering (see Table 1).

TABLE 1. GENERAL CORRELATION BETWEEN DEGREE OF WEATHERING AND COLOUR
FOR CALCAREOUS MUDSTONE

<u>Degree of weathering</u>	<u>Colour</u>
Extremely to highly weathered	Pale yellow to yellow brown, sometimes red
Moderately weathered	Yellow-brown to brown with some red- brown
Slightly weathered	Olivine to olive-brown
Fresh stained	Pale grey
Fresh	Blue-grey

The sandstone-shale-siltstone portion of the sequence has not been observed in the fresh state in the drillcores, and is expected to be moderately weathered at tunnel level.

Extremely weathered coarse-grained volcanic tuff was mapped in two excavations close to the Lakeside Hotel. Experience with similar beds elsewhere indicates that highly weathered tuff might extend to depths of more than 10 m, and interbeds of weathered tuff are expected to have properties similar to those of adjacent rocks.

ROCK STRENGTH AND HARDNESS

Point load strength tests were carried out on drillcore specimens obtained from near tunnel level. Two point load testing machines with different operators were used for testing two sets of similar rock samples.

Individual test results are plotted on the drill logs (Appendix 2) and all results are graphically represented in Figure 2.

Results given are for the point load strength index (Broch & Franklin 1972) standardised for a 50-mm core diameter (Is_{50}) and given in megapascals. Experience with point load strengths indicates that $24 \times Is_{50}$ is quite a reasonable guide to the unconfined compressive strength. (M. Idnurm, BMR, personal communication).

Figure 2 shows that there is some correlation between degree of weathering and point load strength index, although slightly weathered rock exhibits quite a large scatter of values for point load strength.

No specific tests were carried out to determine hardness; however, fresh, fresh stained, and slightly weathered rock is moderately hard, and moderately to extremely weathered rock is soft (Appendix 1).

ROCK MASS DEFECTS

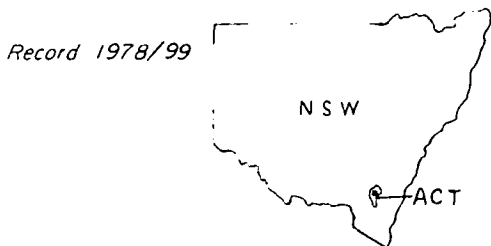
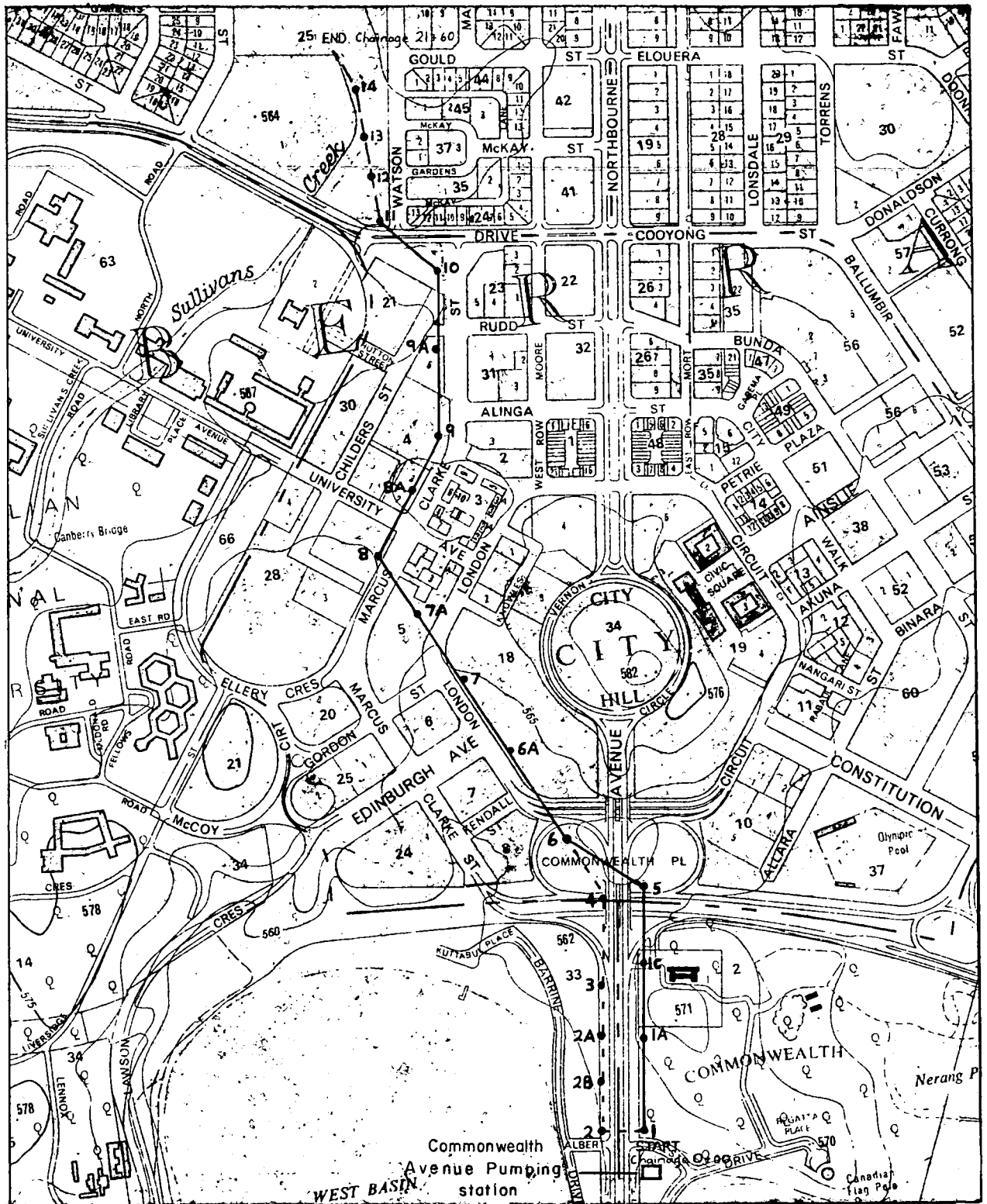
Rock mass defects include bedding, joints, and faults. Their spacing, orientation, roughness, and coatings have been taken into account when rock mass quality, standup times, and support requirements were calculated.

Defect surfaces are mostly clean or ironstained where the rock is relatively unweathered. Where the rock is moderately weathered most surfaces are iron or manganese-coated and some are clay-coated. Defect surfaces are mostly smooth and planar though some rough and planar surfaces were observed in fresh rock.

LOCATION MAP

SULLIVANS CREEK SEWER TUNNEL

Fig 1



SCALE 1:10 000

250 0 500 1000 METRES

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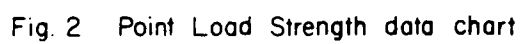


Fig. 2 Point Load Strength data chart

BEDDING

Bedding is generally not visible in drillcore of calcareous mudstone where it is fresh or slightly weathered; bedding in moderately weathered mudstone, shale, siltstone, and sandstone is visible in outcrop, building excavation sites, and trenches, and sometimes in drillcore. Bedding generally dips to the east or northeast at 25° with bedding planes variably spaced. Poles to bedding planes have been plotted and contoured on a stereographic projection (Fig. 3).

JOINTS

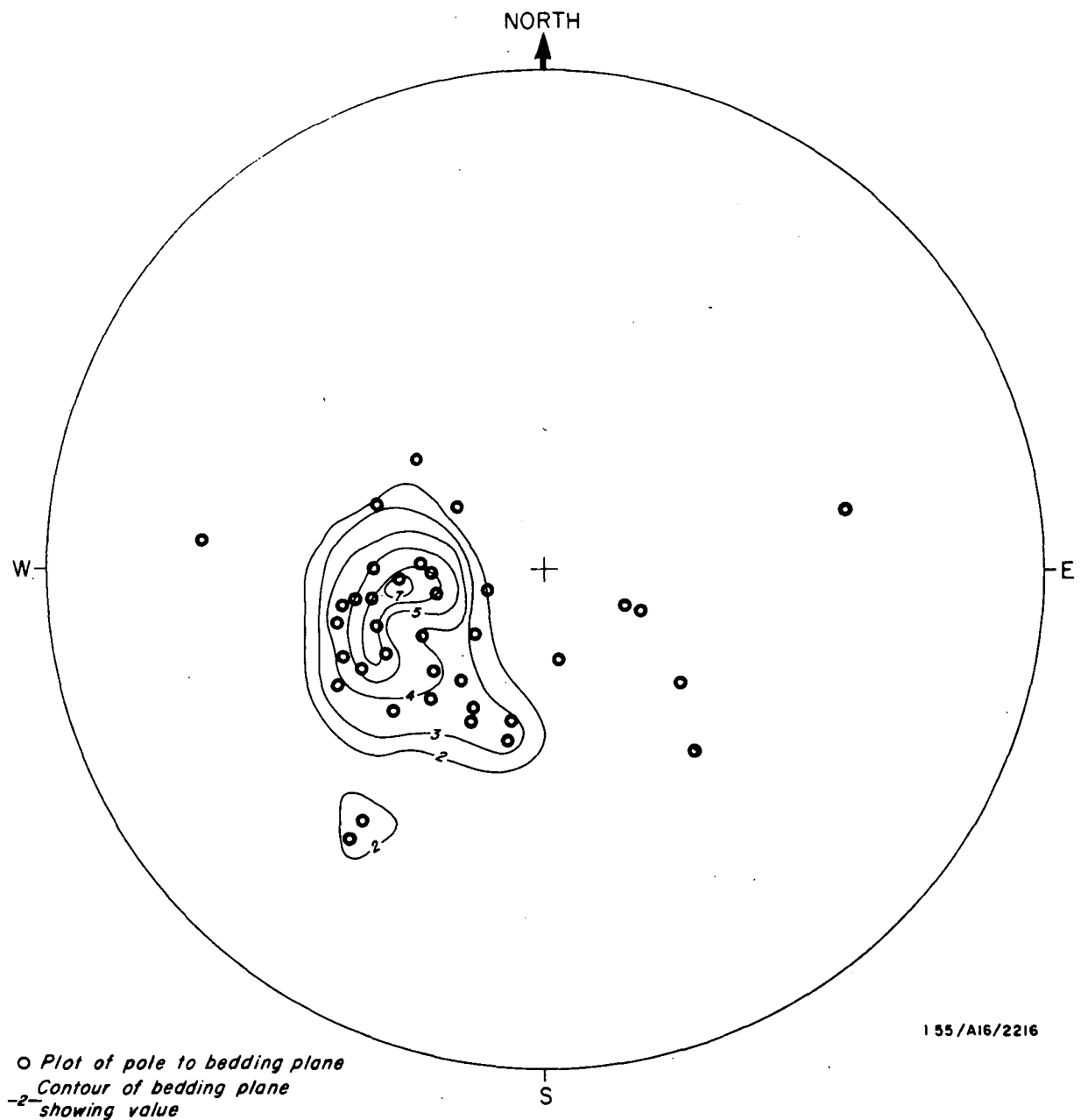
At least three joint sets are expected to occur in the tunnel and four sets may be common for some very closely jointed and fragmented sections. Spacing of the joints determines the rock quality designation (RQD) which is defined as the ratio (expressed as a percentage) of length of core recovered to the total length of core run, counting only those pieces of hard and sound rock 10 cm in length or longer.

Drillcore was not oriented, therefore the effect of joint attitudes on tunnel stability is largely unknown though one set observed in excavations dips at $70^{\circ}/090^{\circ}$ (dip direction).

FAULTS

A number of small faults and sheared and fractured zones are expected to intersect the tunnel excavation. Movement along the faults is thought to have been small (generally less than a few metres). Alteration zones associated with faults are expected to be narrow and should not significantly influence tunnelling conditions.

Small faults could explain variations in the weathered profiles of some drillholes along tunnel section (e.g., holes 8, 8A, 10 and 11).



155/A16/2216

Fig.3 Contoured stereographic projection of poles to bedding for City Hill Shale along tunnel route. Stereograph indicates average dip of beds to be Dip/Dip direction = 25/080

GROUNDWATER

WATER-LEVELS

Water-levels measured in each of the drillholes are plotted in Figures 4a, b, and c. Measurements taken between August to November 1978 show some interesting fluctuations in the groundwater-table.

The water-level in hole 1 remained steady, and was influenced by the normal hydraulic gradient that could be expected away from Lake Burley Griffin.

Holes 1A, 1C, 4, 6, 6A, 7 and 7A had water-levels that were below lake level for part of the time; such observations would not be expected in this groundwater situation. Lowering of the water-table may be due to leakage into the existing buried services, including sewers, or to pumping of water from the basements of nearby buildings.

Water-levels in holes 8 and 10 were abnormally high. This could be attributed either to inflows from perched aquifers or from the inflow or surface water after rainfall. These holes probably do not reflect the groundwater-level in the surrounding area.

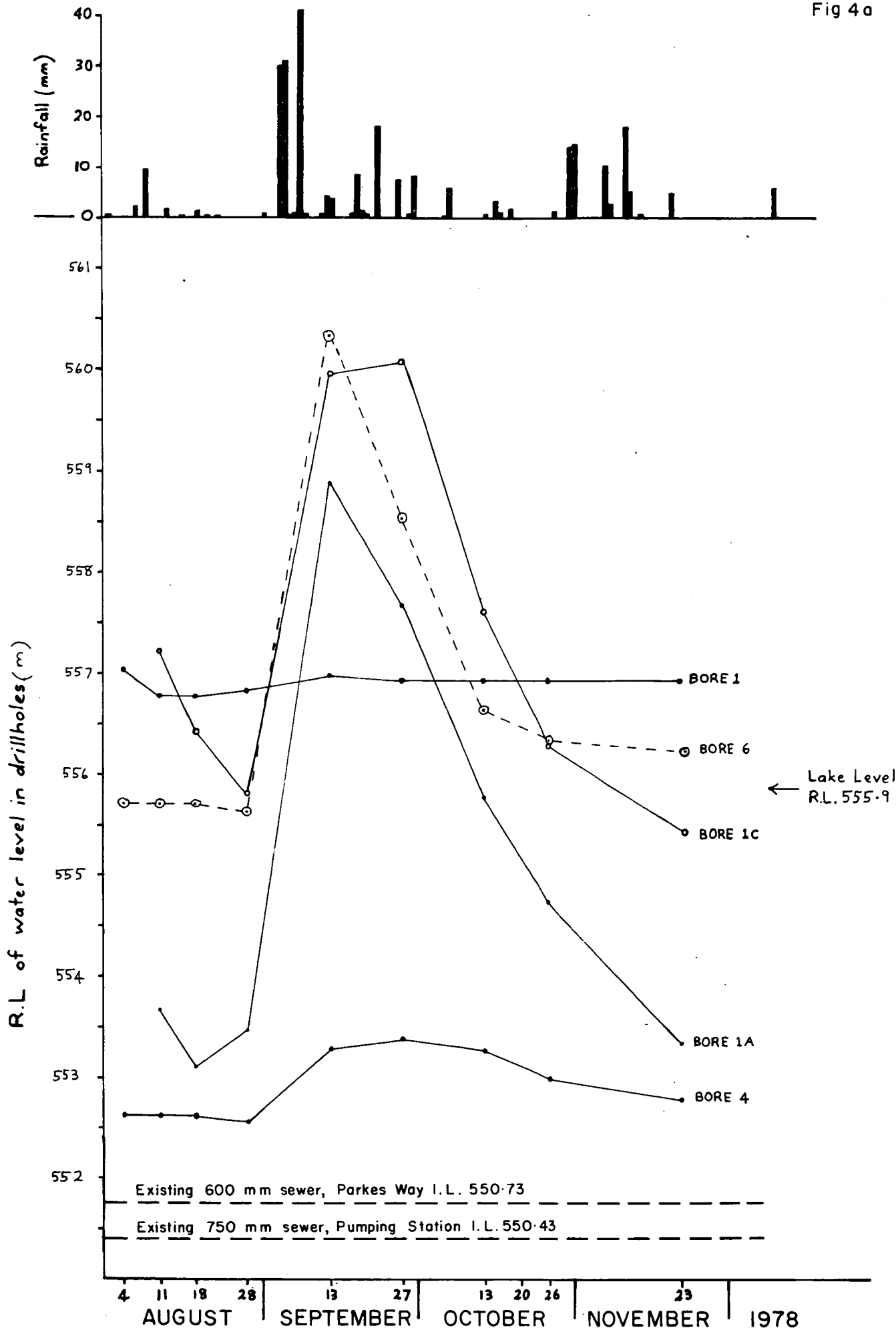
Water-levels in the remaining drillholes provide a reasonable guide to the groundwater conditions in their vicinity.

