

DEPARTMENT OF MINES AND ENERGY
SOUTH AUSTRALIA

REPT.BK.NO. 85/51
STABILITY ASSESSMENT OF REHABILI-
TATION AREA, READYMIX-FARLEY
QUARTZITE QUARRY (RIVERVIEW
HIGHBURY).

GEOLOGICAL SURVEY

by

J.C. BEAL

GROUNDWATER & ENGINEERING

OPEN FILE

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DEPARTMENT OF MINES AND ENERGY
SOUTH AUSTRALIA

REPT BK NO. 85/51
DME NO.
DISK NO. 6

STABILITY ASSESSMENT OF REHABILITATION AREA, READYMIX -
FARLEY QUARTZITE QUARRY (RIVERVIEW), HIGHBURY

ABSTRACT

A slope stability analysis carried out at the Readymix - Farley Riverview quartzite quarry on a rehabilitated slope shows the slope to be unstable when saturated and marginally stable when dry. The lower section of the slope may be rendered stable for the saturated condition by the placement of toe-fill.

A maximum Factor of Safety of 1.3 was calculated.

Recommendations are to cease placement of fill and allow slope to stabilize prior to vegetation.

INTRODUCTION

Waste rockfill from quarrying operations at the Readymix - Farley Riverview Quarry is being used to rehabilitate exposed quarry faces (Figure 1. Plates 1 to 8). This rehabilitation is costed to the Extractive Areas Rehabilitation Fund which is administered by the Dept. for the Environment and Planning who rely upon the Department of Mines and Energy for project recommendations.

Rehabilitation in one area of the quarry is considered by the Mining Branch of this Department to be near completion.

The Chief Inspector of Mines has requested through the Chief Government Geologist that this area be inspected in relation to the following points:

1. The slope stability of the filled portion of the quarry in its present state.
2. The effect on slope stability of additional volumes of the fill.
3. The effect of moisture in the fill.
4. Any means of increasing the slope stability.

METHOD OF ASSESSMENT

Plans, cross-sections and surveyed traverses (Figure 2) of the face under investigation were supplied by Readymix - Farley and formed the basis of profiles (Figures 3 to 6) used in the slope - stability analyses.

Three analyses were used. A quick analysis described by Hoek and Bray. ('Rock slope Engineering' p. 247 - 253) and a more detailed analysis. [Janbu's non-circular failure analysis.]

There have been no field tests carried out and assumed parameter values are based upon commonly accepted values for the situation under discussion.

The third analysis was carried out using the STABL computer programme available to the Department through the South Australian Institute of Technology at The Levels.

All three methods arrive at a value for the Factor of Safety for the rehabilitated slope as it currently stands and after it has been modified with the placement of additional volumes of quarry waste as toe-fill.

THE FACTOR OF SAFETY (F.O.S)

Theoretically any F.O.S. greater than one indicates that a slope is more stable than unstable. For long term stability a minimum F.O.S. of 1.5 is thought to be required at this site although an F.O.S. as low as 1.3 may be considered acceptable for the following reasons:

- . Very large volumes of fill are not involved and costs of clean up operations resulting from a failure will only be minor, although re-vegetation of the slope may need to be carried out.

- Failure is not considered likely to interfere either with the course of the Torrens River or the Gorge Road, although failure may temporarily cut-off the lower quarry access road.

PARAMETER VALUES

Waste-fill slope angle:	35° measured from profiles constructed from client data. (A mean slope angle of 30° has been used for calculations incorporating a toe-fill).
Density of fill:	Material as tipped taken to be 13 kN/m ³ to 19.5 kN/m ³ . For the toe-fill a 10% increase of density from vehicle compaction was assumed.
Density of in situ rock:	Taken to be 25 kN/m ³ .
Cohesion:	Although there are isolated patches of silt-clay, the quarry waste is truck-dumped and a cohesion value of nil has been assigned to the fill. For the STABL analysis a cohesion value of 10 000 kPa has been assigned to the in situ rock.
Coefficient of internal friction	33°. (A value of 35° was assigned to the toe-fill assuming some compaction).
Profile of quarry floor and quarry face:	The floor is taken as horizontal and the quarry face, in part, as 65° for the non-circular slip analysis.
	Quarry profiles from Readymix Survey Plans RVQ3 and RVQ1 did not tie-in neatly with fill-slope calculated from plan supplied by SURTECH PTY.

LTD. and some adjustment was necessary to the cross-sections on RVQ3 and RVQ1 to make the SURTECH data fit to a sensible profile.

Water-table:

Because it is possible for fines to be washed out of the quarry waste and to infill bedrock fractures it is considered necessary, although the fill is in the main free-draining, to assume that a perched water-table may result following prolonged wet periods.

Of at least equal importance to fill-stability as the presence of a water-table is the saturation of the fill by direct precipitation. In pockets of fill having a relatively high silt/clay content the water is retained thus increasing the weight of the fill with no compensating increase in fill cohesion.

RESULTS

The Quick-Analysis

Table 1, Appendix 1, summarizes results of the quick-analysis for wet and dry slopes for varying slope angles.

A F.O.S. of 1.63 was obtained, based on dry fill, a slope of 25° and a friction angle of 38°. The dry condition is unrealistic and the low slope angle is considered to be economically undesirable.

The minimum F.O.S. (0.56) was obtained for a partially saturated fill, a slope of 35° and a high friction angle of 40°. An even lower F.O.S. would have resulted had a lower, more realistic friction angle been used and the slope fully saturated.

Slope angle is considered to be more important to the stable - unstable condition of the rehabilitated slope than expected values of cohesion or saturation. The relationship between slope angle and the F.O.S. is shown in Charts 1 and 2, Appendix 1 for dry and partially saturated slopes.

The accepted F.O.S. for this analysis for the current rehabilitated slope is:

for partly saturated fill :0.6
for dry fill :1.1

The Janbu non-circular slip analysis

Tables 2, 3 and 4, Appendix 1, show calculations using the non-circular slip analysis where it is assumed that to some extent the failure surface will coincide with the quarry face. Results of the calculations are summarized in Appendix 1 and tabulated below:

Table 2:	Dry fill; $c = 0$; $\phi = 35^\circ$; slope angle = 35°	F.O.S. = 1.12
Table 3:	Toe Saturated to water-table, level A. (Figure 4)	F.O.S. = 1.06
Table 3:	Toe saturated to water-table, level B. (Figure 4)	F.O.S. = 0.9
Table 4:	Toe-fill to water-table, level B. (Figure 5)	F.O.S. = 1.3
Table 5:	Toe-fill with compaction and water table at level B. (Figure 6)	F.O.S. = 1.5

From this analysis it can be concluded that maximum slope stability for failures passing through the toe can be achieved by placing a free-draining toe to approximately EL 164 with some compaction of the toe-fill.

The STABL Analysis

The programme repeatedly locates slip-failure surfaces and identifies the slip-circle associated with the minimum F.O.S. which in the cases examined range from 0.956 to 0.984, again underlining the unstable nature of the waste-fill. The slip circles in question are located above both water-table and toe-fill.

DISCUSSION

The analyses show that the placement of a toe-fill at the base of the rehabilitated slope will increase the existing F.O.S. of approx 1.0 to a maximum F.O.S. of 1.4 for failures passing through the toe of the slope. For higher seated failures a F.O.S. of approximately 1.0 prevails for both the wet and dry condition.

The continued tipping of quarry waste onto the existing slope will increase slope steepness and decrease slope stability. Should there be further disposal of waste this must be placed as free-draining toe-fill at an angle (approx. 25°) which will reduce the over-all slope from approximately 35° to 30°.

Toe-fill should be placed in maximum 1.0 metre thick lifts and preferably in 500 mm to 600 mm lifts. The lifts should be compacted by laden quarry trucks ensuring that the trucks traverse the full area of the lift surface and do not limit their movement to rut-routes.

CONCLUSIONS

It is advised that placement of further quarry-waste onto the rehabilitated slope in the area under discussion cease and that the existing slope be allowed to stabilize prior to vegetating. A standing period of two wet seasons is suggested. During this period measures should be taken to drain storm water away from the rehabilitated slope.

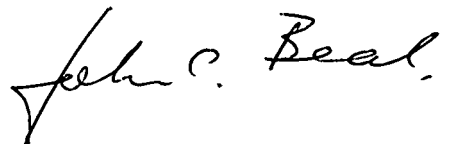
A toe-fill of free draining material having a slope angle no greater than 25° and a height to approximately EL 165 m could be placed along the foot of the rehabilitated slope but because this is not considered essential from a rehabilitation viewpoint use of rehabilitation funds cannot be recommended.

Reasons why placement of further quarry-waste should not be considered may be summarized:

1. Failure along slip-surfaces indicated by the stability analyses do not involve sufficiently large volumes of fill to threaten disruption of the River Torrens or the Gorge Road. It is not therefore essential to design a failure-free slope.

2. By keeping the access berm which runs along the foot of the rehabilitated slope, free of further fill, a narrow safety bench is preserved onto which failed slope material may slide.
3. The STABL analysis indicates slope instability at levels higher than what is considered practicable to take a toe-fill. That is, some failure is considered inevitable and should therefore be allowed to take place.
4. Although not the opinion of this Department, the steep, exposed quarry faces which rise above the rehabilitated slope may be considered by some to warrant treatment of some kind. Further placement of fill to cover these faces is not possible as this would lead to very unstable, oversteep slopes. Such faces will need to be treated, if at all, in some other way (eg. spray seeding with lichen, moss etc.)

If the Department for the Environment and Planning consider rehabilitation of the steep quarry faces desirable then Rehabilitation Funds could perhaps be sought to research a suitable process for treating these steep faces, rather than be used for the unwarranted placement of further quarry waste.



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SENIOR GEOLOGIST

REFERENCES

- HOEK, E.; BRAY, J.W. 'Rock Slope Engineering' p. 247-253 1977.
Inst. Min. Metall. LONDON ISBN 0 900488 36 0.
- SIEGEL, R.A. 'STABL-Computer Analysis of General Slope Stability Problems'. Purdue University, West Lafayette, Indiana, June, 1975.

PLATES 1 to 8



Plate 1.