PERMEABILITY AND RECHARGE

Permeability of the calcareous mudstone is generally expected to be low (25 litres/minute/10 m of tunnel length) along most of the tunnel section, and recharge is likely to depend on rainfall and not surface water features such as Sullivans Creek and Lake Burley Griffin. However, the proximity of Lake Burley Griffin is expected to affect the tunnel near drill-hole 1, where rocks are expected to be more permeable. Inflows in excess of about 125 litres/minute/10 m of tunnel length may occur, and will be maintained if there is a hydraulic connection to the lake.

Initial groundwater flows of 25-125 litres/minute/10 m section can be expected in a number of places as tunnel excavation proceeds. These flows are expected to be short-lived however, and should subside within a matter of days.

Fig 4a



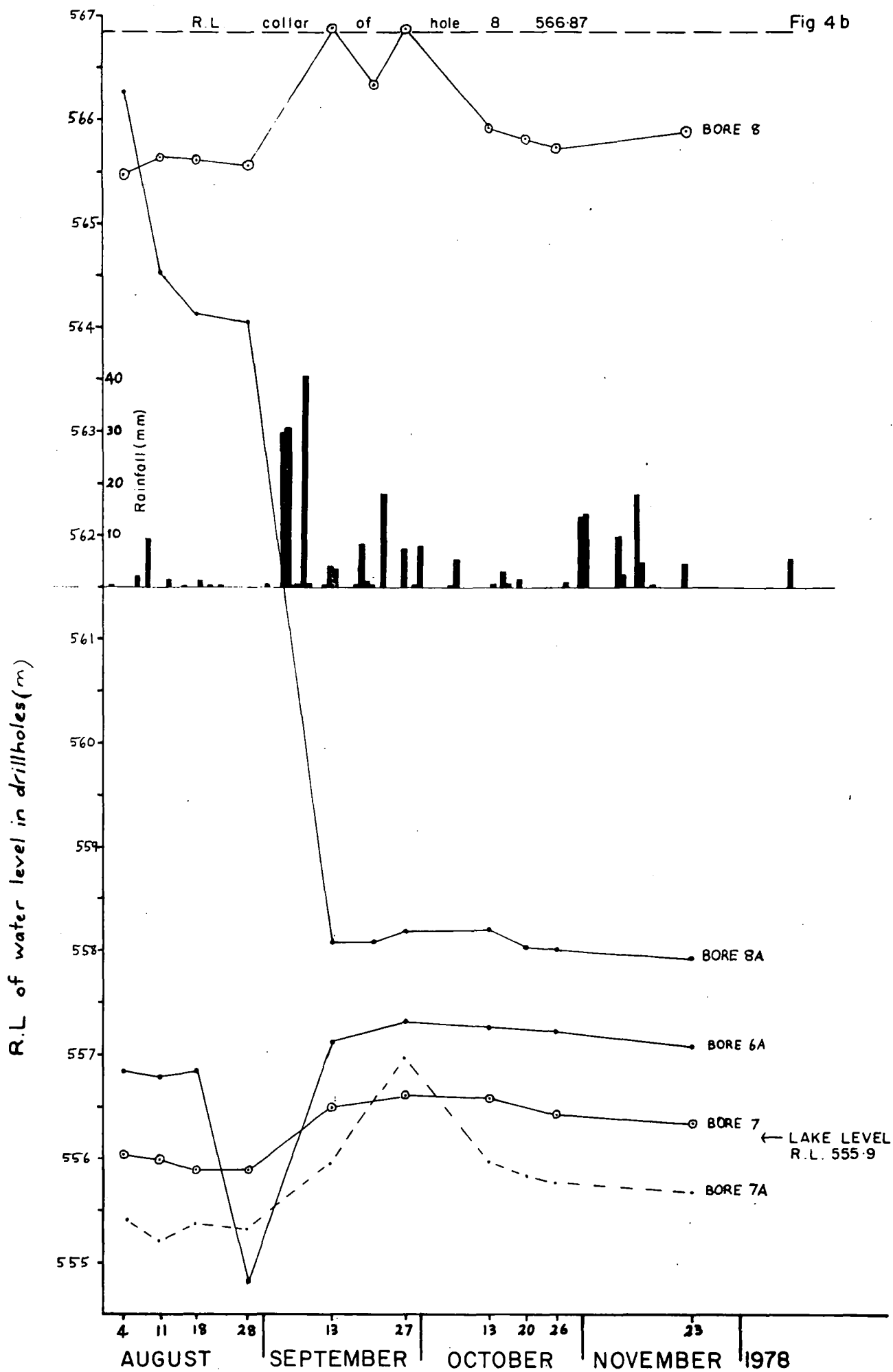
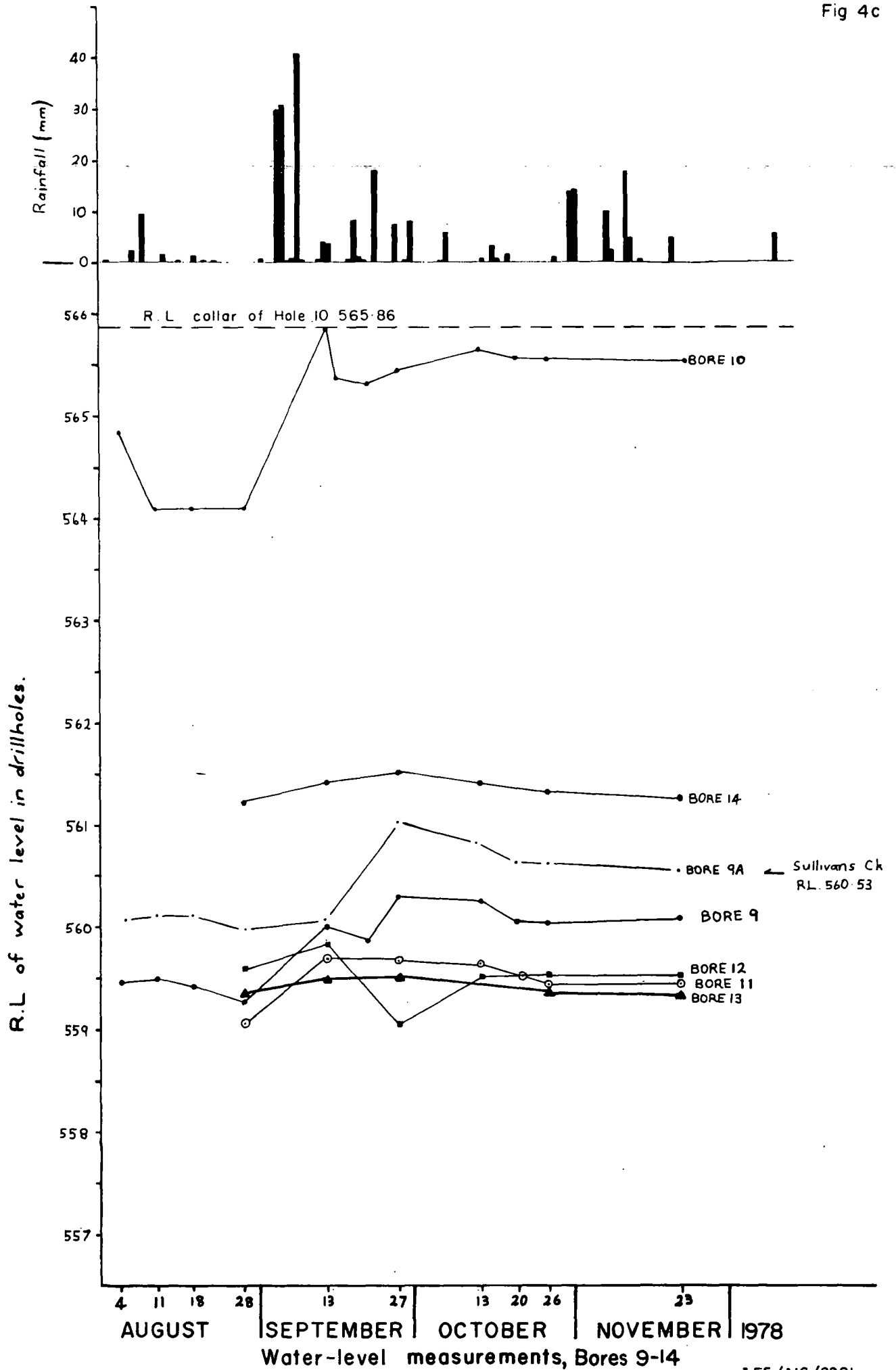


Fig 4c



If limestone is intersected by tunnel excavations (and this is considered unlikely), solution channels could introduce high flows; if such channels were in hydraulic continuity with the lake, such inflows would be hazardous. Limestone in the area lacks continuity and occurs as isolated lenses; if limestone lenses are present along the tunnel line, they would not be expected to have hydraulic continuity with the lake beyond chainage 9+00.

SEISMIC REFRACTION TRAVERSES

To aid prediction of ground conditions for the proposed tunnel, seismic refraction work was done where possible. Because much of the route is along streets in a built-up area, only three short traverses were possible (traversed A, B, C in Plates 1 and 2). The traverses had to be offset from the line of the tunnel because of its proximity to traffic and to underground services.

The seismic records and time-distance curves are available for inspection from the BMR on request.

Based on the measured seismic velocities and the degrees of weathering observed in diamond-drillhole cores, the following general correlation table is presented:

TABLE 2. CORRELATION BETWEEN SEISMIC VELOCITY AND DEGREE OF WEATHERING OF MUDSTONE AND SILTSTONE OF THE CITY HILL SHALE.

<u>Seismic velocity range</u> (m/s)	<u>Expected degree of rock weathering</u>
400 - 500	Soil, fill, and extremely weathered rock
1000 - 1300	Highly weathered rock
1300 - 1900	Moderately weathered rock
2000 - 2900	Moderately to slightly weathered rock
2900	Slightly weathered to fresh rock

VIBRATION TESTING

To provide a guide to charge sizes when blasting the tunnel, holes were drilled on tunnel line and to tunnel depth, and explosive charges detonated in them in order to measure the vibration levels produced at nearby buildings. From these results the maximum charge size permissible, to comply with the Standards Association of Australia (SAA) peak particle velocity limit (19 mm/s), was determined by extrapolation.

Most of the holes had partly collapsed so the charges were detonated at points above tunnel level. In some locations, two separate instruments recorded the same explosion.

The peak particle velocities recorded, along with other details are given in Table 3.

TABLE 3 - VIBRATION RECORDINGS

<u>Depth of charge (m)</u>	<u>Charge size (kg)</u>	<u>Distance from hole to detector (m)</u>	<u>Peak particle velocity (mm/s)</u>	<u>Location of detector</u>
9	0.34	50	5.6	front driveway of Lakeside Hotel
5	0.34	50	4.1	as above
5	1.02	(50	9.8	as above
		(55	7.8	bottom of bottleshop driveway
7.5	0.11	13	13.7	Child Care Centre, Marcus Clarke St
15	0.34	(30	8.6	outside barber's shop, Tasman House (a)
		(35	5.4	outside electricity substation, adjacent to Tasman House (b)
12	1.02	(30	16.2	as above (a)
		(35	7.8	as above (b)

If we assume sinusoidal vibration, then the peak particle velocity (v) is given by the empirical formula:

$$v = \frac{2 \quad f k \quad w}{d}$$

where f is the frequency of the vibration

k is a site constant

w is the weight of explosives per delay

d is the distance from explosion to detector

Thus, for any particular location, f, k, and d are fixed, so v is proportional to w. Using this proportionality we can roughly predict the charge size which would produce a peak particle velocity of 19 mm/s, the limit recommended by the SAA for vibrations resulting from blasting in a built-up area.

From the figures shown in Table 1 the following maximum charge sizes were calculated:

Lakeside Hotel	- 3.5 kg per delay
Tasman House	- 1.5 kg per delay
Child Care Centre	- 0.25 kg per delay.

Note that these figures are derived from an empirical formula, and serve only as a guide.

Vibration testing results indicate a severe limit to the charge size that can be used near the Child Care Centre. Such charge sizes may be impractical for tunnel excavation and special tunnelling methods may have to be considered for this section.

TUNNELLING CONDITIONS

Predicted tunnelling conditions based on a calculated rock mass quality, degree of weathering, standup times, RQD, and RCN are given in Plates 1, 2, and 3. Rock mass quality, roof support pressures, and support requirements at tunnel level have been estimated by using the methods proposed by Barton, Lien, & Lunde (1974), and are set out in Tables 4 and 5.

ROCK MASS QUALITY

Rock mass quality Q is calculated from the formula:

$$Q = \frac{RQD}{J_n} \times \frac{J_r}{J_a} \times \frac{J_w}{SRF}$$

Where: RQD = Rock quality designation
 J_n = Joint set number
 J_r = Joint roughness number
 J_a = Joint alteration number
 J_w = Joint water reduction factor
 SRF = Stress reduction factor

These parameters were calculated at each drillhole for the zone of influence relevant to tunnel construction - namely the 2 m above and 1 m below crown where drill holes were on tunnel line, and for the total length of the hole above spring line where the drillholes were off tunnel line.

ROOF SUPPORT PRESSURE (P_{roof})

Roof support pressure is calculated from the formula

$$P_{\text{roof}} = \frac{2.0}{J_r} Q^{\frac{1}{3}}$$

Where P_{roof} = permanent roof support pressure in kilograms per square centimetre. In practice P_{roof} is obtained graphically from known values of Q and J_r. Where the value of P_{roof} obtained in this way exceeded $\frac{1}{2} d$ (density x depth) the approximate value of $\frac{1}{2} d$ is given.

SUPPORT REQUIREMENTS

Estimates for tunnel support requirements are obtained by plotting the rock mass quality Q versus the equivalent dimension (Barton & others 1974). The equivalent dimension is defined for this project as follows:

$$\text{Equivalent dimension} = \frac{\text{Diameter (m)}}{ESR}$$

Where ESR (excavation support ratio) depends essentially on the purpose for which the excavation is required. For this tunnel the value of ESR has been taken as 1.6 - 1.25 and the diameter as 2.5 m

$$\text{Equivalent dimension} = \frac{2.5}{1.6} - \frac{2.5}{1.25} = 1.56 - 2.0$$

Up to five support categories are expected in excavating the tunnel; these are set out below in Table 4.

TABLE 4. SUPPORT REQUIREMENTS

<u>Support category</u>	<u>Description of rock mass quality (Q)</u>	<u>Description of support requirement</u>
0	Ranges from Poor - good	No support required
25	Very poor	Systematic bolting (untensioned, grouted) 1 m + reinforced mesh or chain mesh, or small groups of steel sets at 1.5 m + light timber lagging
29	Very poor	Systematic bolting (untensioned grouted) 1 m + shotcrete 3-5 cm + reinforced mesh, or steel sets at 1.5 m + light-medium timber lagging
33	Extremely poor	Systematic bolting (grouted, post-tensioned) 1 m + shotcrete (mesh reinforced) 5-10 cm, or steel sets at 1.0-1.5 m + medium timber lagging.
36 or 37	Exceptionally poor	Shotcrete 10-50 cm (mesh reinforced) + systematic bolting (grouted, post-tensioned) 0.5-1.0 m, or steel sets at heavy to complete timber lagging.

Support given in terms of rock bolts, mesh, and shotcrete is based on 200 case records (Barton & others, 1974), and is probably the most suitable for machine-excavated sections. Where conventional drilling and blasting is used, steel supports and timber may be more appropriate.

Estimated rock mass quality, roof support pressure and support requirements at tunnel level have been calculated for each drillhole relevant to the present tunnel alignment; they are set out in Table 5.

ESTIMATION OF STANDUP TIMES

Standup times for the rock mass intersected at each drillhole on or close to tunnel alignment have been estimated using the rating method developed by Bieniawski (1974). The method takes into account the unconfined strength (or point load strength), RQD, joint spacing, joint orientation, joint condition, and groundwater inflow.

Ratings were applied to the zone of influence relevant to tunnel construction (see ROCK MASS QUALITY) and are set out in Table 6. The standup times given by this method agree fairly well with support calculations, but in a number of instances appear to underestimate the standup times observed for similar rock elsewhere. Standup times range from less than 1 hour in poor rock to about 1 month in fair rock.

TABLE 5. ESTIMATED ROCK MASS QUALITY, ROOF SUPPORT PRESSURES, AND SUPPORT REQUIREMENTS AT TUNNEL LEVEL (after Barton & others, 1974).

Hole No.	RQD %	Jn		RQD/Jn	Jr/Ja	Jw/SRF	Rock mass quality (Q)		P _{roof} (kg/cm ²)	Support category
		No of joint sets	Rating				value	Description		
1	20	2+ random	6	20/6	2.0/1.0	1.0/1.0	0.66/10	0.65 0.44	V. poor V. poor	1.0 25
1A	45	3	9	45/9	1.0/1.0	1.0/5	0.66/5	1.0 0.66	Poor V. poor	0 25
1C	27	3	9	27/9	2.0/1.0		1.0/7.5	0.8	V. poor	1.25 25
4	0	4	15	0/15	1.0/1.0		1.0/5	0	Exceptionally poor	2.0? 37
5	27	3	9	27/9	1.5/2.0	0.66/5.0		0.29	V. poor	2.0 29
6	0	4	15	0/15	1.0/1.0		1.0/2.5	0	Exceptionally poor	2.0? 37
6A	0	4	15	0/15	1.0/1.0		1.0/5.0	0	Exceptionally poor	2.0? 37
7	40	4	15	40/15	1.5/2.0		1.0/2.5	0.8	V. poor	1.5 25
7A	68	3	9	68/9	1.5/1.0		1.0/1.0	11.3	Good	0.5 0
8	43	2 + R	6	93/6	1.5/1.0		0.66/1.0	15.4	Good	0.5 0
8A	39	2 + R	6	39/6	1.5/1.0		0.66/5.0	1.2	Poor	1.0 0

TABLE 5 (Continued)

Hole No.	RQD %	Jn		RQD/Jn	Jr/Ja	Jw/SRF	Rock mass quality (Q)		P _{roof} (kg/cm ²)	Support category
		No of joint sets	Rating				value	Description		
9	32	4	12	32/12	1.0/1.0	1.0/5.0	0.53	V. Poor	2.0	25
9A	3	3	9	3/9	1.0/1.0	1.0/1.0	0.33	V. poor	2.0	29
10	16	4	15	16/15	1.0/1.0	0.8/2.5	0.34	V. poor	2.0	29
11	33	3 + R	12	33/12	1.0/1.0	1.0/5.0	0.55	V. poor	2.0	25
12	64	2	4	64/4	1.0/1.0	1.0/5	3.2	Poor	1.0	0
13	54	4 + R	15	54/15	1.5/1.0	1.0/1.0	5.4	Fair	0.8	0
14	53	3 + R	12	53/12	1.0/3.0	1.0/1.0	1.4	Poor	1.5	0

TABLE 6. ESTIMATION OF STANDUP TIMES AT TUNNEL LEVEL (after Bieniawski, 1974)

Hole	Strength		RQD		Joint ratings			Groundwater		Total	Rock	Standup
no.	u.c.s. (M.Pa)	Rating	%	Rating	Spacing	Orientation	Condition	Inflow/ 10 metres (Litre min)	Rating	rating	class	time
1	< 25	0	20	3	3	5	10	< 25 25-125	8 5	29 26	Poor (IV)	Less than 1 hour
1A	< 25	0	42	9	0	6	10	< 25 25-125	8 5	43 40	Poor (IV)	10 Hours
1C	< 25	0	27	8	10	6	10	< 25	8	42	Poor (IV)	10 Hours
4	< 25	0	0	3	5	3	10	< 25	8	28	Poor (IV)	Less than 1 hour
5	25-50	1	27	8	8	8	7	< 25	8	40	Poor (IV)	8 hours
6	< 25	0	0	3	5	3	10	< 25	8	29	Poor (IV)	About 1 hour
6A	< 25	0	0	3	5	3	10	< 25	8	29	Poor (IV)	About 1 hour
7	< 25	0	40	8	10	6	10	< 25	8	42	Poor (IV)	10 Hours
7A	50-100	2	68	14	15	6	10	< 25	8	55	Fair (III)	About 1 week
8	50-100	2	93	20	15	6	12	about 25	6	61	Fair	About 1 month
8A	< 25	0	39	8	10 5	6	10	25 -125	5	39 34	Poor	10 Hours 2 Hours

TABLE 6 (continued)

Hole	Strength		RQD		Joint ratings			Groundwater		Total	Rock	Standup
no.	u.c.s. (M.Pa)	Rating	%	Rating	Spacing	Orientation	Condition	Inflow/ 10 metres (Litre min)	Rating	rating	class	time
9	< 25	0	32	8	10	6	10	< 25	8	42	Poor	10 Hours
9A	< 25	0	3	3	5	6	10	< 25	8	32	Poor	2 Hours
10	< 25	0	16	3	10	6	10	25-125		5	34	Poor
11	50-100	2	33	8	10	6	10	< 25	8	44	Poor	10
12	50-100	2	64	14	10	6	10	< 25	8	54	Fair	1 Week
13	100-200	5	54	14	10	6	10	< 25	8	53	Fair	2 days
14	100-200	5	53	14	10	6	6	< 25	8	49	Poor	2 days

EXCAVATION METHOD AND POTENTIAL OVERBREAK

The section of tunnel from chainage 0+00 to 7+30 is capable of being excavated by a road-header type tunnel-excavating machine. The remainder of the tunnel is likely to require conventional drilling and blasting, although a small section from chainage 14+80 to 16+90 could probably be excavated by a tunnel excavation machine.

Should drilling and blasting methods be used for the entire tunnel excavation then large overbreak sections can be expected, particularly where the rock mass quality is extremely to exceptionally poor (see Plates 1, 2, and 3).

If a tunnelling machine is used for some sections of tunnel, then overbreak will be reduced, and steel mesh, rock bolts, and/or shotcrete could be installed where required. Steel supports and timber lagging may be more expedient where drill and blast methods of excavating are used.

Excavation of the tunnel section near drillhole 1 may intersect groundwater inflows larger than those found elsewhere along the tunnel. Excavation of this section after the remainder of the tunnel has depressed the potentiometric surface in adjacent areas could significantly reduce the quantity of water to be pumped.

A tunnelling option is available between manholes 11 and 15 via manholes 12, 13, and 14, and the expected geological conditions are shown in Plate 3. Alternatively a direct tunnelling option is available for the section north of manhole 11, thereby eliminating manholes 12, 13, and 14. The direct tunnel option would diverge by up to 40 m from the proposed alignment; however, from the limited information available, the rock intersected is likely to be similar to that along the proposed alignment, and tunnelling conditions would not be expected to diverge greatly from those of the proposed alignment.

RECOMMENDATIONS

- (1) Piezometers should be monitored prior to and during construction.
- (2) A road-header type tunnel-excavation machine could be considered for much of the tunnel in conjunction with steel mesh and rock bolt support system (shotcrete where necessary).

- (3) A geologist should log all excavations and advise on geological aspects of tunnel excavation and support during construction.
- (4) A 1 m-diameter drillhole should be drilled for inspection by tenderers. It should show the range of rock conditions that are likely to be intersected during tunnel construction and should also show the worst conditions that are likely to be intersected. Such a hole located at drill hole 6A would meet these requirements.
- (5) Vibration measurements should be recorded during tunnel blasting to ensure that the blasting is carried out within the vibration limits specified by the SAA.

REFERENCES

BUREAU OF MINERAL RESOURCES: Project file 1978/101 (Restricted).

BARTON, N., LIEN, R., & LUNDE, J., 1974 - Engineering classification of rock masses for the design of tunnel support. Rock Mechanics, 6(4), 189-236.

BIENIAWSKI, Z.T., 1974 - Geomechanics classification of rock masses and its application in tunnelling. Advances in Rock Mechanics; Proceedings of 3rd ISRM Congress, 2A.

BROCH, E., & FRANKLIN, J.A., 1972 - The point load strength test. International Journal Rock Mechanics & Mining Science, 9 669-97.

HENDERSON, G.A.M., 1978 - Geological notes on proposed Sullivans Creek to Commonwealth Avenue pumping station sewer augmentation and possible extension to King Edward Terrace, Canberra, A.C.T. Bureau of Mineral Resources, Australia, Record 1978/24 (unpublished).

APPENDIX 1
DEFINITION OF TERMS

1. WEATHERING OF ROCK

FRESH	: No discolouration or loss in strength
FRESH STAINED	: Limonitic staining along fractures, rock otherwise fresh and shows no loss of strength.
SLIGHTLY WEATHERED	: Rock is slightly discoloured, but not noticeable lower in strength than the fresh rock.
MODERATELY WEATHERED	: Rock is discoloured and noticeable weakened; N - size drill core generally cannot be broken by hand across the rock fabric.
HIGHLY WEATHERED	: Rock is discoloured and weakened; N-size drill core can generally be broken by hand across the rock fabric.
EXTREMELY WEATHERED	: Rock is decomposed to a soil, but the original rock fabric is mostly preserved.

2. PERCUSSIVE STRENGTH OF ROCK

STRONG TO VERY STRONG	: Cannot be broken by repeated blows with a hammer.
MODERATELY STRONG	: Rock broken by 3 or 4 blows.
WEAK	: Rock broken by one blow.

3. HARDNESS OF ROCK

HARD TO VERY HARD	: Impossible to scratch with knife blade.
MODERATELY HARD	: Shallow scratches with knife blade.
SOFT	: Deep scratches with knife blade.

4. DEFECT SPACING

WIDE	: 300-100 cm
MODERATELY WIDE	: 100-30 cm
CLOSE	: 10-3 cm
FRAGMENTED	: < 3 cm

5. BEDDING

LAMINATED	: 10 mm thick
THINLY BEDDED	: 10-100 mm thick
THICKLY BEDDED	: 100 mm thick

6. ROCK CONDITION NUMBER

- | | |
|---|---|
| 1 | <u>Hard and intact:</u> Rock very hard and strong with no significant joints or other defects including bedding plane partings. No support required. |
| 2 | <u>Hard, widely jointed rock:</u> As above but bedding plane partings and joints spaced 1-3 m and tight. Joints rarely continuous for more than a few metres. No support necessary. |
| 3 | <u>Moderately jointed or bedded:</u> Rock generally hard and strong, with continuous joints or bedding plane partings spaced 0.5-1.0 m and usually fairly tight. Minor water seepages likely. Rock may be a little blocky in places and may need rock bolt support where defects are unfavourably oriented to tunnel alignment. Shotcrete may be effective. |
| 4 | <u>Moderately jointed and seamy:</u> As above but defect surfaces generally clay coated and loose. Rock may be criss-crossed with seams or shears and may be moderately weathered. Support generally required (1-m spaced sets) especially where opening is greater than 3 m in diameter. Shotcrete may be effective in places. |

5

Closely jointed and seamy: Closely jointed, seamy, and fractured rock. Rock may be highly or extremely weathered. Where defect surfaces are open and clean water inflows may occur. Steel sets required with heavy lagging. Shotcrete may be effective if applied before ground commences movement.




APPENDIX 2

GEOLOGICAL LOGS OF DIAMOND-DRILLHOLES

HOLE No. 1
Canberra 247


ANGLE FROM HORIZONTAL (θ) 90° DIRECTION
CO-ORDINATES 602 970 N. 210 682 E. R L OF 560.23m

SHEET 1 OF 2

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography Negative No.
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained	 10 Oct 73 water level date shown Water Inflow	Depth (m) B & W Colour
Core barrel type <u>Trefus</u>	SW - Slightly weathered		<u>1.5 - 6.1</u> <u>M 23 24 L</u>
<u>NMLC</u>	MW - Moderately weathered		<u>6.1 - 11.0</u> -----
Driller <u>V.O. Brisco</u>	HW - Highly weathered	 Partial drilling water loss	-----
Commenced <u>25.7.78</u>	EW - Extremely weathered	 Complete drilling water loss	-----
Completed <u>26.7.78</u>	<u>Notes</u>		-----
Logged by <u>P. Yeoden Brook</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		-----
Vertical scale <u>1:100</u>	* Water Pressure Tests		-----

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION
CO-ORDINATES 603.050N 210.688E R L OF 561.66m

SHEET 1 OF 1

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography Negative No
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained	 10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	Water inflow	1.5 - 6.3 M 23241
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	6.3 - 11.0
Driller <u>V. O'Brien</u>	HW - Highly weathered	Complete drilling water loss	
Commenced <u>1/8/78</u>	EW - Extremely weathered		
Completed <u>1/8/78</u>	<u>Notes</u>		
Logged by <u>P. Vanden Brack</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

PROJECT Gould St. - Commonwealth Ave Sewer Tunnel
LOCATION East side of Commonwealth Ave. - - - - -

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION ---
CO-ORDINATES 603 216 N 210 709 E R L OF 563.98m




SHEET 1 OF 1

Drilling information						Rock Substance		Rock Mass Defects					Rock condition No (interpretative)
Method	Drilling rate	Casing	Water	Pressure test *	Lift & % core recovery	Depth (metres)	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	Point load 1.0 strength 5.0 strength 10.0 strength	Defect spacing (cm)	R.O.D.	Defect description thickness, type, inclination, planarity, roughness, coating strength	
Auger							No core						
						100	Clay 1:0	HW					Fragmented with some clay seams.
						200	Pale yellow-brown to pale red-brown mudstone.	MW			50		Closely jointed, mostly stained with Manganese. Some thin clay seams.
						280	Pale yellow mudstone. Clay 0:1	HW			20		seam 2-3 cm wide
						380	Pale yellow-brown to red brown mudstone.				0		
						480					35		Closely jointed with Manganese staining on joints
						580					50		
						680	Fragmented rock	MW			35		Mostly closely jointed with manganese staining on moderately dipping joints
						780	Fragmented rock				30		
						880					45		
						980					47		
						1000	End of Hole R.L. = 552.98						

Drill type <u>Pienger</u>	<u>Weathering</u>	<u>Water</u>	Core Photography Negative No
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained	10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Tiefus</u>	SW - Slightly weathered	Water Inflow	1.05 - 6 m 23241
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	6.0 - 11.0
Driller <u>V. O'Brien</u>	HW - Highly weathered	Complete drilling water loss	
Commenced <u>2/8/78</u>	EW - Extremely weathered		
Completed <u>2/8/78</u>	<u>Notes</u>		
Logged by <u>P. VandenBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION ---
CO-ORDINATES 603 975 N 210 615 E R.L. OF 560.11 m

SHEET 1 OF 1

Drill type <u>Pioneer</u>	<u>Weathering</u>	Water	Core Photography Negative No
Feed <u>Hydraulic</u>	Fr - Fresh	 10 Oct 73 water level date shown Water inflow	Depth (m) B & W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered		<u>1.5 - 6.65</u> <u>M23241</u>
<u>NMLS</u>	MW - Moderately weathered	 Partial drilling water loss	<u>6.65 - 11.0</u>
Driller <u>Vin O'Brien</u>	HW - Highly weathered	 Complete drilling water loss	
Commenced <u>24.7.78</u>	EW - Extremely weathered		
Completed <u>24.7.78</u>	<u>Notes</u>		
Logged by <u>P. VandenBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL 90° DIRECTION
CO-ORDINATES 603 078N 210 623E R.L. OF 560.2SHEET 1 OF 1

Drilling Information					Rock Substance		Rock Mass Defects			
Method	Drilling rate	Casing	Water	Pressure test	Lift & % core recovery	Depth (metres)	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	Defect spacing	Defect description thickness, type, inclination, planarity, roughness, coating, strength
Auger Diamond core						1	Not cored.			
						2	Gravelly clay loam	GC		Fill
						3	Orange sand clay	CI		
						4	Pale yellow-brown MUDSTONE			50 - clay seam and very closely fractured.
						5	Pale orange-brown MUDSTONE			clay seam
						6				30 - very closely fractured to fragmented.
						7				40 - very closely fractured. fragmented.
						8	Pale yellow grey MUDSTONE			55 - Closely jointed with black Manganese staining on horizontal, moderately and steeply dipping joint surfaces.
						9				55 - clay shear
						10				
						11	End of hole: R.L. = 549			

Drill type Pioneer
Feed Hydraulic
Core barrel type Triefus
NMLC
Driller Via O'Brian
Commenced 31/7/78
Completed 31/7/78
Logged by P. Vanden Broek
Vertical scale 1:100

Weathering
Fr - Fresh
SW - Slightly weathered
MW - Moderately weathered
HW - Highly weathered
EW - Extremely weathered

Notes
Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis
* Water Pressure Tests

Water
10 Oct 73 water level date shown
Water inflow
Partial drilling water loss
Complete drilling water loss

Core Photography Negative No
Depth (m) B & W Colour
2.15-7 M 3241
7-11.0

HOLE No. 2B
Canberra 252

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION -
CO-ORDINATES 503.038 N 210.617 E R.L. OF 561

SHEET 1 OF 1

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography Negative No.
Feed <u>Hydraulic</u>	Fr - Fresh	▼ 10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Triplex</u>	SW - Slightly weathered	▶ Water inflow	1.5 - 6.65 023241
<u>NMLC</u>	MW - Moderately weathered	▶ Partial drilling water loss	6.65 - 11.0
Driller <u>V. O. Brian</u>	HW - Highly weathered	▶ Complete drilling water loss	
Commenced <u>3:8:78</u>	EW - Extremely weathered		
Completed <u>7:8:78</u>	<u>Notes</u>		
Logged by <u>P. VanderBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION
CO-ORDINATES 603165N 210631E R.L. OF 562.68mSHEET 1 OF 1

Drilling information					Rock Substance		Rock Mass Defects							
Method	Drilling rate	Casing	Water	Pressure test #	Lift & % core recovery	Depth (metres)	Graphic log Score loss	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	30 Point load 100 strength 300 strength 1000 strength	Defect spacing (cm) 10 30 100 300	R O D	Defect description thickness, type, inclination, planarity, roughness, coating strength	Rock condition No. (interpretative)
Auger			4.8.78					Not cored						
Diamond core						95		Pale yellow-brown MUDSTONE	HW				- Core fragmented	
						2			MW			10	Closely jointed with black manganese staining on moderately dipping joints.	
						100		Clay 2.0	EW			25	Possible Shear	
						3		Pale yellow-brown MUDSTONE	MW				- Core fragmented	3-4
						4		Clay 2.0	EW				- Possible shear	
						5		Pale yellow-brown MUDSTONE					- Closely jointed with black manganese staining on joints.	
						6			MW			0	Fragmented.	4
						7							Fragmented.	
						60		Fragmented rock					Very closely jointed	
						90		Core loss					Fragmented, possibly up to 1m thick	
						100		Pale yellow-brown to core loss						
						10		pale yellow MUDSTONE	MW			15	Closely jointed at high angle and Manganese coated.	4
						11						20	Clay coating on high angle joints from 10.5 to 11.5 m.	
						11.50		End of hole R.L = 551.18						

Drill type Pioneer
Feed Hydraulic
Core barrel type Trietus
NMLC
Driller Van O'Brien
Commenced 24/2/78
Completed 21/7/78
Logged by B. Vanden Broek
Vertical scale 1:100

Weathering

Fr - Fresh
SW - Slightly weathered
MW - Moderately weathered
HW - Highly weathered
EW - Extremely weathered

Notes

Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis
Water Pressure Tests

Water

10 Oct 73 water level date shown
Water Inflow
Partial drilling water loss
Complete drilling water loss

Core Photography Negative No.