Neg. No. 34467

General view looking North



Plate 2

Neg. NO. 34468

General view looking north-east



Plate 3

Neg. No. 34469

Berm at EL 164



Plate 4

Neg. NO. 34470

Berm at EL 198



Plate 5

Neg. No. 24471

Berm at EL 250



Plate 6

Neg. No. 34472

Above berm EL 198



Plate 7

Neg. No. 34473

West of rehabilitated slope



Plate 8

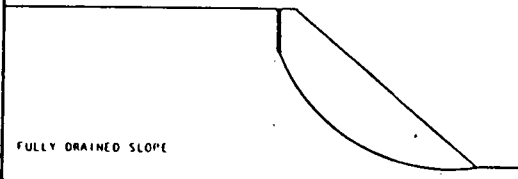
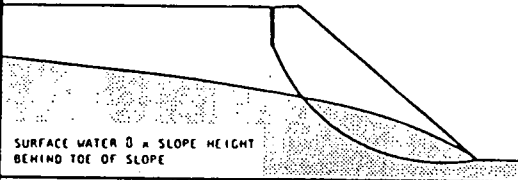
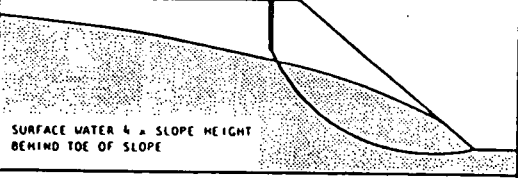
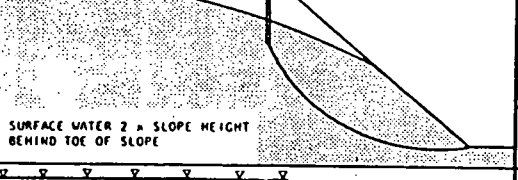

Neg. No. 34474

East of rehabilitated slope

APPENDIX 1

SLOPE STABILITY CALCULATIONS AFTER HOEK AND BRAY

CIRCULAR FAILURE

GROUNDWATER FLOW CONDITIONS	CHART NUMBER
 <p>FULLY DRAINED SLOPE</p>	1
 <p>SURFACE WATER 0.5 SLOPE HEIGHT BEHIND TOE OF SLOPE</p>	2
 <p>SURFACE WATER 1.0 SLOPE HEIGHT BEHIND TOE OF SLOPE</p>	3
 <p>SURFACE WATER 2.0 SLOPE HEIGHT BEHIND TOE OF SLOPE</p>	4
 <p>SATURATED SLOPE SUBJECT TO HEAVY SURFACE RECHARGE</p>	5

(Taken from Hoek and Bray)

The five degrees of groundwater conditions shown in the above figure indicate which of five circular failure charts given in 'Rock Slope Engineering' by E. Hoek and J.W. Bray to choose when using their 'Quick-Analysis' for calculating the factor of safety of a soil mass. Particular care is needed for Chart 2 and Chart 3 when calculating values for $\tan \phi$ for cohesionless soils.

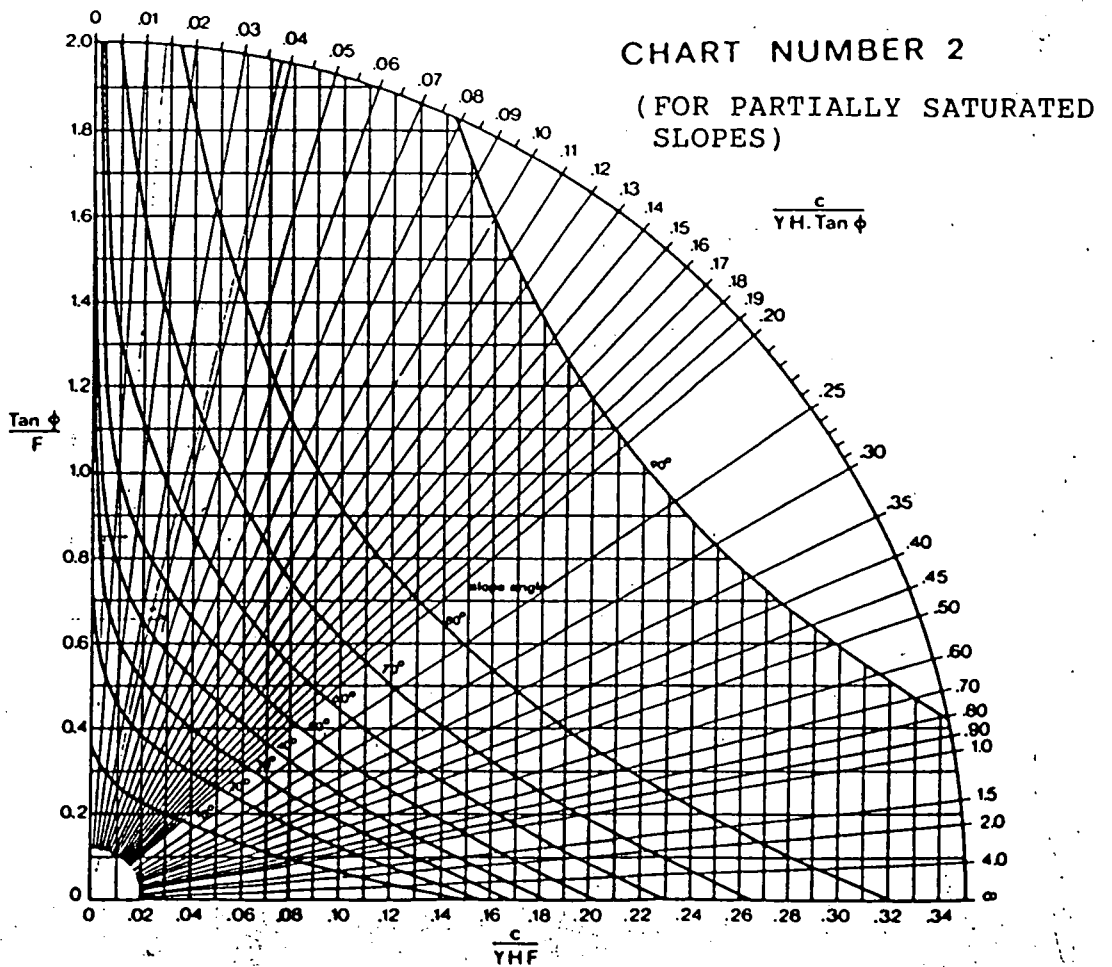
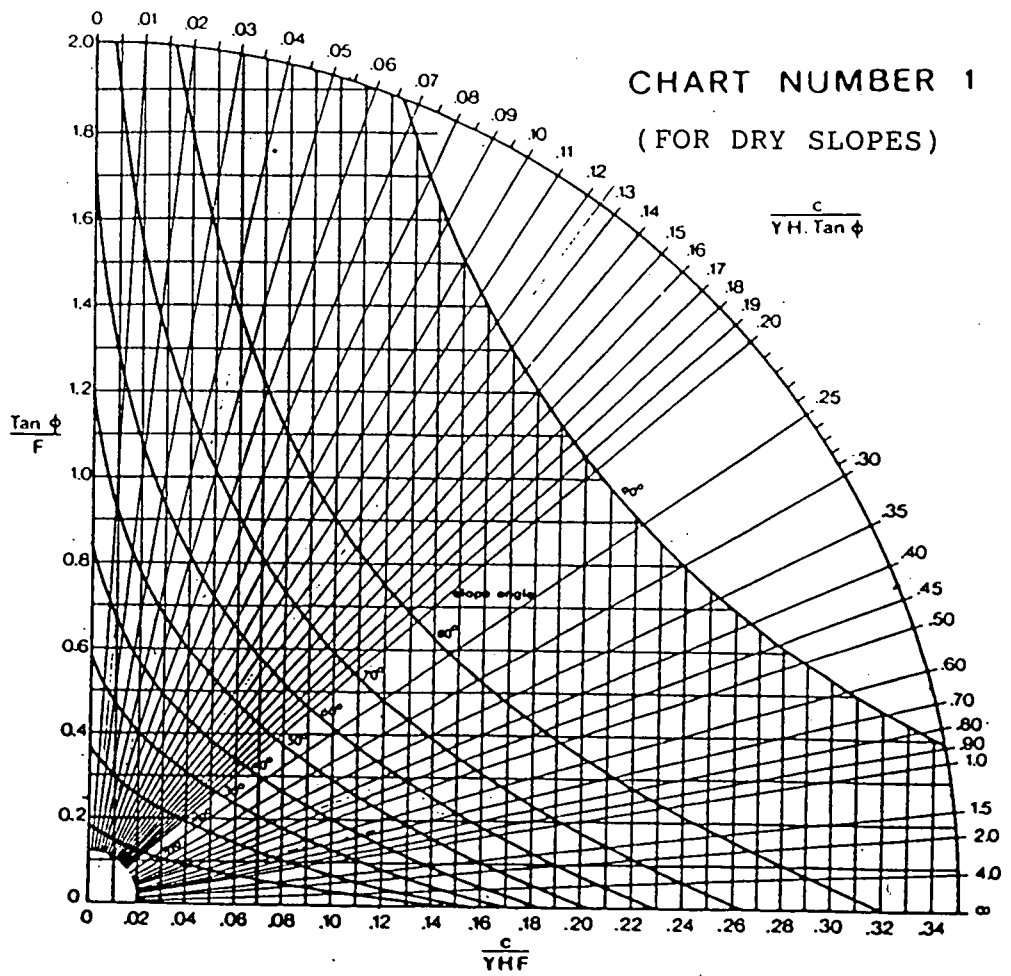


CHART FOR PARTIALLY SATURATED SLOPES

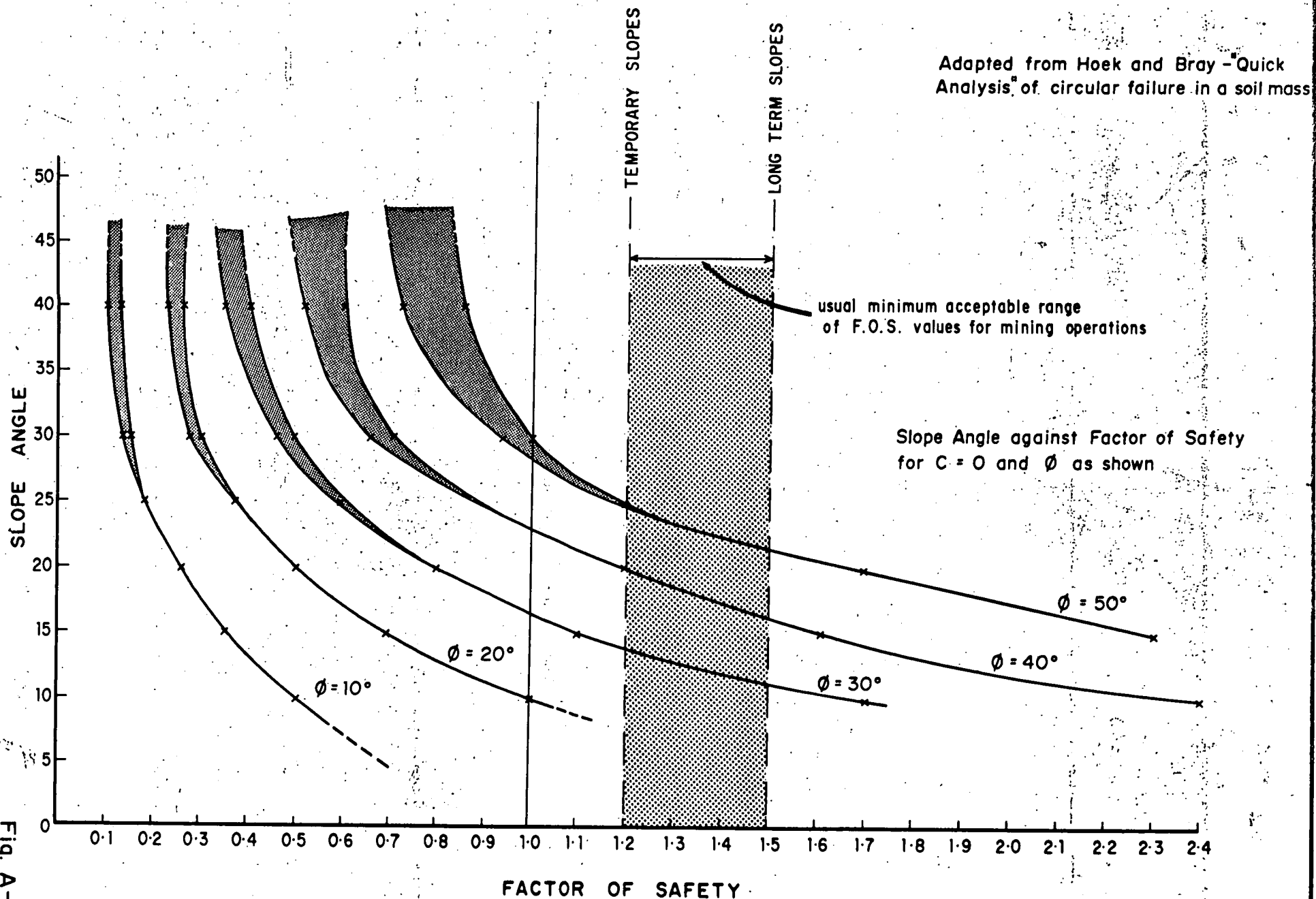
SEC. 5394 HD. YATALA


RIVERVIEW QUARRY - REHABILITATED SLOPE STABILITY ANALYSIS

DEPARTMENT OF MINES AND ENERGY
SOUTH AUSTRALIA

COMPILED J.B.	DRAWN M.B.	DATE Sept '85	CHECKED
SCALE C.D.O.	PLAN NUMBER S18381	DATE 16/11/85	

Fig. A-1





DEPARTMENT OF MINES AND ENERGY
SOUTH AUSTRALIA

RIVERVIEW QUARRY - REHABILITATED SLOPE STABILITY ANALYSIS

SEC. 5394 HD. YATALA

CHART FOR DRY SLOPES

COMPILED J. B.	DRAWN M. B.	DATE Sept 85	CHECKED
SCALE C.D.O.		PLAN NUMBER S18382	

Fig. A-2

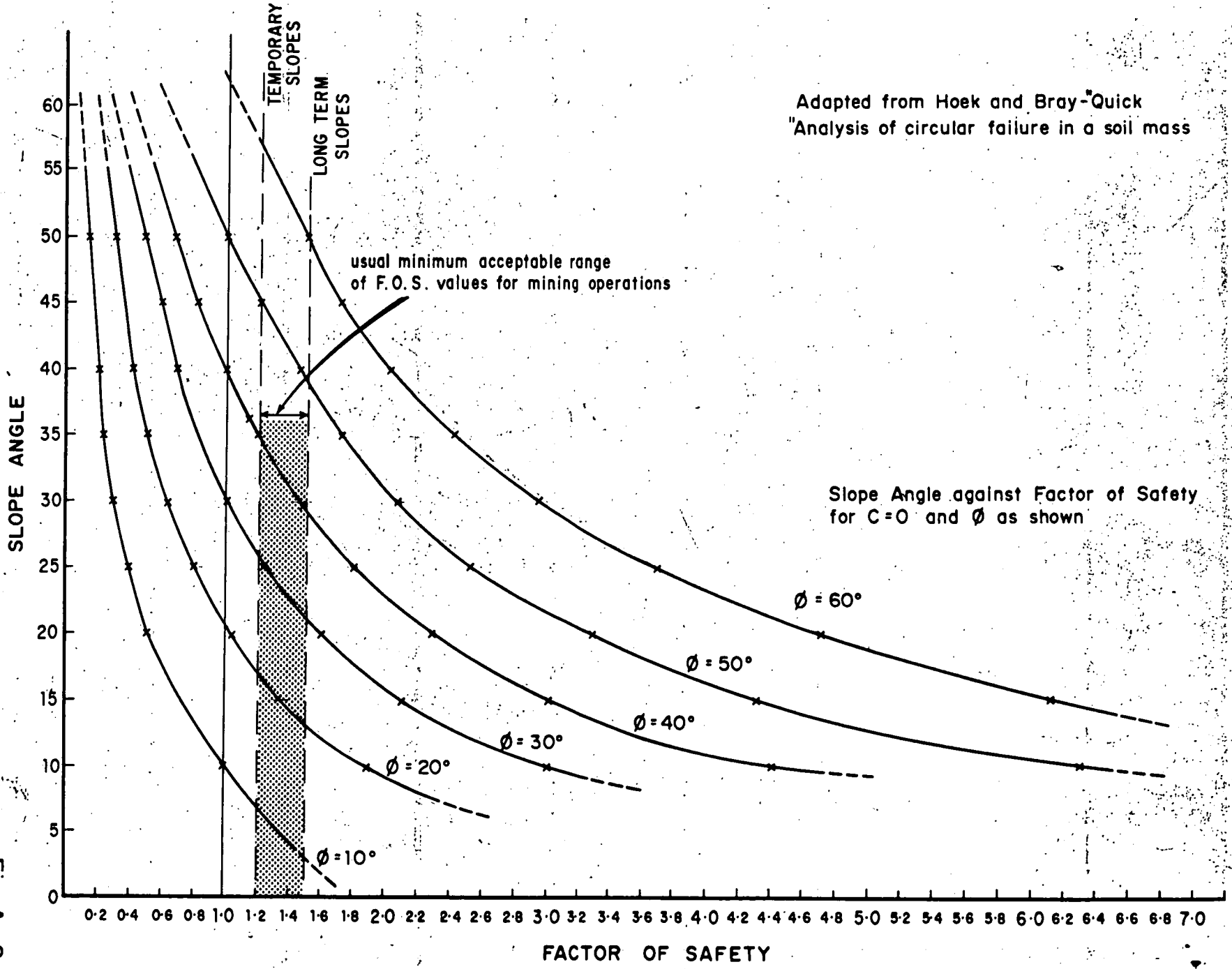


TABLE 1 QUICK ANALYSIS BY HOEK & BRAY

	CHART	SLOPE ANGLE	TAN ϕ	C	γ	H	$\frac{C}{H \cdot \tan \phi}$	$\frac{\tan \phi}{F}$	F
1	2 (Partially Saturated)	35°	40°=0.84	Nil				1.5	<1
2	2	33°	35°=0.7	Nil				1.2	<1
3	2	35°	38°=0.78	Nil				1.3	<1
4	2	30°	40°=0.84	Nil				1.4	<1
5	2	25°	38°=0.78	Nil				1.0	<1
6	2	25°	40°=0.84	Nil				0.84	1.0
7	2	23°	38°=0.78	Nil				0.78	1.0
8	1 (Dry)	33°	35°=0.7	Nil				0.7	1.0
9	1	35°	38°=0.78	Nil				0.7	1.1
10	1	30°	40°=0.84	Nil				0.7	1.2
11	2	30°	38°=0.78	(50lb/ft ²) 2.5 KPa	(100lb/ft ³) 16 KN/M ³	(60ft) 20m	0.010	0.53	
12	1	30°	38°=0.78	2.5 KPa	(80lb/ft ³)	20m	0.012	0.50	
13	1	25°	38°=0.78	Nil				0.48	

Co-ordinates for centre of
critical circle: $x = 0.1H = 2M$
 $y = 1.5H = 30M$

APPENDIX 2

SLOPE STABILITY CALCULATIONS USING STABL

DATE : July '25

(Taken from 'ROCK SLOPE ENGINEERING'; E Hoek, J W Bray, p247-253)

L	20	<p>a) Section through sliding mass showing slice boundaries and geometrical parameters</p>	<p>b) Calculation of average water pressure u on base of slice</p>	<p>c) Slice parameters used in the stability analysis</p>	<p>SYMBOLS</p> <p>f_0 correction factor</p> <p>a geometrical functions</p> <p>c cohesive strength</p> <p>ϕ angle of friction</p> <p>p average weight per unit width of slice</p> <p>u average water pressure on base of slice</p> <p>L chord length of failure surface</p> <p>d depth of failure surface</p>	<p>Calculation of FACTOR OF SAFETY</p> $F = f_0 \frac{\sum (c + (p-u) \tan \phi) \Delta s}{\sum \Delta W \tan \alpha + Q}$ <p>If Q is zero and Δs is constant</p> $F = f_0 \frac{\sum (c + (p-u) \tan \phi)}{\sum p \cdot \tan \alpha}$ <div>$F.O.S = \frac{f_0 \sum \frac{c}{a}}{\sum \Delta W \cdot \tan \alpha}$</div>
d	12.5					
$\frac{d}{L}$	0.156					
f_0 from chart	1.051					
$\gamma_{\text{material 1}}$	13.4 kN/m ³	<p>DESCRIPTION OF MATERIAL & SLOPE : $\text{Slope angle} = 35^\circ$; $c = 0$; $\phi = 35^\circ$ Quarry waste</p>				
$\gamma_{\text{material 2}}$	16.4 kN/m ³					
γ_w	9.82 kN/m ³					

Note : $1 \text{ lb/ft}^3 = 0.16 \text{ kN/m}^3$, $1 \text{ psi} = 6.89 \text{ kPa}$, $1 \text{ lb/ft}^2 = 0.04788 \text{ kPa}$

SLICE	α	$\tan \alpha$	h_m	h_u	u $u = h_u \tan \alpha$	δ_m	p $p = \delta_m h_m$	Δx	ΔW $\Delta W = p \Delta x$	$p-u$	$\Delta W \cdot \tan \alpha$	c	ϕ	$\tan \phi$	X $X = (c + (p-u) \tan \phi) \Delta x$	TRIAL 1 (F = 1)		TRIAL 2 (F = 1)		TRIAL 3 (F = 1)	
																n_u use chart	$\frac{X}{n_u}$	n_u	$\frac{X}{n_u}$	n_u	$\frac{X}{n_u}$
1	65°	2.14	3	0.24		13		5	39	22.4	417.3	0	35°	0.7	$(50.7) \times 39 = 136.5$	0.46	297				
2	56°	1.49	4	"		"		"	52	"	309.4	"	"	"	$3.5 \times 52 = 182$	0.75	243				
3	56°	1.49	5	"		"		"	65	"	386.2	"	"	"	" $\times p$ 228	0.75	304				
4	56°	1.49	6	"		"		"	72	"	464.1	"	"	"	" 273	0.75	205				
5	40°	0.84	6.5	"		"		"	84.5	"	354.9	"	"	"	" 296	0.93	318				
6	36°	0.58	5.2	"		"		"	75.4	"	218.7	"	"	"	" 264	1.05	251				
7	22°	0.40	5.4	"		"		"	70.2	"	140.4	"	"	"	" 246	1.1	224				
8	15°	0.27	4.6	"		"		"	57.8	"	90.7	"	"	"	" 209	1.1	170				
9	12°	0.32	4.2	"		"		"	54.6	"	87.36	"	"	"	" 191	1.1	174				
10	7°	0.31	3.5	"		"		"	45.5	"	70.5	"	"	"	" 159	1.1	145				
11	10°	0.18	3	"		"		"	39	"	35.1	"	"	"	" 137	1.08	127				
12	6°	0.11	2.7	"		"		"	35.1	"	19.3	"	"	"	" 123	1.07	115				
13	4°	0.07	1.7	"		"		"	22.1	"	7.7	"	"	"	" 97	1.05	73				
14	-30°	0.52	1.2	"		"		"	15.6	"	45.2	"	"	"	" 55	0.44	125				

water table @ water table @

Mod. fixed for water table (B)

Mod. fixed for water table (B)