Depth (m) B & W Colour
0.4-5.65 13 23 24 L
5.65-10.8
10.8-11.5

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION -----
CO-ORDINATES 603 351 N 210 670 E R L OF 560.19m

SHEET 1 OF 1

Drilling information					Rock Substance		Rock Mass Defects							
Method	Drilling rate	Casing	Water	Pressure test *	Lift & % core recovery	Depth (metres)	Graphic log	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	Point load strength 10.0 is (50) MPa	Defect spacing (cm)	R Q D	Defect description thickness, type, inclination, planarity, roughness, coating strength	Rock condition No.
Auger						0		Not cored						
Diamond core						1								
						100		MUDSTONE Pale yellow - grey to yellow - brown	HW				Closely fractured; iron staining on joints.	4
						80		No Core	EW				Core fragmented; black manganese staining on joints. EW clay section from 3.75 to 4.00 m. General absence of clay coatings on joints.	5
						90		MUDSTONE Pale yellow - orange	HW					
						7		No core						
						98		MUDSTONE Pale yellow orange	MW				Closely fractured.	
						100							Core fragmented; black Manganese staining on joints. Clay 10cm wide at 8.9m	
								End of hole R.L. = 550.19						

Drill type <u>PIONEER</u>	Weathering	Water	Core Photography Negative No.
Feed <u>Hydraulic</u>	Fr - Fresh	10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	Water Inflow	1.0 - 6.0 m 23241
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	6.0 - 10.0
Driller <u>V.O. Brian</u>	HW - Highly weathered	Complete drilling water loss	
Commenced <u>29.7.78</u>	EW - Extremely weathered		
Completed <u>29.7.78</u>	Notes		
Logged by <u>P. L. ...</u>	Banding & joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	Water Pressure Tests		

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION
CO-ORDINATES 603412N 210749E R L OF 564-19SHEET 1 OF 1

Drilling information					Rock Substance		Rock Mass Defects						
Method	Drilling rate	Casing	Water	Pressure test *	Lift & % core recovery	Depth (metres)	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	30 Point load 100 strength 500 strength 1000 strength (kPa)	Defect spacing (cm)	R Q D	Defect description thickness, type, inclination, planarity, roughness, coating strength	Rock condition No (interpretive)
Auger						1	Not Cored						
Diamond core						2	35cm core loss	HW-EW					
						3						Most joints clay coated or surfaces highly weathered	
						4					50	closely jointed zone	
						5					30		
						6	Pale yellow-brown MUDSTONE	MW			70	Most joints stained with limonite and/or manganese. Most joints at 30°-45° and 65°-80°. At least 3 sets visible.	
						7	(City Hill shale)				50	Rock moderately strong	
						8					10	10cm wide fractured zone close intersecting joints	3
						9	Bedding possibly at about 20°-25° (dip)				90	High angle joints (65°-80°) are smoother than others and are probably normal to bedding	
						10					50		
						11		HW-MW			0	Closely jointed and fractured. Joint surfaces mostly highly weathered and stained; some clay coated.	4
						12		HW			30		
						13		MW			40		
						14		HW			30	Sheared zone at about 50°. Minor amount of clay	3-4
						15		MW			50		
								End of hole					

Drill type <u>Explor</u>	Weathering	Water	Core Photography Negative No
Feed <u>Hydraulic</u>	Fr - Fresh	10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Tiefus</u>	SW - Slightly weathered	Water inflow	<u>m23241</u>
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	
Driller <u>G. Curry</u>	HW - Highly weathered	Complete drilling water loss	
Commenced <u>24/1/79</u>	EW - Extremely weathered		
Completed <u>24/1/79</u>	Notes		
Logged by <u>DC Purcell</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

HOLE No. 6
Canberra 255

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION ---
CO-ORDINATES 603.485N 210.598E R L OF 562.83m

SHEET 1 OF 1

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography Negative No.
Feed <u>Hydraulic</u>	Fr - Fresh	▼ 10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	▶ Water inflow	2.3-7.3 m 2324
<u>NMLC</u>	MW - Moderately weathered	▶ Partial drilling water loss	7.3-11.0
Driller <u>V. O'Brien</u>	HW - Highly weathered	▶ Complete drilling water loss	
Commenced <u>18:7:78</u>	EW - Extremely weathered		
Completed <u>18:7:78</u>	<u>Notes</u>		
Logged by <u>P. VandenBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

HOLE No. 6_A
Canberra 256

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION ---
CO-ORDINATES 603 640N 210 532E R L OF 562.69m

SHEET 1 OF 1

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography Negative No.
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained	10 Oct 73 water level date shown	Depth (m) B B W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	Water Inflow	283-81 m 23241
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	81-112
Driller <u>V. O'Brien</u>	HW - Highly weathered	Complete drilling water loss	
Commenced <u>1:7:78</u>	EW - Extrematly weathered		
Completed <u>1:7:78</u>	<u>Notes</u>		
Logged by <u>P. VandenBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	*Water Pressure Tests		

HOLE No. 7
Canberra 257

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION ---
CO-ORDINATES 603 723 N 210 488 E R L OF 563.79 m

SHEET 1 OF 1

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography Negative No
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained	10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	Water inflow	1.45 - 6.4 m 23241
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	6.4 - 11.5
Driller <u>Y. O'Brien</u>	HW - Highly weathered	Complete drilling water loss	11.5 - 13.0
Commenced <u>6.7.78</u>	EW - Extremely weathered		
Completed <u>7.7.78</u>	<u>Notes</u>		
Logged by <u>P. VandenBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION
CO-ORDINATES 603 863N 210 423E R L OF 555.40mSHEET 1 OF 1

Drilling information					Rock Substance		Rock Mass Defects							
Method	Drilling rate	Casing	Water	Pressure test #	Lift & % core recovery	Depth (metres)	Graphic log & core loss	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	0-3 Peni load 1-0 strength 5-0 strength 10-0 strength	Defect spacing (cm)	R O D	Defect description thickness, type, inclination, planarity, roughness, coating strength	Rock condition No (interpretative)
Auger						0								
						1		No core						
Diamond core						2	90	Quartz gravel						
						3	80	Yellow - brown weathered MUDSTONE	EW HW MW			10	Totally fragmented.	4
						4	100	Pale brown to pale grey MUDSTONE (City Hill Shale)	SW	X		50	Closely jointed with iron staining on joint surfaces.	3
						5	100		FS			90	Moderately jointed with some drilling induced breaks	
						6	100							
						7	100		SW			30	Closely spaced joints, iron stained and moderately to steeply dipping.	3-4
						8	100			X		10		
						9	100		FS	X		30	Mostly drilling induced breaks on rough irregular surfaces.	3
						10	100	Pale grey to blue-grey MUDSTONE , pyritic from 13.8 m to 13.9 m	Fr	X		80	Moderately to closely jointed, limonite staining on joints.	
						11	100		SW			60	Possibly some cleavage developed by shearing pressure; probably not parallel to bedding.	3-4
						12	100					30		
						13	100		Fr			10	Closely jointed and some drilling induced breaks.	
						14	100		FS			10	Joint's surface not stained.	
						15	100					0		
						15	End of hole RL = 550.40							

Drill type Pioneer
Feed Hydraulic
Core barrel type Triefus
NMLS
Driller VO'Brian
Commenced 29.6.78
Completed 30.6.78
Logged by P.VanderBroek
Vertical scale 1:100Weathering
Fr - Fresh FS - Fresh Stained
SW - Slightly weathered
MW - Moderately weathered
HW - Highly weathered
EW - Extremely weathered

Notes

Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis
* Water Pressure Tests

Water

10 Oct 73 water level date shown
Water inflow
Partial drilling water loss
Complete drilling water loss

Core Photography Negative No

Depth (m) B & W Colour
13-7.0 M23241
7.0-11.4
11.4-15.0





GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION ---
CO-ORDINATES 603 967 N 210 322 E R.L. OF 566.87mSHEET 1 OF 1

Drilling information					Rock Substance		Rock Mass Defects						
Method	Drilling rate	Casing	Water	Pressure test *	Lift & % core recovery	Depth (metres)	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	Point load 1-0 strength 3-0 strength 10-0 strength	Defect spacing (cm)	D. O. R.	Defect description thickness, type, inclination, planarity, roughness, coating strength	Rock condition No. (interpretative)
Auger					15.9.78	0	No core						
					28.8.78	1							
					3.9.78	2							
Diamond core						93	MUDSTONE Blue-grey in colour where SW to Fr	MW			45	Bedding and joint partings fractured, some clay	5
						100					0	Joints and bedding plane partings gradually become more widely spaced with depth. Generally no clay on defect surfaces.	
						100		SW			60		
						100					75	Bedding at about 60°	
						100					33	Joints generally follow bedding and often normal to bedding and dip at about 30° (range 10°-30°)	3-4
						100					85		
						100					60	Most defects stained from surface to about 8 m Sheared zones at : 7.5 - 7.6 m 8.15 - 8.35 m (little or no clay)	
						100		Fr to FS			80		
						100					100		
						100					100		
						100					100		
						100					25	Closely intersecting joints at 15.4 m and between 12.9 to 13.7 m	3 Some 2
						100					80		
						100					95		
						100					95		
						100					80		
					18								
End of hole: R.L. = 548.87													

Drill type Pioneer
Feed Hydraulic
Core barrel type Triefus
NMLC
Driller V. Q'Brien
Commenced 12.7.78
Completed 13.7.78
Logged by D. Purcell
Vertical scale 1:100Weathering
Fr - Fresh FS - Fresh Stained.
SW - Slightly weathered
MW - Moderately weathered
HW - Highly weathered
EW - Extremely weathered
Notes
Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis
* Water Pressure TestsWater
10 Oct 73 water level date shown
Water inflow
Partial drilling water loss
Complete drilling water lossCore Photography Negative No
Depth (m) B & W Colour
2.10 = 6.2 M 23 24 L
6.7 = 11.55
11.55 = 15.4
15.4 = 18.0

SHEET 1 OF 1

Drill type <u>Pioneer</u>	<u>Weathering</u>	Water	Core Photography Negative No
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained	 10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	 Water Inflow	1.65 - 6.6 70 23 24 1
<u>NMLC</u>	MW - Moderately weathered	 Partial drilling water loss	6.6 - 10.5
Driller <u>V. O'Brien</u>	HW - Highly weathered	 Complete drilling water loss	11.5 - 16.4
Commenced <u>4-7-78</u>	EW - Extremely weathered		16.4 - 18.05
Completed <u>5-7-78</u>	<u>Notes</u>		
Logged by <u>P. VanderBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

HOLE No. 9
Canberra 261

SHEET 1 OF 1

Drill type <u>Pioneer</u>	<u>Weathering</u>	<u>Water</u>	<u>Core Photography Negative No</u>
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained	10 Oct 73 water level date shown	Depth (m) B B W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	Water inflow	20 - 70 m 23 24 1
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	70 - 120
Driller <u>V. O'Brien</u>	HW - Highly weathered	Complete drilling water loss	12 - 150
Commenced <u>10-7-78</u>	EW - Extremely weathered		
Completed <u>11-7-78</u>	<u>Notes</u>		
Logged by <u>P. VanderBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

HOLE No. 9A
Canberra 262

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION _____
CO-ORDINATES 604.330 N 210.535 E R L OF 564.97 m


SHEET 1 OF 1

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography Negative No.
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained	10 Oct 73 water level date shown	Depth (m) B & W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	Water Inflow	20 - 20 ML3241
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	20 - 11.8
Driller <u>V.O'Brien</u>	HW - Highly weathered	Complete drilling water loss	11.8 - 15.0
Commenced <u>5.7.78</u>	EW - Extremely weathered		
Completed <u>6.7.78</u>	<u>Notes</u>		
Logged by <u>P.VanderBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	* Water Pressure Tests		

HOLE No. 10
Canberra 263

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION _____
CO-ORDINATES 604 464 N 210 557 E R L OF 565 86 m

SHEET 1 OF 1

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography Negative No.
Feed <u>Hydraulic</u>	Fr - Fresh	 10 Oct 73 water level date shown Water inflow Partial drilling water loss Complete drilling water loss	Depth (m) B & W Colour
Core barrel type <u>Triefur</u>	SW - Slightly weathered		1.5 - 5.85 3023 241
<u>NMBC</u>	MW - Moderately weathered		5.85 - 10.8
Driller <u>V. O'Brien</u>	HW - Highly weathered		10.8 - 15.05
Commenced <u>28.7.78</u>	EW - Extremely weathered		
Completed <u>28.7.78</u>	<u>Notes</u>		
Logged by <u>P. VandenBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis		
Vertical scale <u>1:100</u>	*Water Pressure Tests		

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION ---
CO-ORDINATES 604.563N 210.436E R L OF 563.07m

SHEET 1 OF 1

Drill type <u>Pioneer</u>	Weathering	Water	Core Photography	Negative No
Feed <u>Hydraulic</u>	Fr - Fresh FS - Fresh Stained.	10 Oct 73 water level date shown	Depth (m)	B & W Colour
Core barrel type <u>Triefus</u>	SW - Slightly weathered	Water inflow	140-60	m 23 24
<u>NMLC</u>	MW - Moderately weathered	Partial drilling water loss	60-100	
Driller <u>V.O'Brien</u>	HW - Highly weathered	Complete drilling water loss	100-120	
Commenced <u>18.8.78</u>	EW - Extremely weathered			
Completed <u>21.8.78</u>	<u>Notes</u>			
Logged by <u>P. VandenBroek</u>	Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis			
Vertical scale <u>1:100</u>	* Water Pressure Tests			

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION
CO-ORDINATES 604 633N 210 437E R L OF 561.92mSHEET 1 OF 1

Drilling information					Rock Substance		Rock Mass Defects						
Method	Drilling rate	Casing	Water	Pressure test #	Lift & % core recovery	Depth (metres)	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	Point load 1.0 strength 5.0 strength 10.0 strength 15.0 strength	Defect spacing (cm)	R O D	Defect description thickness, type, inclination, planarity, roughness, coating strength	Rock condition No (interpretive)
Auger						0							
						1							
					14.9-78 28.8-78	2	Not cored						
					60	3	Topsoil (Driller's report)					Core loss	
					100	4	Grey calcareous MUDSTONE (city Hill Shale)	SW	*		20	Closely to moderately jointed rock with manganese staining on joint surfaces.	
					100	5					40	Fragmented	
					100	6					50	Fragmented	3-4
					80	7	Blue-grey calcareous MUDSTONE	Fr	*		80	clay seam	
						8	End of hole RL= 553.92				85	Mostly moderately jointed with uncoated joint surfaces.	3
												Core loss	

Drill type Pioneer
Feed Hydraulic
Core barrel type Triefus
NMLS
Driller V.O'Brian
Commenced 21.8.78
Completed 21.8.78
Logged by P.VandenBroek
Vertical scale 1:100Weathering
Fr - Fresh
SW - Slightly weathered
MW - Moderately weathered
HW - Highly weathered
EW - Extremely weathered

Notes

Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis
* Water Pressure Tests

Water

10 Oct 73 water level date shown
Water inflow
Partial drilling water loss
Complete drilling water loss

Core Photography Negative No

Depth (m) B & W Colour
21.6.9 M23241
6.9-8.0

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (8) 90° DIRECTION
CO-ORDINATES 604 70E N 210 43E R L OF 562.74 mSHEET 1 OF 1

Drilling information					Rock Substance		Rock Mass Defects						
Method	Drilling rate	Casing	Water	Pressure test *	Lift & % core recovery	Depth (metres)	Graphic log Bore loss	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	Defect spacing (cm)	R O D	Defect description thickness, type, inclination, planarity, roughness, coating strength	Rock condition No (interpretative)
Auger						0							
						1		Not cored					
						2							
						2.100		Dark grey organic SILT	OL				
						2.300		Green-grey MUDSTONE	SW				
						2.500							
						2.700		core loss					
						2.900							
						3.100							
						3.300							
Diamond core	PIPE					3.500							
						3.700							
						3.900							
						4.100							
						4.300							
						4.500							
						4.700							
						4.900							
						5.100							
						5.300							
Tunnel						5.500							
						5.700							
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						40.900							
						41.100							
						41.300							
						41.500							

GEOLOGICAL LOG OF DRILL HOLE

ANGLE FROM HORIZONTAL (θ) 90° DIRECTION
CO-ORDINATES 604 780N 210 439E R.L. OF 562.52 mSHEET 1 OF 1

Drilling information					Rock Substance		Rock Mass Defects						
Method	Drilling rate	Casing	Water	Pressure test #	Lift & % core recovery	Depth (metres)	Substance description rock type, grain characteristics, colour, structure, minor components	Weathering	Point load 100 strength 100 Is (50 MPa)	Defect spacing (cm)	R.O.D.	Defect description thickness, type, inclination, planarity, roughness, coating strength	Rock condition No (inter-plate)
Auger						0							
						1							
						2	Not cored						
						3	Dark grey organic SILT	OL					
						4	Pale green-grey, fresh stained, calcareous MUDSTONE						
						5	(City Hill Shale)	FS					
						6							
						7							
Diamond Core	PIPE	Tunnel				8	Clay						
							End of hole R.L. = 554.52						

Drill type Pioneer
Feed Hydraulic
Core barrel type Triplex
NMLI
Driller V.O. Brian
Commenced 23.8.78
Completed 23.8.78
Logged by P. VandenBroek
Vertical scale 1:100
Record 1978/99

Weathering
Fr - Fresh FS - Fresh Stained.
SW - Slightly weathered
MW - Moderately weathered
HW - Highly weathered
EW - Extremely weathered
Notes
Bedding & Joint Planes - Angles are measured relative to a plane normal to the core axis
* Water Pressure Tests

Water
10 Oct 73 water level date shown
Water inflow
Partial drilling water loss
Complete drilling water loss

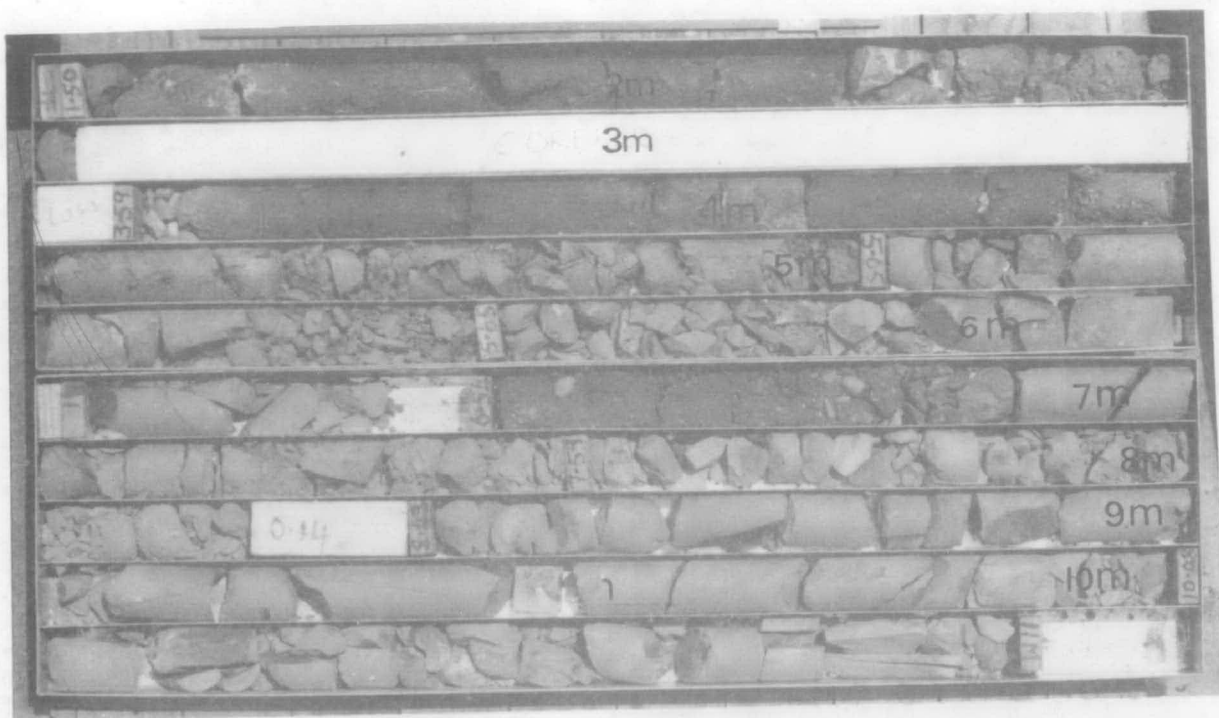
Core Photography Negative No
Depth (m) B & W Colour
20.6-72 M23241
673-80