For details of calculating for water-tables (B) & (C) See working for No. 85/19

		(A)	(B)
TRIAL	1	2	3
$\frac{\tan \phi}{F}$	0.7	0.7	0.7
F. O. S.	1.12	1.06	0.99

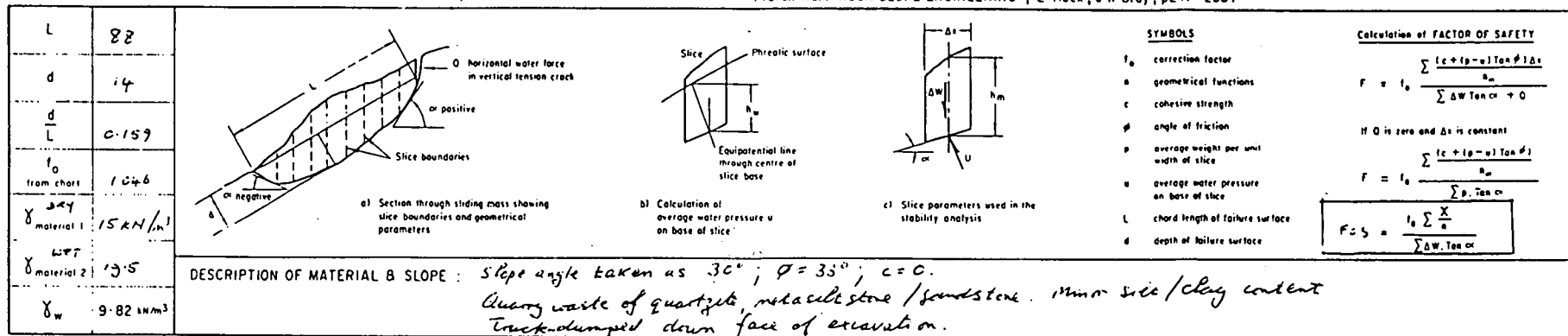
Final FACTOR OF SAFETY : 1.0

PROJECT: REAYMIT RIVERVIEW QU.
REHABILITATION SLOPE
CALCULATED: J. BEAL DATE: JULY '85

GROUNDWATER AND ENGINEERING BRANCH

SLOPE STABILITY ASSESSMENT USING JANBU'S NON-CIRCULAR FAILURE ANALYSIS

(Taken from 'ROCK SLOPE ENGINEERING'; E. Hoek, J.W. Bray, p247-253)



Note: $1 \text{ lb}/11^3 = 0.16 \text{ kN}/\text{m}^3$

SLICE	α	$\tan \alpha$	h_m	h_u	u $u = h_u \gamma_w$	γ_m	p $p = \gamma_m h_m$	Δx	ΔW $\Delta W = p \Delta x$	$p-u$	$\Delta W \cdot \tan \alpha$	c	ϕ	$\tan \phi$	X $X = (c + (p-u) \tan \phi) \Delta x$	TRIAL 1 (F=1)		TRIAL 2 (F=1.3)		TRIAL 3 (F=)	
																n_m use chart	$\frac{X}{n_m}$	n_m	$\frac{X}{n_m}$	n_m	$\frac{X}{n_m}$
1	65°	2.14	8			15	120	5	600	120	1200	0	33°	0.65	$(0.65 \cdot 5) \times 120 = 390$	0.42	929				
2	50°	1.19	10			15	150	5	750	150	893	0	"	"	$3.25 \times 150 = 488$	0.72	678				
3	50°	1.19	12.5			15	127.5	5	938	127.5	1116	0	"	"	$" \times 128 = 611$	0.72	809				
4	50°	1.19	16			15	240	5	1200	240	1428	0	"	"	$" \times 240 = 720$	0.72	1023				
5	40°	0.84	16.3			15	245	5	1223	245	1027	0	"	"	$" \times 245 = 796$	0.70	384				
6	30°	0.58	15			15	225	5	1125	225	653	0	"	"	$" \times 225 = 731$	1.02	717				
7	22°	0.40	4			15	210	5	1050	210	420	0	"	"	$" \times 210 = 623$	1.08	632				
8	15°	0.27	12	0.5	4.9	15	180	5	900	175	243	0	"	"	$" \times 175 = 589$	1.09	522				
9	12°	0.52	10	1.3	12.8	15	150	5	750	137	240	0	"	"	$" \times 137 = 445$	1.09	408				
10	17°	0.31	10	3.0	29.5	16	160	5	800	131	248	0	"	"	$" \times 131 = 426$	1.09	391				
11	10°	0.17	11	3.5	34.4	16	176	5	880	142	158	0	"	"	$" \times 142 = 462$	1.07	432				
12	6°	0.11	12.5	4.0	39.3	16	200	5	1000	161	110	0	"	"	$" \times 161 = 523$	1.06	493				
13	2°	0.03	12.5	3.8	37.3	16	200	5	1000	163	30	0	"	"	$" \times 163 = 530$	1.02	520				
14	-4°	0.07	10	3.5	34.4	17	170	5	850	136	60	0	"	"	$" \times 136 = 442$	0.94	470				
15	-4°	0.25	7	2.5	24.6	17	119	5	595	94	149	0	"	"	$" \times 94 = 306$	0.79	387				
16	-25°	0.47	2.5	0.8	7.9	16	40	5	200	32	94	0	"	"	$" \times 32 = 104$	0.58	179				
															8153	$\sum \Delta W \cdot \tan \alpha$		$\sum \frac{X}{n_m}$		9516	
																$\times F_0$		$\times F_0$		10131	

TRIAL	1	2	3
$\frac{\tan \phi}{F}$	0.65	0.5	
F.O.S.	1.23	1.3	

Final FACTOR OF SAFETY : 1.3

RVI. DAT

PROFILE READY MIX SLOPE STABILITY ANALYSIS

12 7
 47.0 0.0 90.0 66.0 1
 90.0 66.0 115.0 62.0 2
 115.0 62.0 188.0 103.0 3
 188.0 103.0 250.0 103.0 3
 250.0 103.0 452.0 248.0 2
 452.0 248.0 486.0 267.0 1
 486.0 267.0 600.0 267.0 1
 315.0 52.0 452.0 248.0 1
 200.0 69.0 250.0 103.0 2
 115.0 62.0 200.0 69.0 2
 90.0 66.0 112.0 52.0 1
 112.0 52.0 315.0 52.0 1

* SOIL

3
 25.0 26.0 10000.0 60.0 0.0 0.0 1
 13.0 17.5 0.0 33.0 0.0 0.0 1
 14.5 18.5 0.0 33.0 0.0 0.0 1

WATER

1 9.9

3

47.0 0.0

95.0 59.0

600.0 150.0

CIRCLE

3 5 95.0 250.0 315.0 600.0 0.0 20.0 0.0 0.0

SOIL TYPE	MOIST Wt. kN/m ³	Satur. wt.	C	ϕ	pore pressure param.	pore pressure const.	piezometric surface no.
1	25	26	0	60	—	—	1
2	13	17.5	0	33	—	—	1
3	14.5	18.5	0	33	—	—	1

KVI (Example)

FOLLOWING ARE DISPLAYED THE TEN MOST CRITICAL OF THE TRIAL FAILURE SURFACES EXAMINED. THEY ARE ORDERED - MOST CRITICAL FIRST.

FAILURE SURFACE SPECIFIED BY 6 COORDINATE POINTS

POINT NO.	X-SURF (M.)	Y-SURF (M.)
1	260.00	103.00
2	278.65	111.91
3	305.15	125.94
4	328.60	144.68
5	348.17	167.41
6	354.18	177.78

*** 0.956 ***

[circle described by 1 on plot]

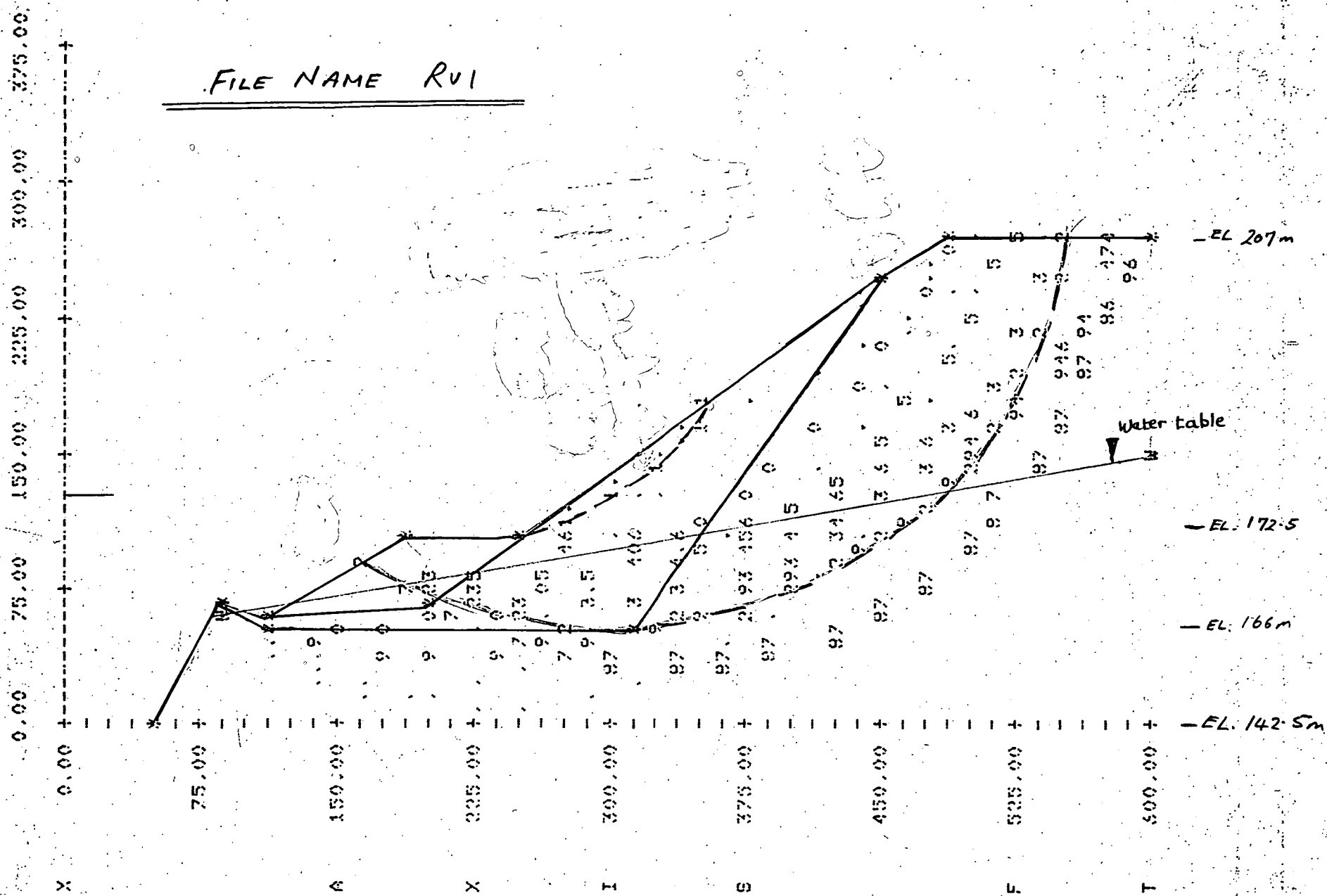
FAILURE SURFACE SPECIFIED BY 19 COORDINATE POINTS

POINT NO.	X-SURF (M.)	Y-SURF (M.)
1	167.50	91.49
2	193.94	77.34
3	221.91	66.44
4	250.94	58.88
5	280.66	54.79
6	310.65	54.23
7	340.31	57.19
8	369.91	63.64
9	398.14	73.50
10	426.12	86.61
11	450.37	102.91
12	473.54	121.87
13	494.31	143.52
14	512.39	167.44
15	527.53	193.36
16	539.52	220.94
17	548.19	249.58
18	551.27	267.00

*** 15.089 ***

[circle described by 2 on plot.]