APPENDIX 3

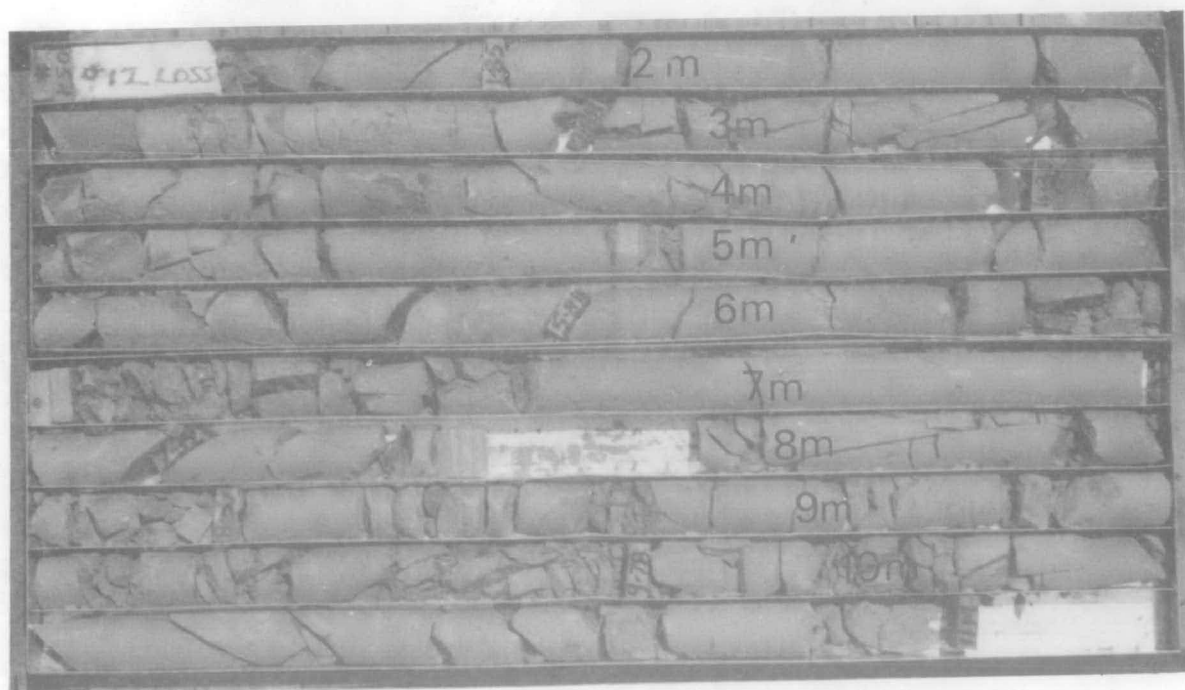
DRILLCORE PHOTOGRAPHS

SULLIVANS CREEK SEWER TUNNEL

HOLE 1

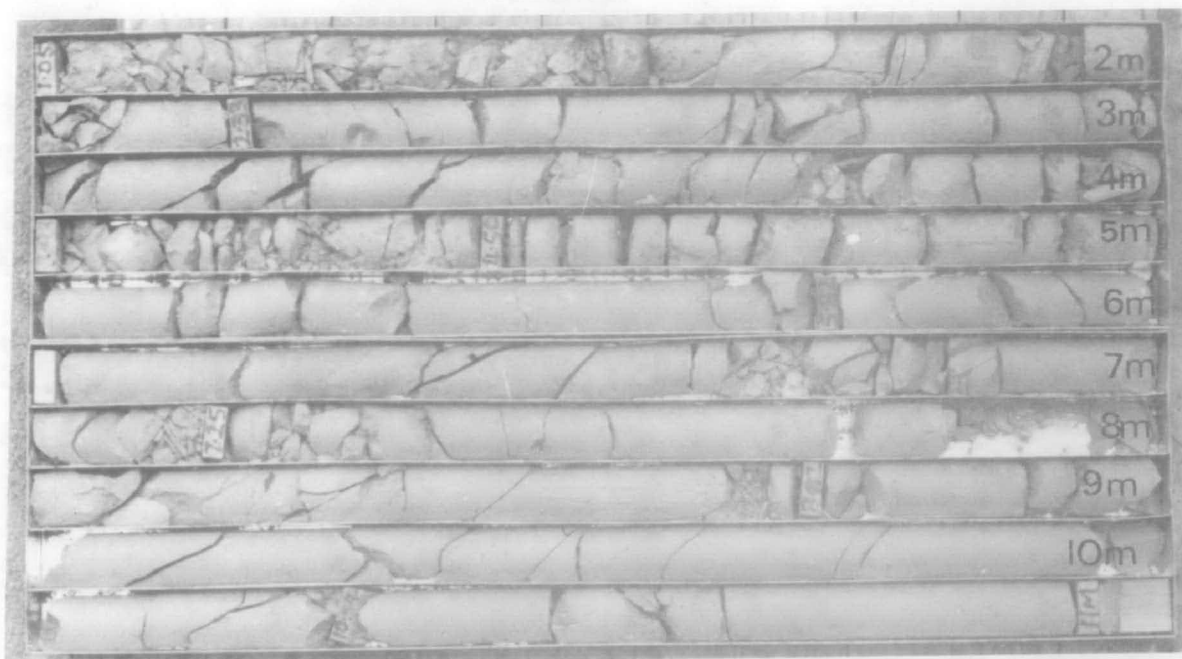


HOLE 1A

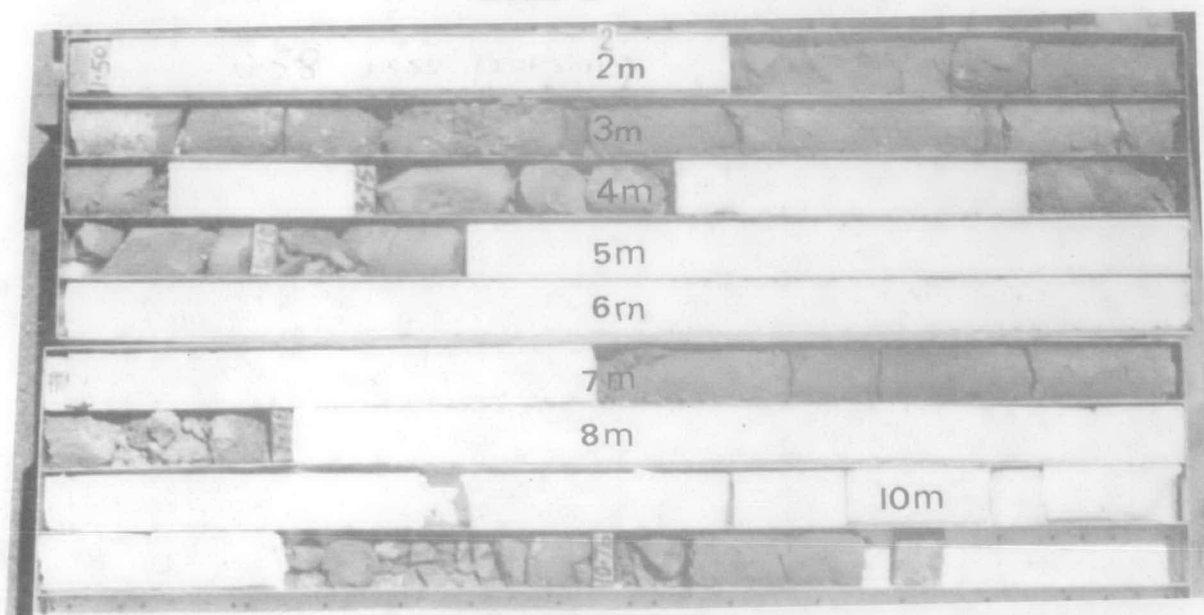


SULLIVANS CREEK SEWER TUNNEL

HOLE 1C



HOLE 2

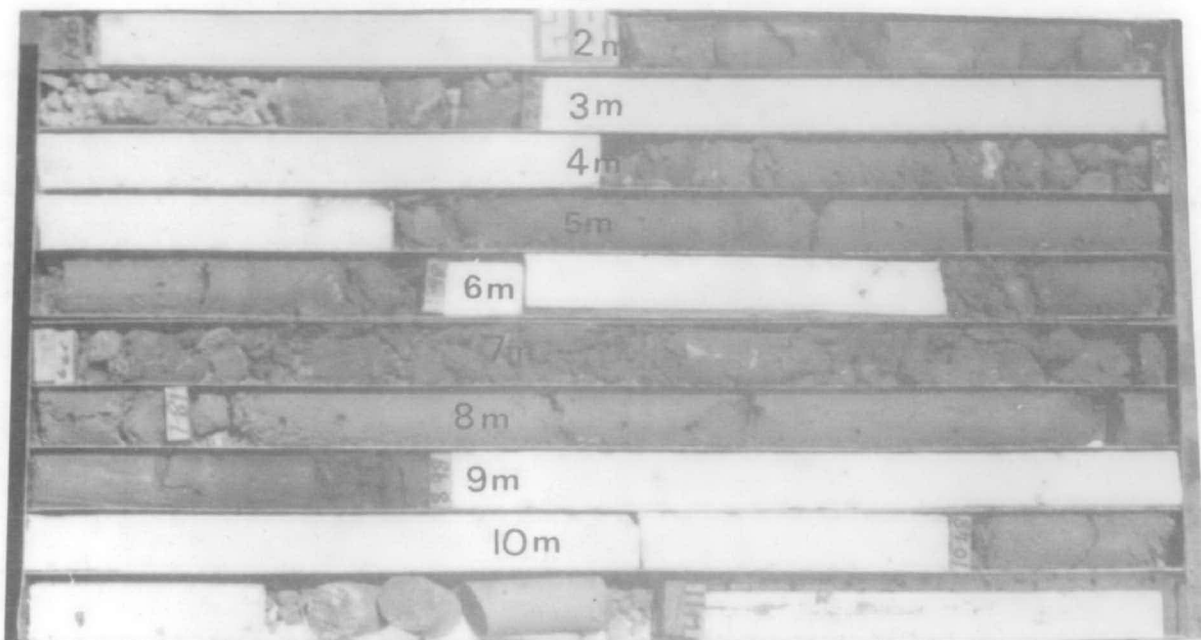


HOLE 2A

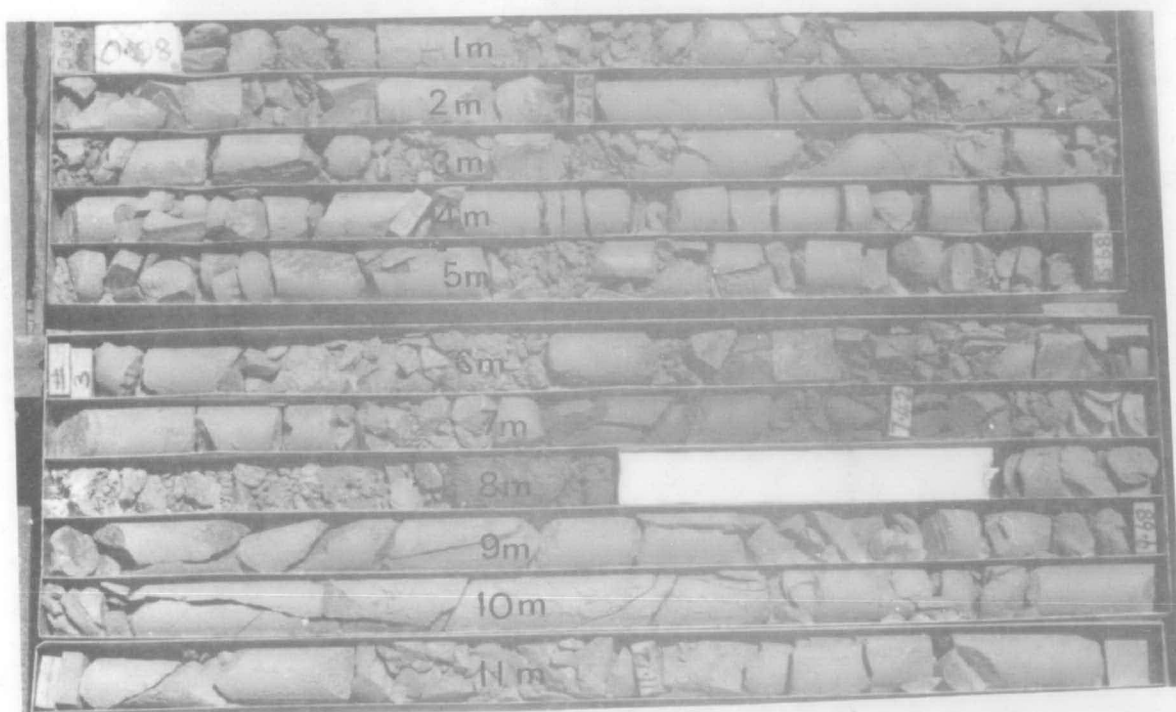


SULLIVANS CREEK SEWER TUNNEL

HOLE 2B



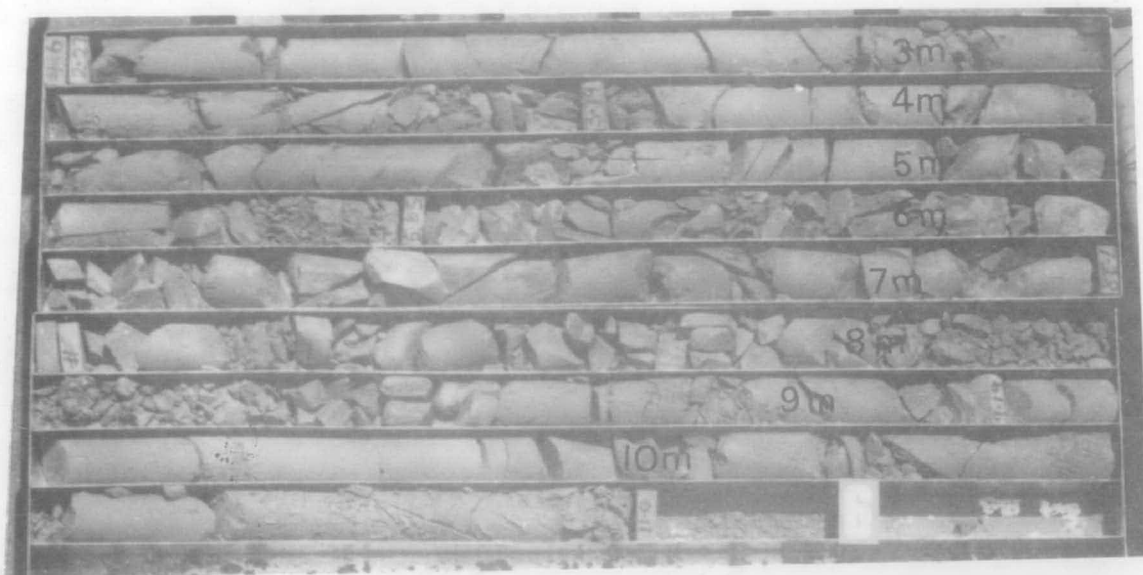
HOLE 3



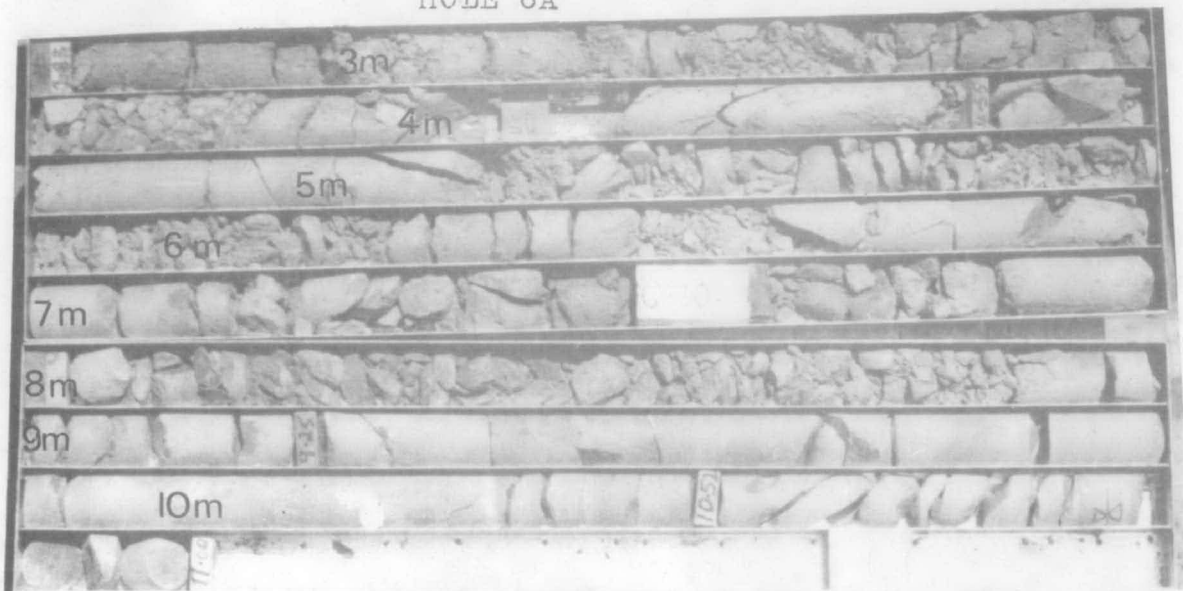
HOLE 4



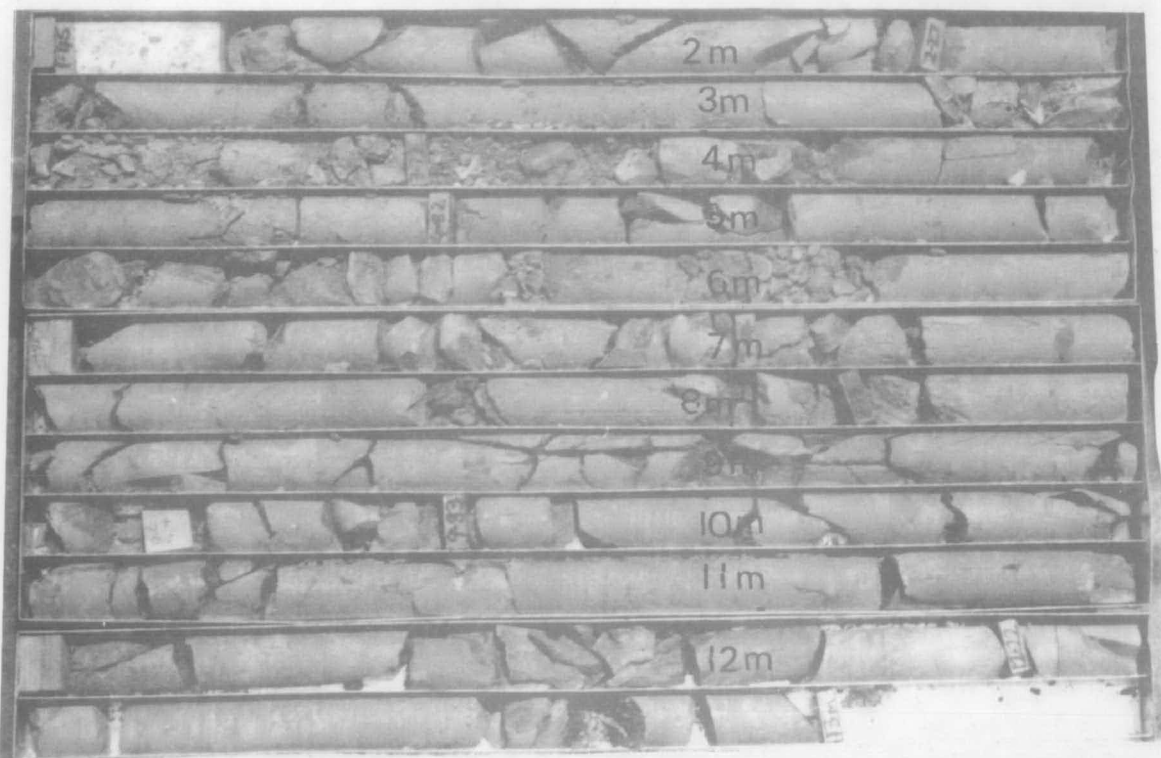
SULLIVANS CREEK SEWER TUNNEL
HOLE 6



HOLE 6A

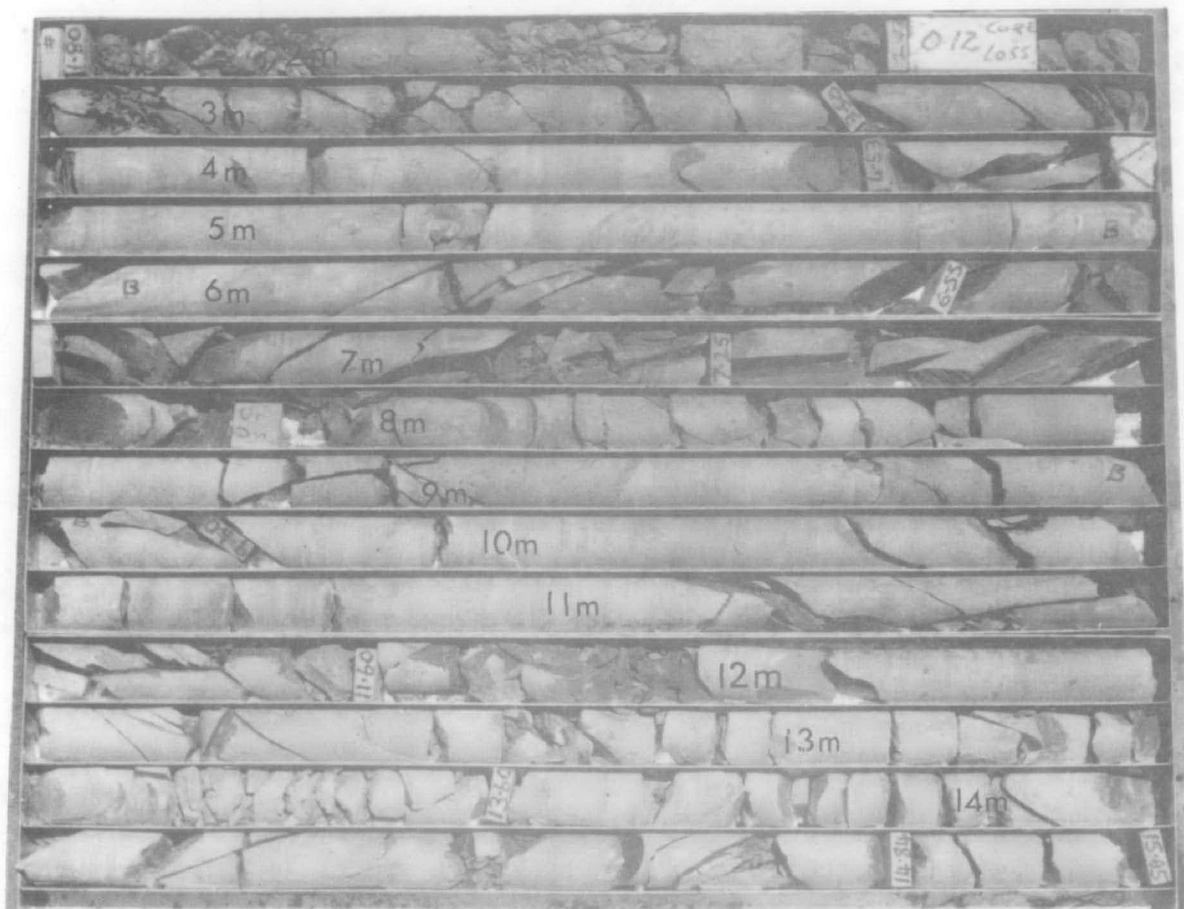


HOLE 7

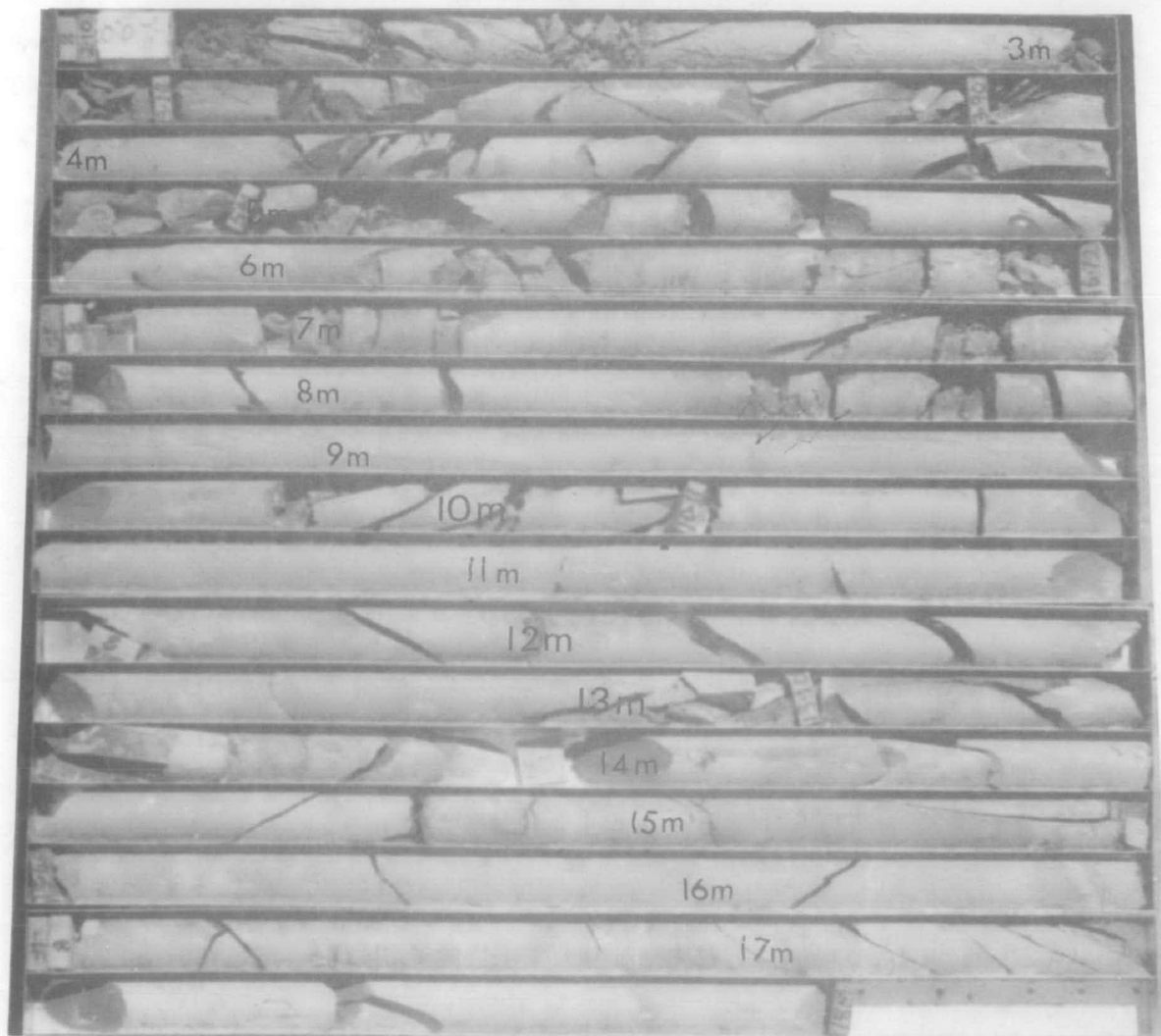


SULLIVANS CREEK SEWER TUNNEL

HOLE 7A

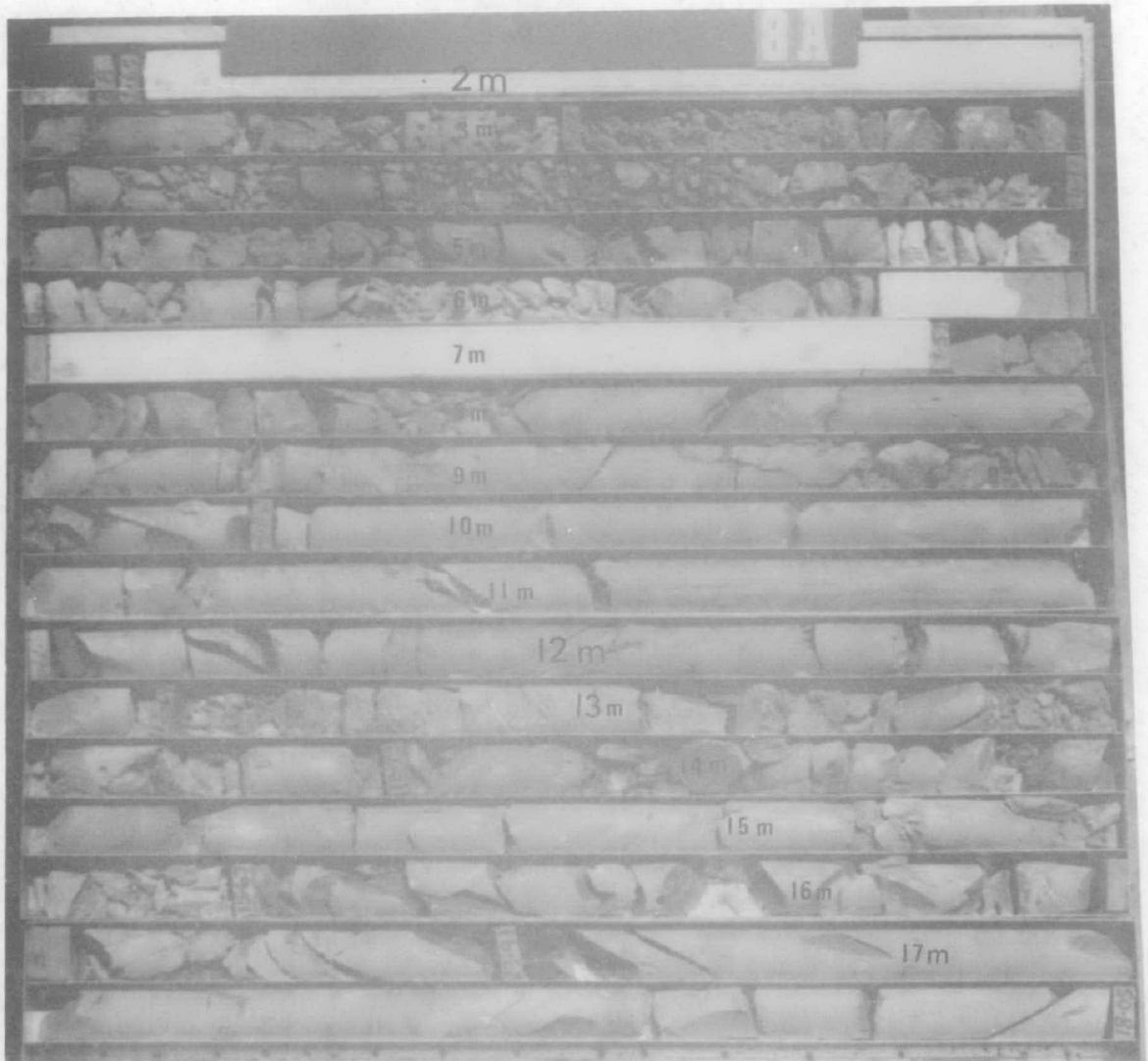


HOLE 8

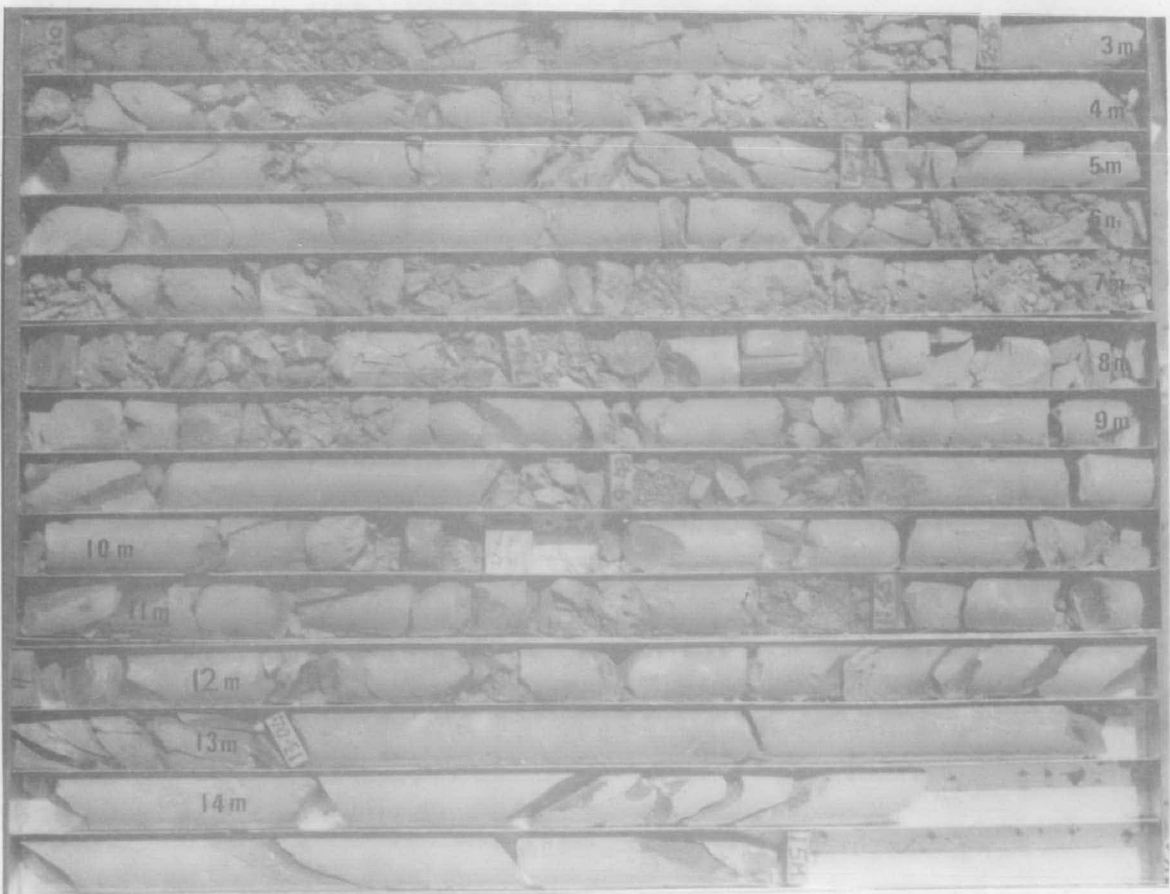


SULLIVANS CREEK SEWER TUNNEL

HOLE 8A



HOLE 9



SULLIVANS CREEK SEWER TUNNEL

HOLE 9A

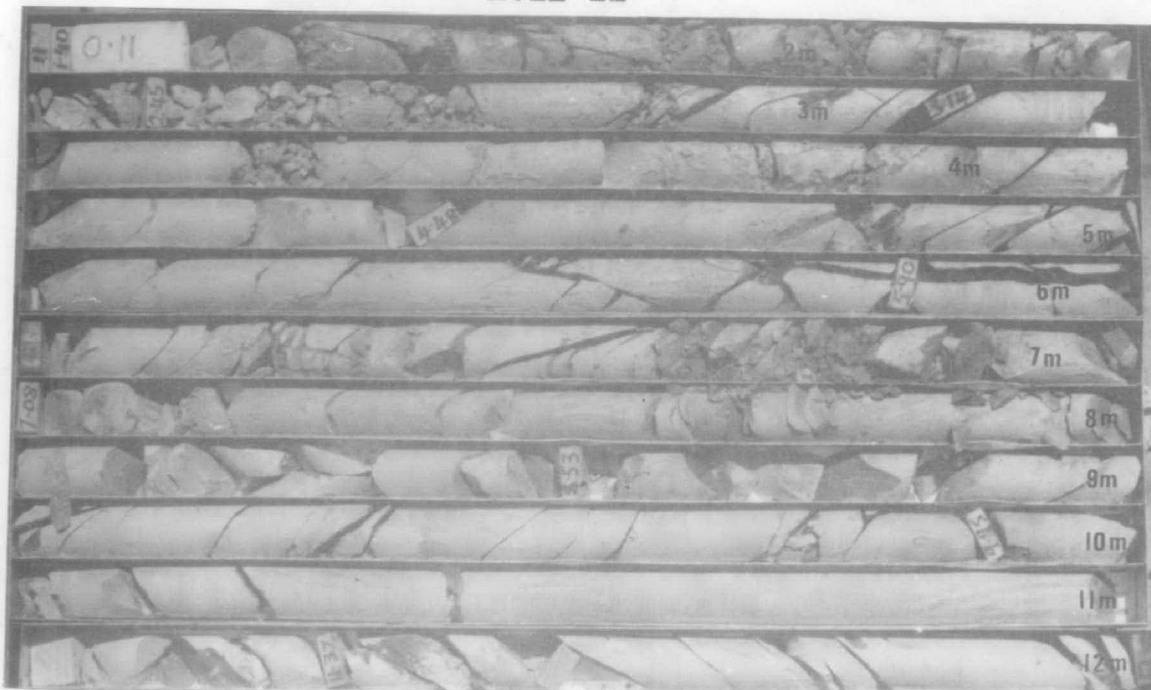


HOLE 10

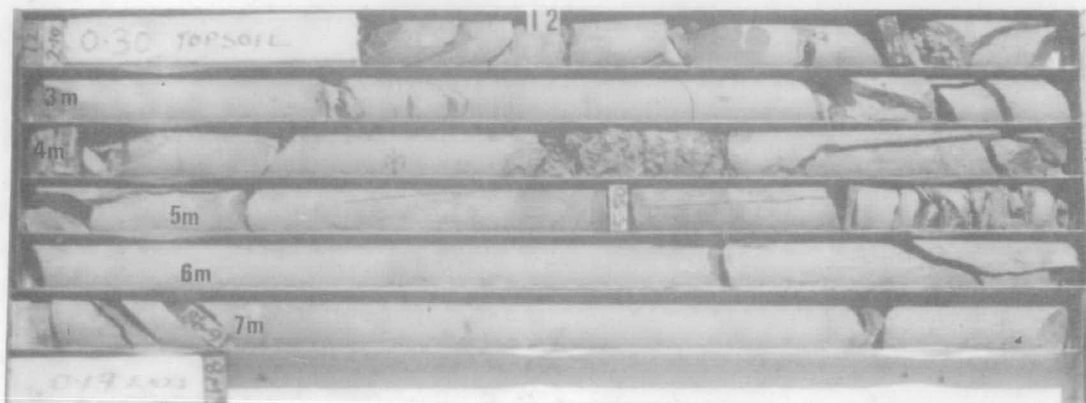


SULLIVANS CREEK SEWER TUNNEL

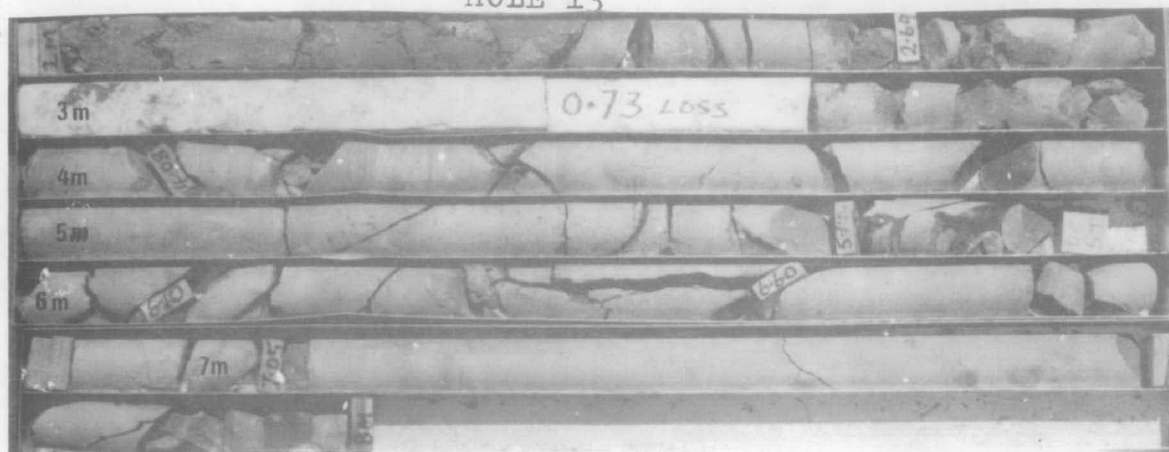
HOLE 11



HOLE 12

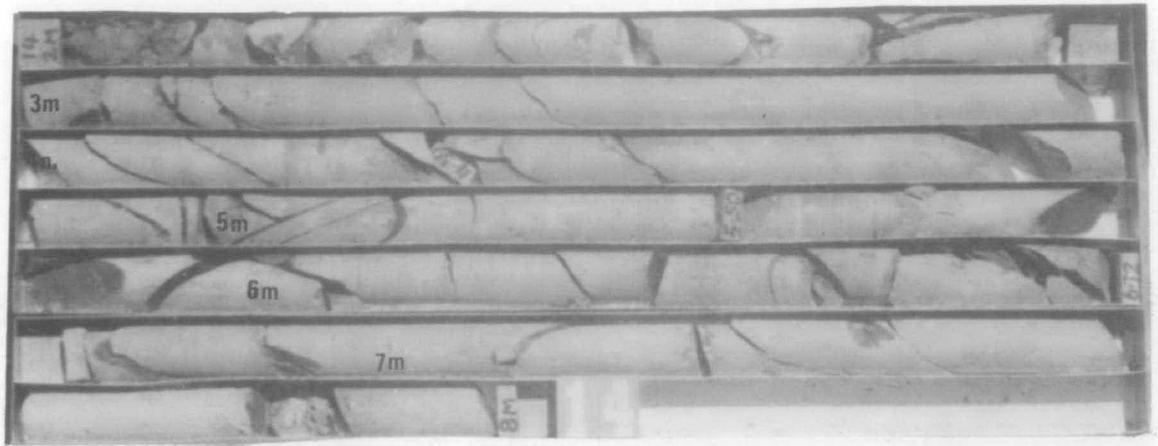


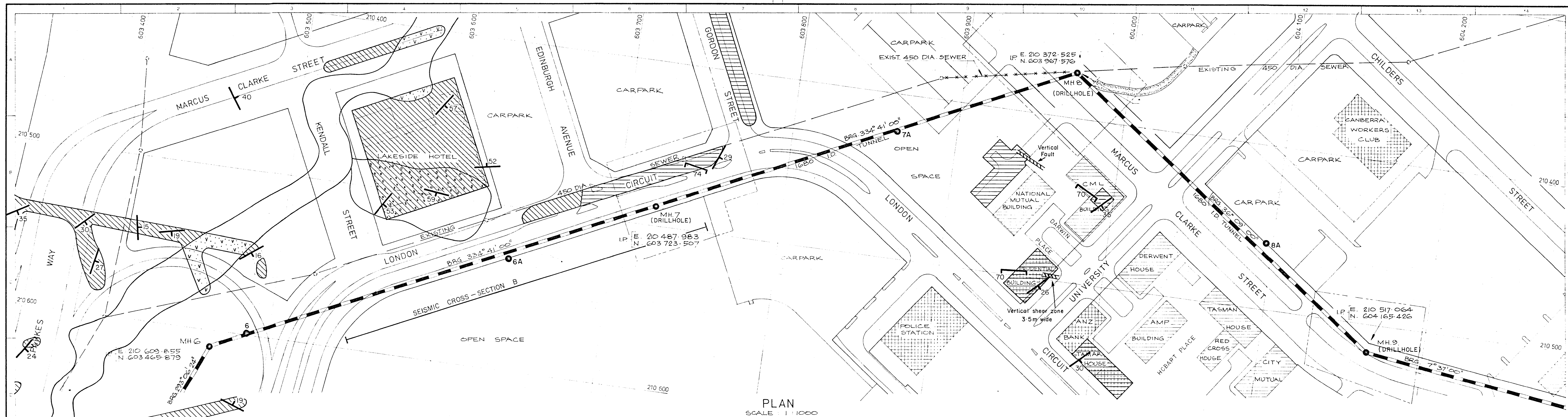
HOLE 13



SULLIVANS CREEK SEWER TUNNEL

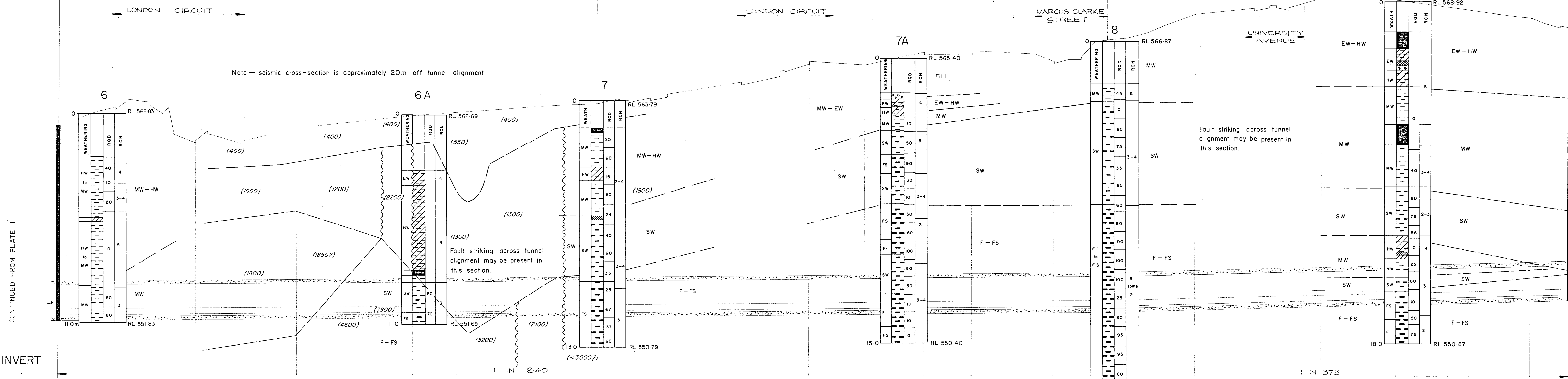
HOLE 14





PLAN
SCALE 1:1000

MANHOLE NO	MANHOLE
INVERT LEVEL	552.09 552.04 552.04
EXISTING SURFACE LEVEL	562.1 563.50 563.98 562.69 563.34 552.359 552.384 552.45 552.60 552.87 552.00 552.73 552.66 552.706 552.16 552.97 552.92 552.98 557.48 553.338 553.963



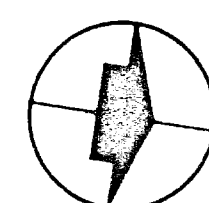
GRADE OF INVERT

DATUM LEVEL 548.00metres

TUNNEL CHAINAGE	7+00	8+00	9+00	10+00	11+00	12+00	13+00	14+00
ROCK TYPE, HARDNESS AND STRENGTH AT T.L.	Mostly pale yellow-brown to reddish brown, moderately weathered MUDSTONE with sections of pale grey to blue-grey, slightly weathered to fresh, calcareous MUDSTONE. Rocks will vary from soft and weak to hard and strong. Is (50)=3.8-4.2 expected from CH 7+40 to 8+20.		Mostly blue-grey calcareous MUDSTONE, moderately hard, moderately strong. Is (50)=1.5 Mpa.		Blue-grey fresh calcareous MUDSTONE, hard and strong. Is (50)=2.5-3.8 Mpa.		Variable from blue-grey fresh calcareous MUDSTONE to pale yellow-brown, lightly to moderately weathered MUDSTONE. Hard & strong in places, soft & weak in some sections.	
ROCK MASS DEFECTS	Bedding variable from thin bedded to thick bedded, dip/dip direction=25/060 expected, orientation expected to be fair. Four joint sets observed in core, these were unaltered, smooth and very closely spaced with some sections fragmented. Widely spaced shears and clay seams are expected and one or two small faults could occur along this section. Defect orientations are expected to be unfavourable.		Bedding commonly not visible in core due to unweathered nature of rock. Four joint sets observed in core; these were unaltered, smooth and closely spaced. Defect orientation expected to be fair. Generally, seams & shears should be absent.		