FILE NAME RV1



RV2.DAT

PROFILE

READYMIX SLOPE STABILITY ANALYSIS

9 6
 47.0 0.0 90.0 66.0 1
 90.0 66.0 115.0 62.0 2
 115.0 62.0 200.0 69.0 2
 200.0 69.0 452.0 248.0 2
 452.0 248.0 486.0 267.0 1
 486.0 267.0 600.0 267.0 1
 315.0 52.0 452.0 248.0 1
 90.0 66.0 112.0 52.0 1
 112.0 52.0 315.0 52.0 1

*SOIL

3
 25.0 26.0 10000.0 60.0 0.0 0.0 1
 13.0 17.5 0.0 33.0 0.0 0.0 1
 14.5 18.5 0.0 33.0 0.0 0.0 1

WATER

1 9.8

3

47.0 0.0
 95.0 60.0
 600.0 108.0

CIRCLE

3 5 115.0 280.0 250.0 460.0 0.0 30.0 0.0 0.0

*

SOIL TYPE	moist wt. kN/m ³	sat. wt	c	ϕ	pore pressure param.	pore pressure constant	piezometric surface number
1	25	26	0	60	—	—	1
2	13	17.5	0	33	—	—	1
3	14.5	18.5	0	33	—	—	1

RV2 (Example)

FOLLOWING ARE DISPLAYED THE TEN MOST CRITICAL OF THE TRIAL FAILURE SURFACES EXAMINED. THEY ARE ORDERED - MOST CRITICAL FIRST.

FAILURE SURFACE SPECIFIED BY 8 COORDINATE POINTS

POINT NO.	X-SURF (M.)	Y-SURF (M.)
1	280.00	125.83
2	309.39	131.85
3	337.61	142.03
4	364.08	156.15
5	398.24	173.93
6	409.61	194.98
7	427.74	218.89
8	438.54	238.44

0.984 ***

[circle described by 1 on plot]

FAILURE SURFACE SPECIFIED BY 3 COORDINATE POINTS

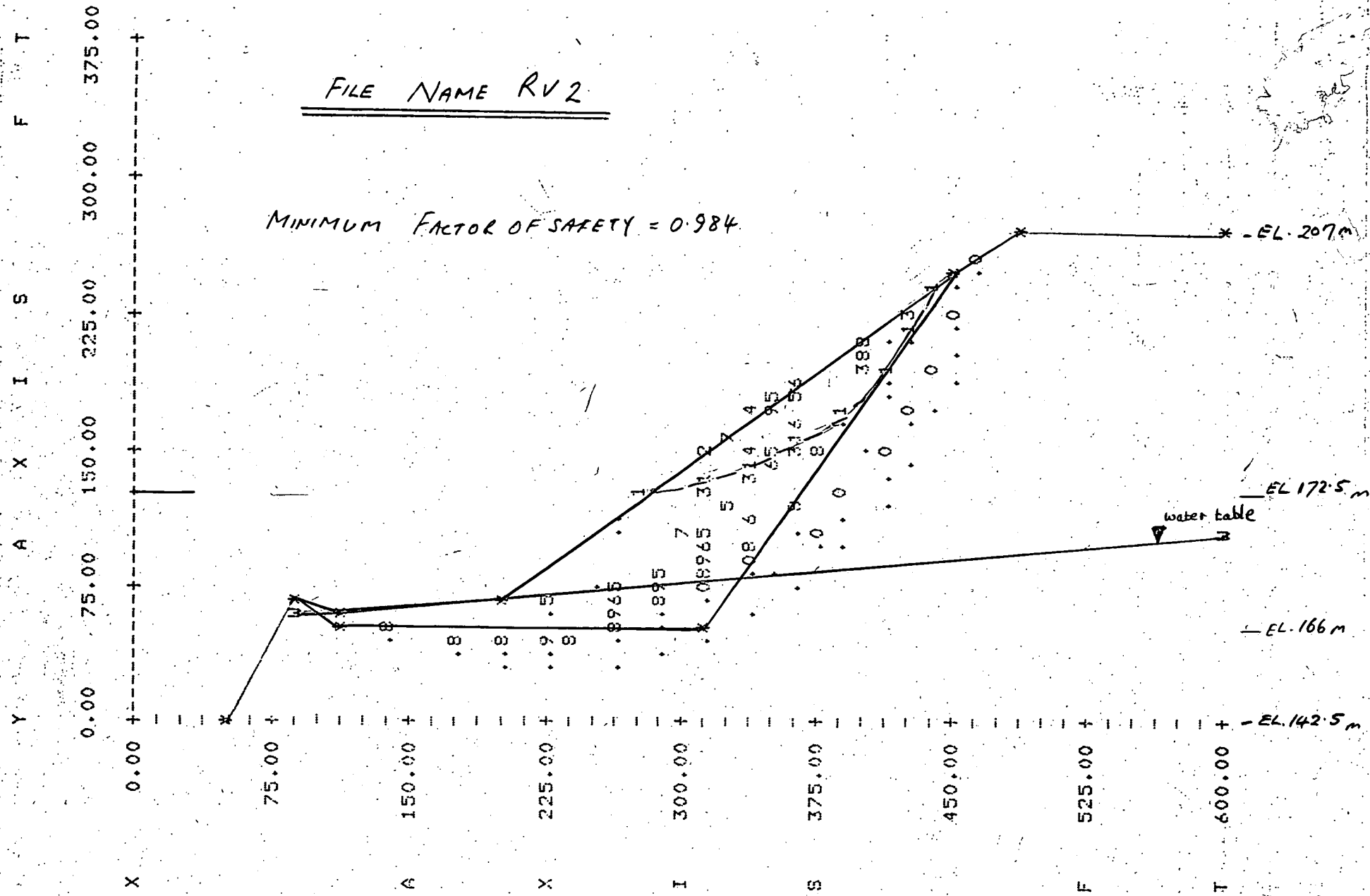
POINT NO.	X-SURF (M.)	Y-SURF (M.)
1	280.00	125.83
2	307.19	138.50
3	317.23	152.27

1.000 ***

[circle described by 2 on plot]

FILE NAME RV2

MINIMUM FACTOR OF SAFETY = 0.984



RV3.DAT

PROFILE

READYMIX SLOPE STABILITY ANALYSIS

9 6

47.0 0.0 90.0 66.0 1
 90.0 66.0 115.0 62.0 2
 115.0 62.0 200.0 69.0 2
 200.0 69.0 452.0 248.0 2
 452.0 248.0 486.0 267.0 1
 486.0 267.0 600.0 267.0 1
 315.0 52.0 452.0 248.0 1
 90.0 66.0 112.0 52.0 1
 112.0 52.0 315.0 52.0 1

* SOIL

3
 25.0 26.0 10000.0 60.0 0.0 0.0 1
 13.0 17.5 0.0 33.0 0.0 0.0 1
 14.5 18.5 0.0 33.0 0.0 0.0 1

CIRCLE

3 5 115.0 280.0 250.0 460.0 0.0 30.0 0.0 0.0

*

SOIL TYPE	Moist Wt. KN/m ³	Sat. Wt.	c	ϕ	Pore press	Pore const.	Piez. surface No.
1	25	26	0	60	—	—	1
2	13	17.5	0	33	—	—	1
3	14.5	18.5	0	33	—	—	1

RV3 (Example)

FOLLOWING ARE DISPLAYED THE TEN MOST CRITICAL OF THE TRIAL FAILURE SURFACES EXAMINED. THEY ARE ORDERED - MOST CRITICAL FIRST.

FAILURE SURFACE SPECIFIED BY 8 COORDINATE POINTS

POINT NO.	X-SURF (M.)	Y-SURF (M.)
1	280.00	125.83
2	309.39	131.85
3	337.61	142.03
4	364.08	156.15
5	388.24	173.93
6	409.61	194.98
7	427.74	218.89
8	438.54	238.44

*** 0.984 ***

FAILURE SURFACE SPECIFIED BY 3 COORDINATE POINTS

POINT NO.	X-SURF (M.)	Y-SURF (M.)
1	280.00	125.83
2	307.19	138.50
3	317.23	152.27

*** 1.000 ***

DATE : JULY '85

(Taken from 'ROCK SLOPE ENGINEERING'; E Hoek, J W Bray, p247-253)

Note : $1 \text{ lb/ft}^3 = 0.16 \text{ kN/m}^3$, $1 \text{ psi} = 6.89 \text{ kPa}$, $1 \text{ lb f/ft}^2 = 0.04788 \text{ kPa}$

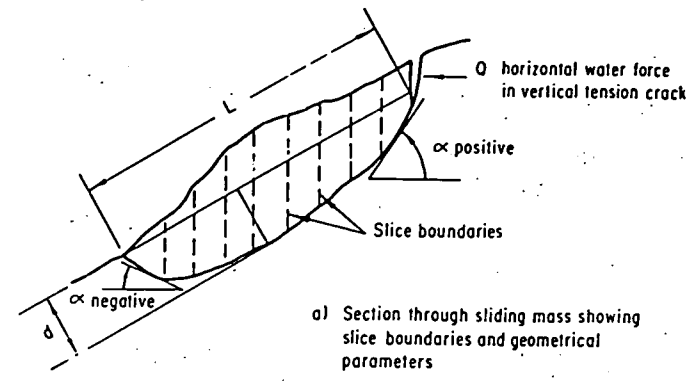
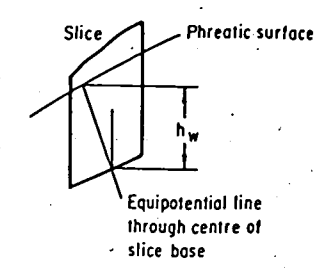
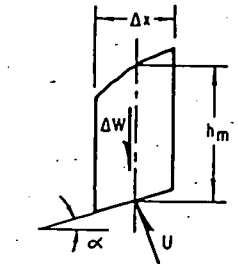
		(A)	(B)
TRIAL	1	2	3
$\frac{\tan \phi}{F}$	0.7	0.7	0.7
F. O. S.	1.12	1.06	0.99

SHEET 1 OF 2

PROJECT: READYMIT RIVERVIEW QU.
REHABILITATION SLOPE
 CALCULATED: J. BEAL DATE: JULY '85

SLOPE STABILITY ASSESSMENT USING JANBU'S NON-CIRCULAR FAILURE ANALYSIS

(Taken from 'ROCK SLOPE ENGINEERING'; E Hoek, J W Bray, p247-253)

L	88	 <p>a) Section through sliding mass showing slice boundaries and geometrical parameters</p>	 <p>b) Calculation of average water pressure u on base of slice</p>	 <p>c) Slice parameters used in the stability analysis</p>	SYMBOLS	Calculation of FACTOR OF SAFETY	
d	14				f_0	correction factor	$F = f_0 \frac{\sum (c + (p-u) \tan \phi) \Delta x}{\sum \Delta W \tan \alpha + Q}$
$\frac{d}{L}$	0.159				n	geometrical functions	
f_0 from chart	1.046				c	cohesive strength	<p>If Q is zero and Δx is constant</p> $F = f_0 \frac{\sum (c + (p-u) \tan \phi)}{\sum p \cdot \tan \alpha}$
$\gamma_{\text{material 1}}$	15 kN/m ³				ϕ	angle of friction	
γ_{WET}	19.5				p	average weight per unit width of slice	$F.O.S. = \frac{f_0 \sum \frac{X}{n}}{\sum \Delta W \cdot \tan \alpha}$
γ_w	9.82 kN/m ³	u	average water pressure on base of slice				
DESCRIPTION OF MATERIAL & SLOPE : <u>Slope angle taken as 30°; $\phi = 33^\circ$; $c = 0$.</u> <u>Quarry waste of quartzite, metacilt stone / sandstone. minor silt / clay content.</u> <u>Truck-dumped down face of excavation.</u>							

Note : 1 lb/ft³ = 0.16 kN/m³

SLICE	α	$\tan \alpha$	h_m	h_w	u $u = h_w \gamma_w$	γ_m	p $p = \gamma_m h_m$	Δx	ΔW $\Delta W = p \Delta x$	p - u	$\Delta W \cdot \tan \alpha$	c	ϕ	$\tan \phi$	X $X = (c + (p-u) \tan \phi) \Delta x$	TRIAL 1 (F=1)		TRIAL 2 (F=1.3)		TRIAL 3 (F=1.3)	
																n_α use chart	$\frac{X}{n_\alpha}$	n_α	$\frac{X}{n_\alpha}$	n_α	$\frac{X}{n_\alpha}$
1	65°	2.14	8			15	120	5	600	120	1284	0	33°	0.65	$(0.65 \times 5) \times 120 = 390$	0.42	929				
2	50°	1.19	10			15	150	5	750	150	893	0	"	"	$3.25 \times 150 = 488$	0.72	678				
3	50°	1.19	12.5			15	187.5	5	938	187.5	1116	0	"	"	" $\times 188 = 611$	0.72	849				
4	50°	1.19	16			15	240	5	1200	240	1428	0	"	"	" $\times 240 = 780$	0.72	1083				
5	40°	0.84	16.3			15	245	5	1223	245	1027	0	"	"	" $\times 245 = 796$	0.90	884				
6	30°	0.58	15			15	225	5	1125	225	653	0	"	"	" $\times 225 = 731$	1.02	717				
7	22°	0.40	14			15	210	5	1050	210	420	0	"	"	" $\times 210 = 683$	1.08	632				
8	15°	0.27	12	0.5	4.9	15	180	5	900	175	243	0	"	"	" $\times 175 = 569$	1.09	522				
9	18°	0.32	10	1.3	12.8	15	150	5	750	137	240	0	"	"	" $\times 137 = 445$	1.09	408				
10	17°	0.31	10	3.0	29.5	16	160	5	800	131	248	0	"	"	" $\times 131 = 426$	1.09	391				
11	10°	0.18	11	3.5	34.4	16	176	5	880	142	158	0	"	"	" $\times 142 = 462$	1.07	432				
12	6°	0.11	12.5	4.0	39.3	16	200	5	1000	161	110	0	"	"	" $\times 161 = 523$	1.06	493				
13	2°	0.03	12.5	3.8	37.3	16	200	5	1000	163	30	0	"	"	" $\times 163 = 530$	1.02	520				
14	-4°	0.07	10	3.5	34.4	17	170	5	850	136	60	0	"	"	" $\times 136 = 442$	0.94	470				
15	-14°	0.25	7	2.5	24.6	17	119	5	595	94	149	0	"	"	" $\times 94 = 306$	0.79	387				
16	-25°	0.47	2.5	0.8	7.9	16	40	5	200	32	94	0	"	"	" $\times 32 = 104$	0.58	179				
											8153	$\leftarrow \sum \Delta W \cdot \tan \alpha$			$\sum \frac{X}{n_\alpha}$	9514		10131			

TRIAL	1	2	3
$\frac{\tan \phi}{F}$	0.65	0.5	
F.O.S.	1.23	1.3	

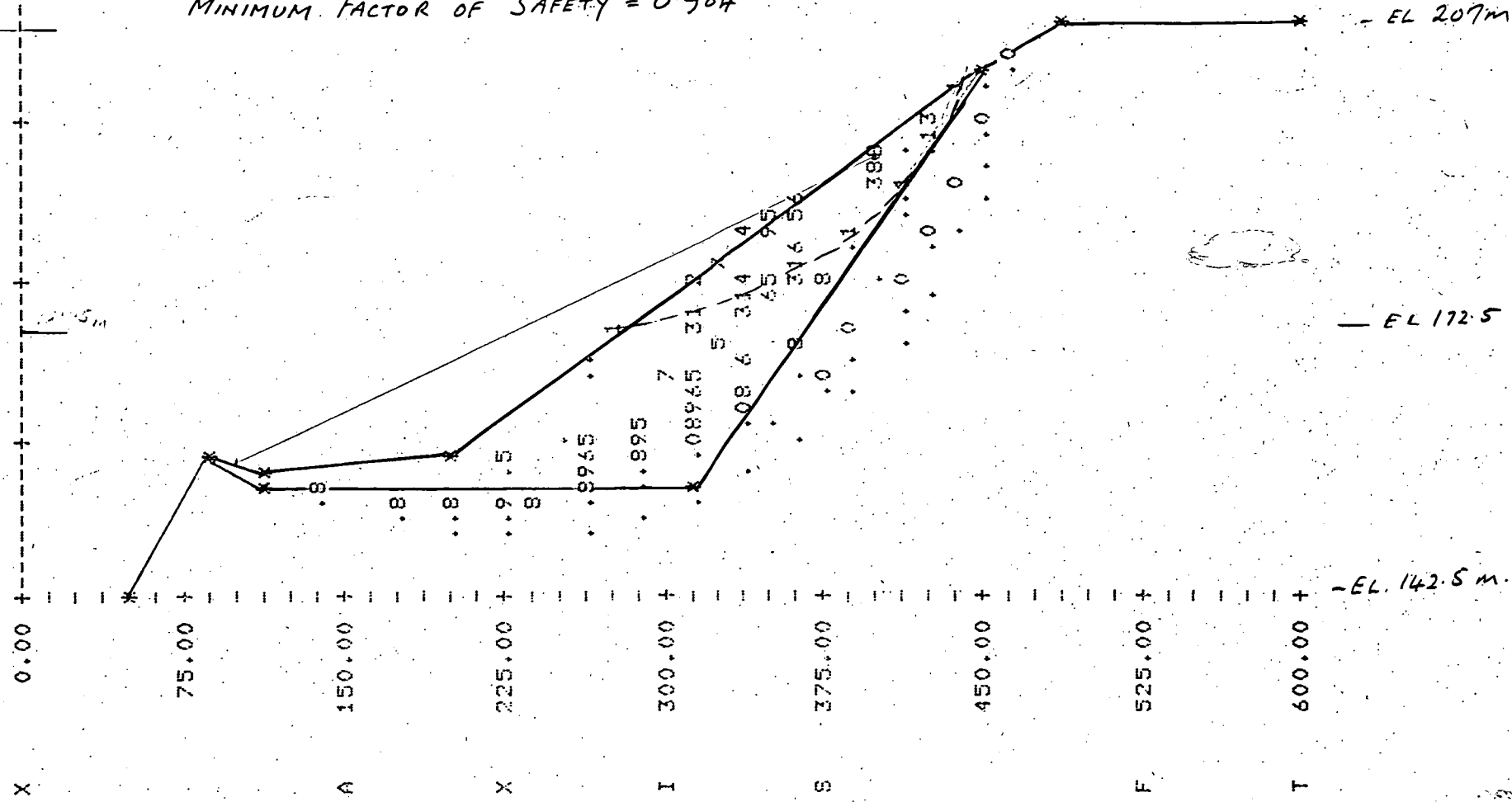
Final FACTOR OF SAFETY : 1.3

T
 F
 S
 I
 X
 A
 Y

25m = 75. Scale from: 2m

FILE NAME RV3...