Bedding not visible in core due to unweathered nature of rock. Two or three joint sets observed in core, joints are tight, uncoated, rough, planar and moderately wide to closely spaced. Defect orientations are expected to be fair. Seams, shears and faults are expected to be absent.		Thinly bedded, dip/direction=25/060 expected, orientation of bedding favourable. Two joint sets & some random jointing observed in core. Joints tight; surfaces uncoated or stained, rough, planar & closely spaced.	
ROCK MASS QUALITY(Q)	Sections of extremely poor to very poor ground, possibly some exceptionally poor sections (possibly O).		Very poor to poor proceeding NW (O.8)		Good (11.3-15.4)		Fair to poor (1.2)	
STANDUP TIME	About 1 hour calculated from drill holes 6 and 6A.		10 hours calculated from drill hole 7		1 week to one month calculated from drill holes 7A and 8		2-10 hours calculated from drill hole 8A	
OVERBREAK (O.B)	CH 7+40 approximate limit of machine excavation from the southern portal. Some large OB (>1m) expected if drill and blast methods are used from CH 8+00 to 8+80.		Some O.B. pockets to 0.5m expected		Minimum O.B. expected, some pockets to 0.3m expected, depending on defects		O.B. pockets to 1m possible in blocky and seamy sections	
SUPPORT	Large sections of support category 33 and 29 with possibly small sections of category 25		Categories 25 and O improving towards CH 10+00		Category O		Category O to 25 with some 29 possible	
GROUNDWATER INFLOWS	Minor inflows only expected. Water table fluctuates widely in drill hole 6 from within 2m of surface after heavy rain to 2m above crown during dry spells		Minor inflows initially, section expected to dry out after excavation		Minor inflows initially, most of section expected to dry out after excavation.		Minor inflows initially, much of section expected to dry out after excavation.	

LONGITUDINAL SECTION

SCALE: VERTICAL 1:100
HORIZONTAL 1:1000



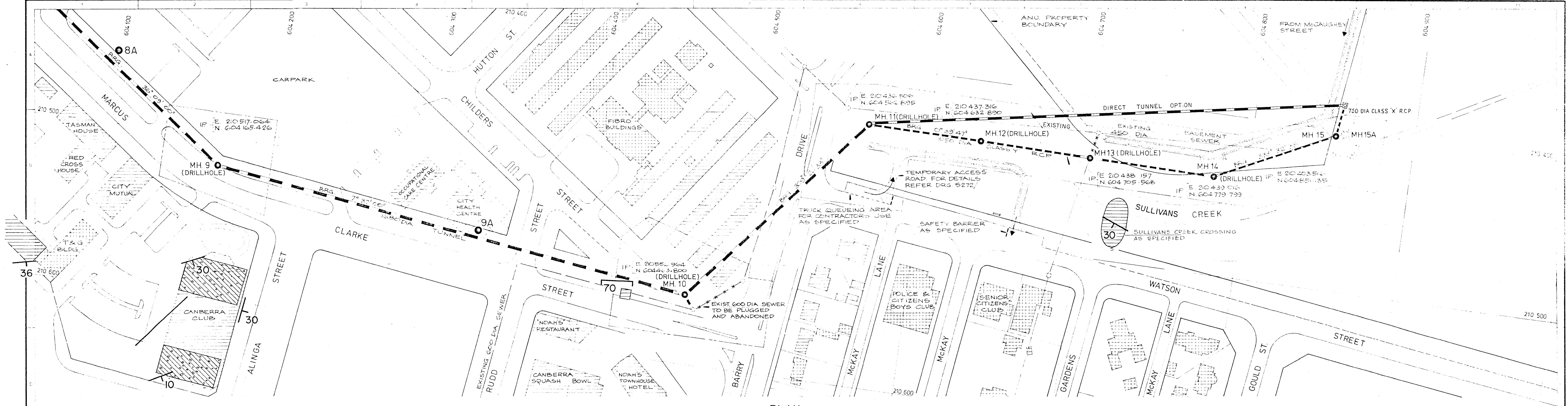
NOTE
1. FOR LEGEND REFER TO PLATE 1

REVISION	DESCRIPTION	DATE	REVISION	DESCRIPTION	DATE
1			2		
3			4		
5			6		
7			8		
9			10		

NO	DESCRIPTION	AUTHOR	DATE	CHECKED	DATE
1					
2					
3					
4					
5					
6					

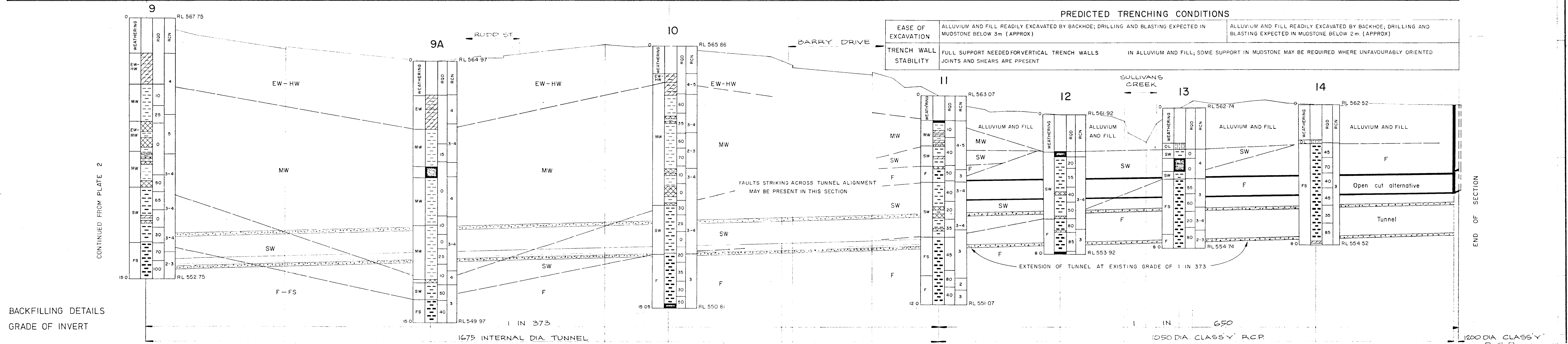
0	10	20	30	40	50
SCALE					
BASE MAP SURVEY					
Supplied by Scott and Furphy Eng. Pty Ltd					
GEOLOGIST BY					
P.H. Vanden Broek					
COMPILED BY					
G. Sporkman					
CHECKED AND APPROVED					
D.C. PURCELL					
SENIOR GEOLOGIST					
E.G. Wilson					
SUPERVISING GEOLOGIST					

COMMONWEALTH OF AUSTRALIA	
BUREAU OF MINERAL RESOURCES	
CANNBERRA, ACT	
TITLE	
SULLIVANS CREEK SEWER TUNNEL	
PROJECT	
GEOLOGICAL PLAN AND SECTION	
CHAINAGE: 6+05 to 14+05	
Record 1977/99	DRAWN BY
155/A16/2214	



PLAN
SCALE: 1:1000

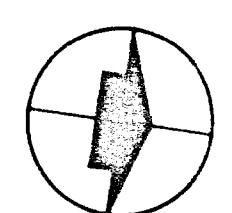
MANHOLE NO	INVERT LEVEL	EXISTING SURFACE LEVEL
MH 9	553.38	557.48