MINIMUM FACTOR OF SAFETY = 0.984



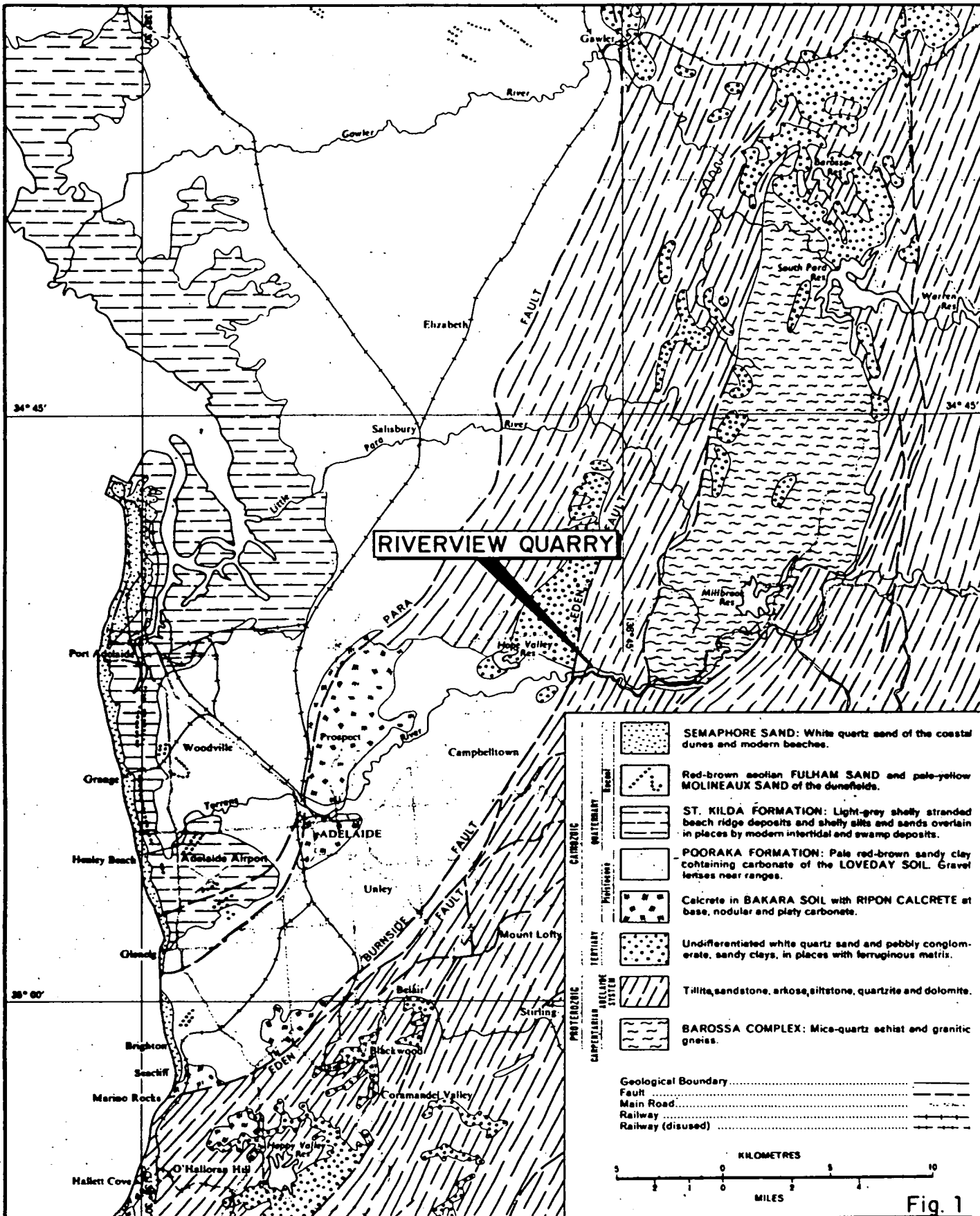
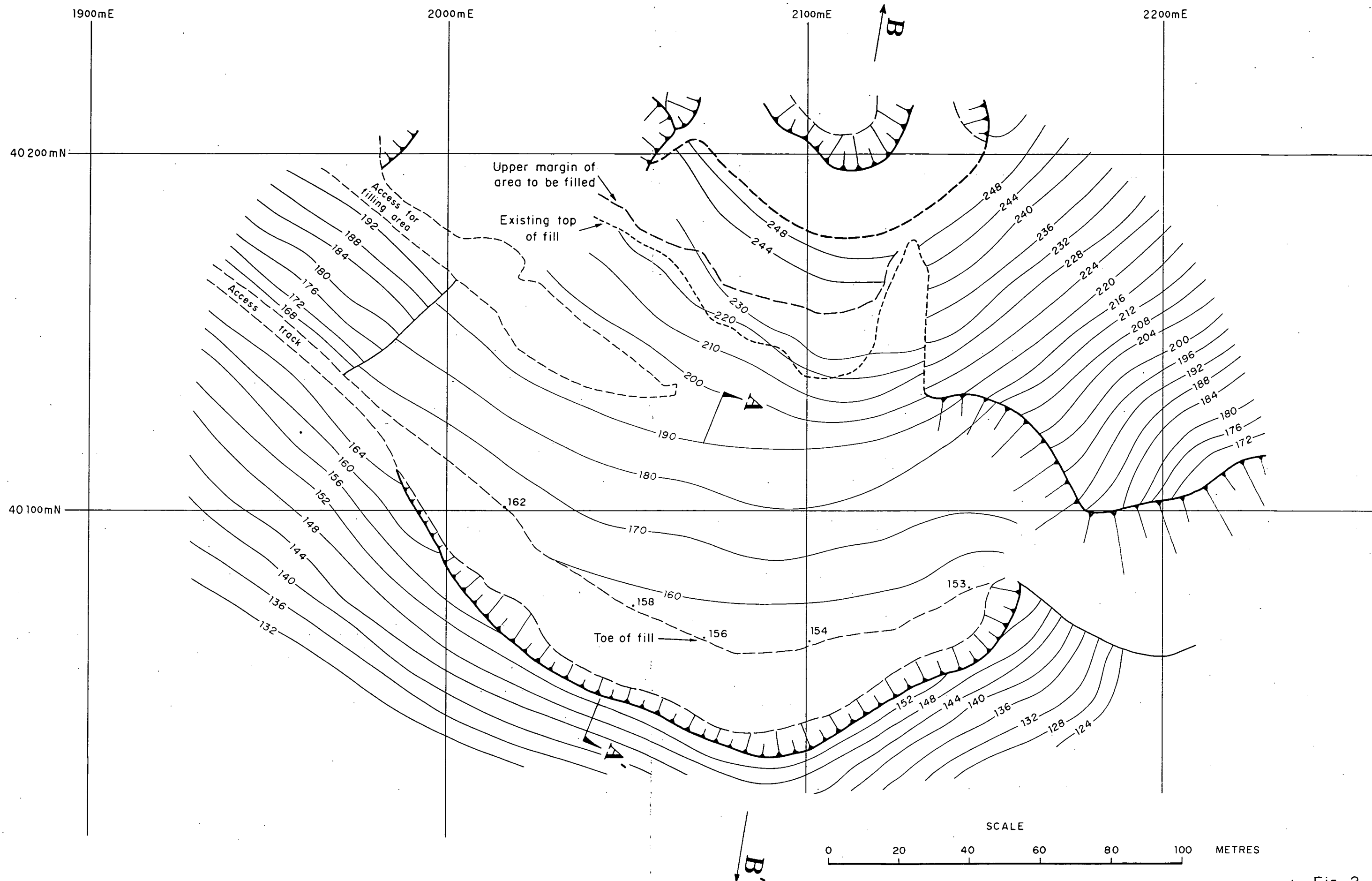



Fig. 1

COMPILED J.B.	C.D.O. DATE
DRAWN M.B.	SCALE 1:250 000
DATE July '85	PLAN NUMBER
CHECKED	S18289



For Sections, see Plans 85-377, 85-378

Fig. 2

 DEPARTMENT OF MINES AND ENERGY SOUTH AUSTRALIA	COMPILED J. B.	<i>UR</i> 18.11.85 C.D.O. DATE
	DRAWN M. B.	SCALE 1:1000
	DATE July '85	PLAN NUMBER
	CHECKED	85-376
RIVERVIEW QUARRY - REHABILITATED SLOPE STABILITY ANALYSIS SEC. 5394 HD. YATALA CONTOUR PLAN BASED ON SURTECH PTY LTD PLAN		

B

B'

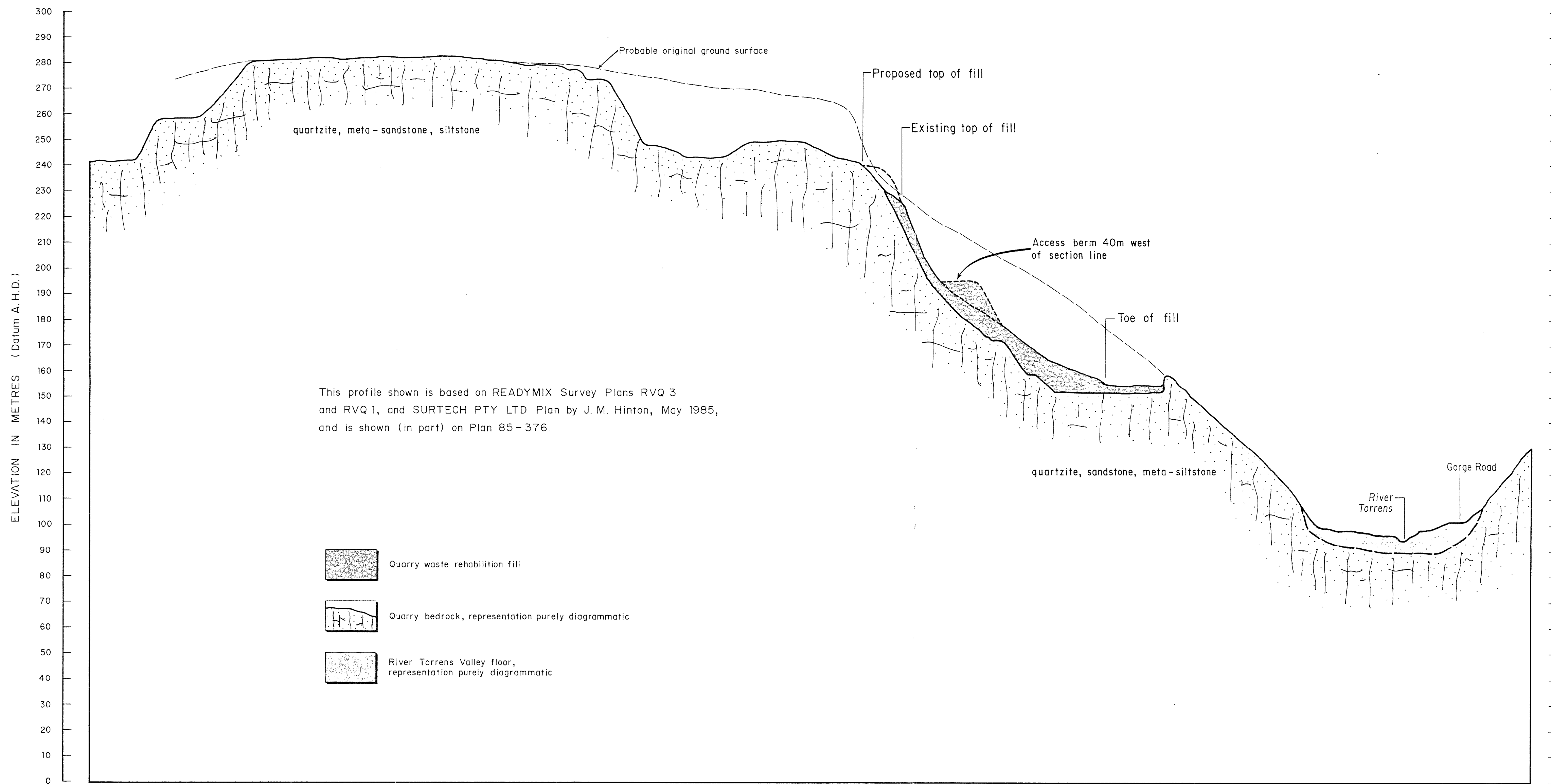



Fig. 3

For location, see Plan 85-376

SCALE

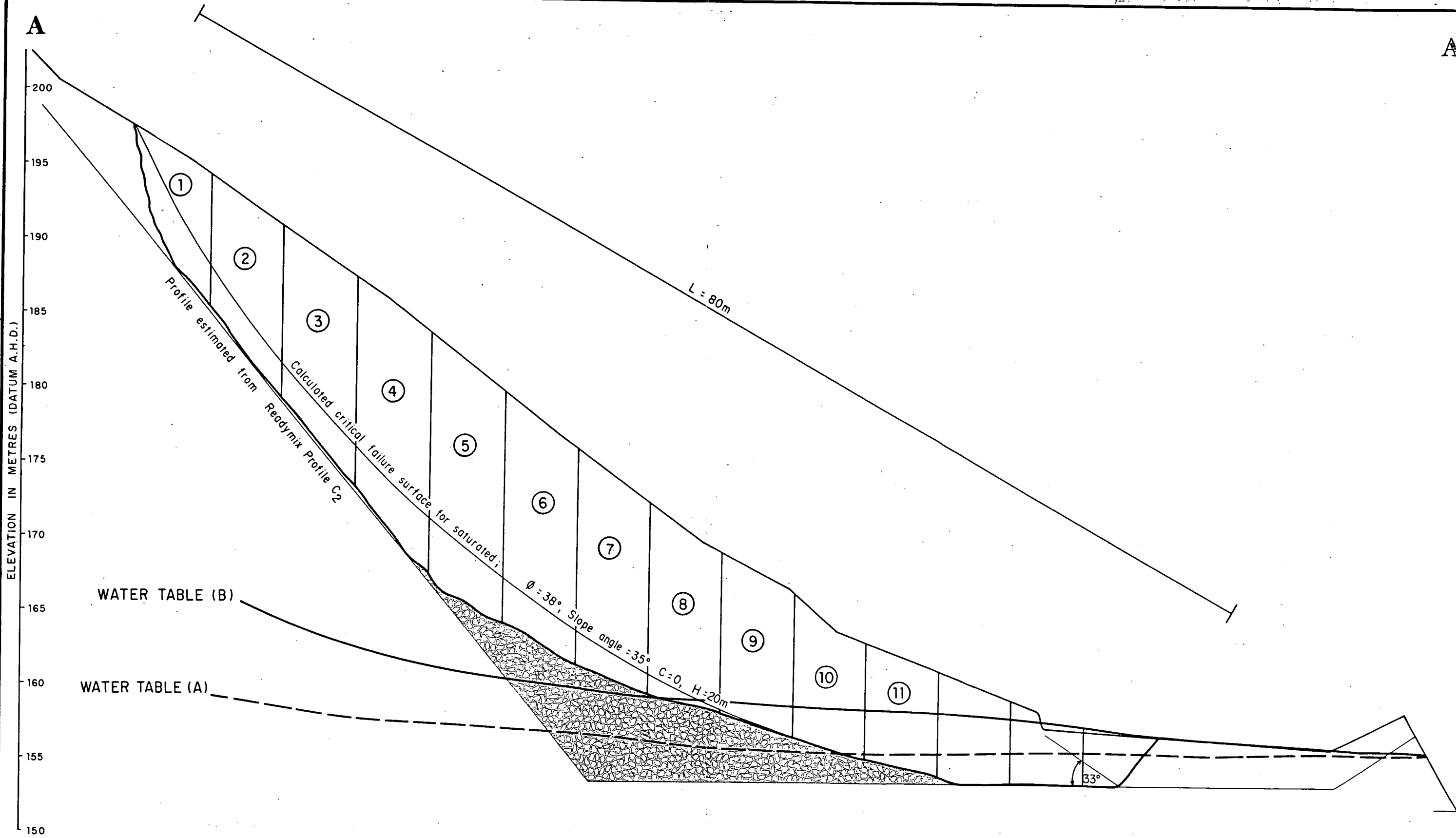
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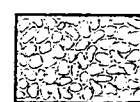
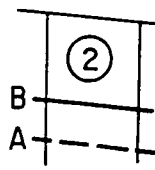
 DEPARTMENT OF MINES AND ENERGY SOUTH AUSTRALIA	COMPILED J. B.	18-11-85 C.D.O. DATE
	DRAWN M. B.	SCALE 1 : 1000
	DATE July '85	PLAN NUMBER
	CHECKED	85-377