MH 9A	553.38	557.48
MH 10	554.16	557.48
MH 11	554.52	557.48
MH 12	557.66	557.48
MH 13	557.28	557.48
MH 14	557.39	557.48
MH 15	557.51	557.48
MH 15A	557.51	557.48



BACKFILLING DETAILS
GRADE OF INVERT

DATUM LEVEL 548.00m		15+00	16+00	17+00	18+00	19+00	20+00	21+00
TUNNEL CHAINAGE								
ROCK TYPE, HARDNESS AND STRENGTH AT T.L.		PALE GREY-BROWN, SLIGHTLY WEATHERED MUDSTONE, WEAK & SOFT, 1.5(50) 0.6-1.0	MOSTLY PALE TO YELLOW-BROWN, MODERATELY WEATHERED MUDSTONE, WEAK AND SOFT, 1.5(50) 0.6-1.0	MOSTLY PALE GREY SLIGHTLY WEATHERED CALCAREOUS MUDSTONE, MODERATELY HARD, MODERATELY STRONG 1.5(50) 1.0-3.0 INCREASING AMOUNTS OF FRESH BLUE CALCAREOUS MUDSTONE COMING INTO TUNNEL TOWARDS CH 19+50				PALE GREY TO GREEN-GRAY FRESH MUDSTONE, HARD AND STRONG 1.5(60) 1.4-2-5-5
ROCK MASS DEFECTS		FOUR DEFECT ORIENTATIONS OBSERVED IN CORE, MODERATELY WIDE TO CLOSELY SPACED JOINTS, STAINED, PLANAR AND SMOOTH.	THREE JOINT ORIENTATIONS OBSERVED IN CORE, JOINT SURFACES ARE TIGHT, UNALTERED, UNCOATED, CLOSELY SPACED AND ORIENTATIONS ARE EXPECTED TO BE FAIR TO FAVOURABLE. WIDELY SPACED CLAY SEAMS OR SHEARS MAY BE PRESENT	UP TO FOUR JOINTS ORIENTATIONS WERE OBSERVED IN THE CORE, JOINT SURFACES ARE MOSTLY UNCOATED AND STAINED PLANAR AND SMOOTH JOINTS BECOMING TIGHTER TOWARDS CH 19+50. ORIENTATIONS EXPECTED TO BE UNFAVOURABLE. POSSIBILITY OF A SMALL FAULT ALONG THIS SECTION WIDELY SPREAD CLAY SEAMS AND SHEARS LIKELY.				AT LEAST FOUR JOINT ORIENTATIONS OBSERVED IN DRILL CORE JOINT SURFACES ARE UNALTERED, UNCOATED, STAINED SMOOTH PLANAR AND MODERATELY WIDE TO CLOSELY SPACED JOINT ORIENTATIONS EXPECTED TO BE FAIR FEW CLAY SEAMS OR SHEARS ARE EXPECTED
PREDICTED TUNNELLING CONDITIONS	ROCK MASS QUALITY (Q)	VERY POOR (0-53)	VERY POOR (0-33)	POOR TO FAIR (0-34-3-2) TOWARDS CH 19+50				POOR TO FAIR (1-4-5-4)
	STAND-UP TIME	10 HOURS CALCULATED FROM DRILLHOLE 9	2 HOURS CALCULATED FROM DRILLHOLE 9A	2 HOURS, 10 HOURS AND 1 WEEK CALCULATED FROM DRILL HOLES 10, 11 AND 12 RESPECTIVELY				2 DAYS CALCULATED FROM DRILL HOLES 13 AND 14
	OVERBREAK	0.8 POCKETS TO 1m POSSIBLE IN BLOCKY AND SEAMY SECTIONS	DEPENDENT ON METHOD OF EXCAVATION AND DIRECTION OF TUNNEL ADVANCE	DEPENDENT ON EXCAVATION METHODS 0.8 TO 1m IF DRILL AND BLAST METHODS ARE USED				VERY LITTLE OR NO O.B. EXPECTED
	SUPPORT	CATEGORY 25	MOSTLY CATEGORY 25 WITH POSSIBILITY OF 29	MOSTLY CATEGORIES 0 AND 25 WITH POSSIBILITY OF SOME 29				CATEGORY 0
GROUNDWATER INFLOWS		MINOR INFLOWS 1-25 LITRES/MIN PER 10m SECTION LENGTH ONLY	MINOR INFLOWS ONLY	SMALL INFLOWS 25-125 LITRES/MIN PER 10m SECTION EXPECTED				SMALL INFLOWS 25-125 LITRES/MIN PER 10m SECTION EXPECTED

LONGITUDINAL SECTION
SCALES: VERTICAL 1:100
HORIZONTAL 1:1000



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BUREAU OF MINERAL RESOURCES			
SULLIVANS CREEK SEWER TUNNEL			
GEOLOGICAL PLAN AND SECTION CHAINAGE 14+15 TO 21+62			
TO ACCOMPANY			
DRAWN BY			
DRAWING NUMBER			