RIVERVIEW QUARRY - REHABILITATED SLOPE STABILITY ANALYSIS

SEC. 5394 HD. YATALA

PROFILE 1



 Quarry waste not affected by failure surfaces
 Calculation slice with assumed water levels, parameters listed in Appendix 1

For locality, see Plan 85 - 376


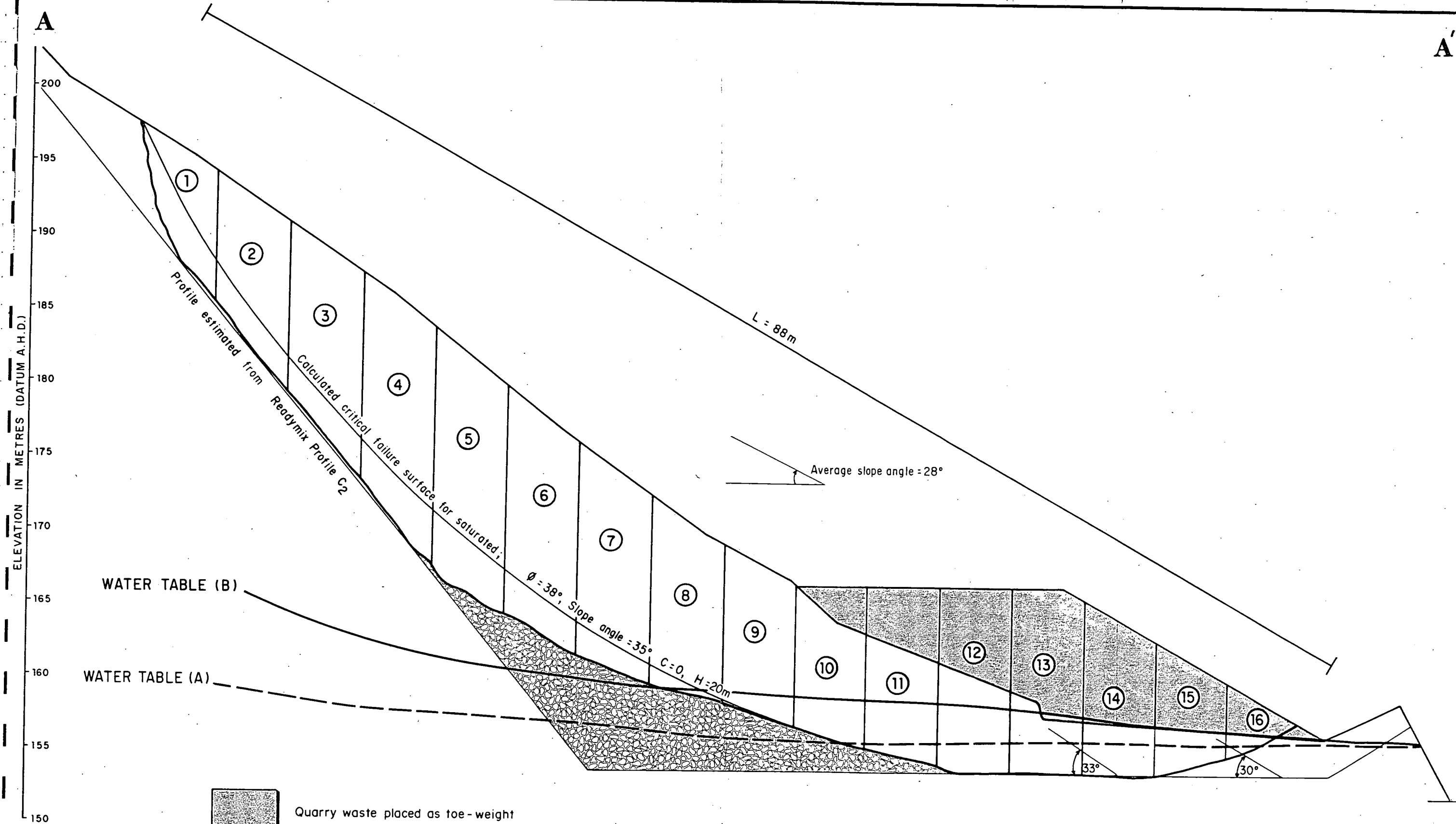
 DEPARTMENT OF MINES AND ENERGY SOUTH AUSTRALIA		COMPILED J. B.	<i>MR</i> 18.11.85 C D O DATE
RIVERVIEW QUARRY - REHABILITATED SLOPE STABILITY ANALYSIS		DRAWN M. B.	SCALE
SEC. 5394 HD. YATALA PROFILE 2		DATE July '85	PLAN NUMBER 85-378
		CHECKED	

Fig. 4

A

A'



Quarry waste placed as toe-weight


Quarry waste not affected by failure surfaces

Calculation slice with assumed water levels, parameters listed in Appendix 1

For locality, see Plan 85-376

Fig. 5

DEPARTMENT OF MINES AND ENERGY SOUTH AUSTRALIA		COMPILED J. B.	18-11-85 C D O DATE
RIVERVIEW QUARRY - REHABILITATED SLOPE STABILITY ANALYSIS		DRAWN M. B.	SCALE
SEC. 5394 HD. YATALA		DATE July '85	PLAN NUMBER
PROFILE 3		CHECKED	85-379

Surface boundaries : 1-7
 Subsurface boundaries : 8-12
 Assumed water table : 
 Soil types : MATERIAL 1-3

Note : Value of parameters shown on this plan show the general condition and may vary from values of calculations recorded in Appendix 2.

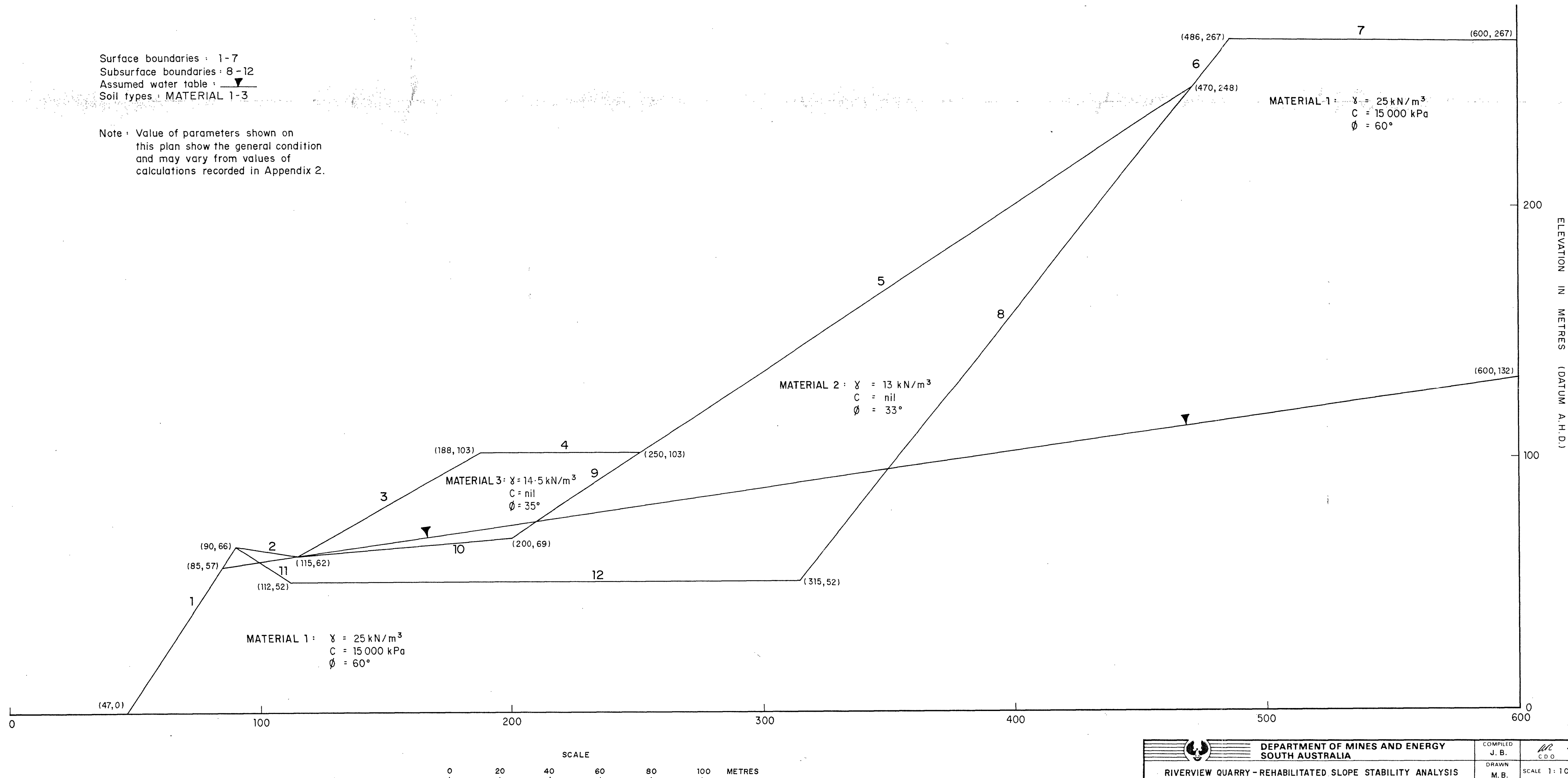



Fig. 6

 DEPARTMENT OF MINES AND ENERGY SOUTH AUSTRALIA	COMPILED J. B.	18-11-85 C.D.O. DATE
	DRAWN M.B.	SCALE 1:1000
	DATE July '85	PLAN NUMBER 85-380
	CHECKED	

RIVERVIEW QUARRY - REHABILITATED SLOPE STABILITY ANALYSIS
 SEC. 5394 HD. YATALA
 PROFILE 4 - 'STABL' ANALYSIS