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6-8 RUE CHASSELOUP-LAUBAT, 75737 PARIS CEDEX 15 - TEL.783 94.00 TELEX:270.844 F

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POINT COUNTING & BULK TESTING OF  
 200 LEVEL  
 OAKLEIGH CREEK MINE - TASMANIA

MICROFILMED

OPEN FILE

BY

M. J. LAWRENCE

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**B.R.G.M. AUSTRALIA**  
**GEOLOGICAL CONSULTANTS**

55 CLARENCE STREET SYDNEY N.S.W. 000  
 G.P.O. BOX 3314 SYDNEY 2001  
 TELEPHONE 29 5721  
 TELEX AA23047

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POINT COUNTING AND BULK TESTING OF 200 LEVELOAKLEIGH CREEK MINE - TASMANIAINTRODUCTION:

Point counting was performed on the 200 level at 1m intervals both north and south of the 1805 raise position. In addition to measuring the vein, the linear amount of wolframite and the width of the drive at the roof other measurements of drive width were taken 1.4m from the floor and at floor level. The average width of the drive at each sample location was important for the calculation of dilution and more particularly the volume of each bulk sample taken along the drives development so that proper weighting could take place to obtain an average value of ore grade for the level.

In the 1805 North 26 1m sample locations gave values out of 130 m. measured or 20% of the 1m intervals; whilst in the 1805 south programme no wolframite was recorded out of 60m. Hence overall the frequency of observed wolframite was 26/189 or 14%.

In addition, bulk samples were taken to compare overall the evaluation of the 200 Level average grade by point counting. Actual head grades are available for 12 samples and recalculated head grades are available for 3 more.

The influence of dilution was also examined and 3 samples were analysed to see if the expected percentage of vein quartz fragments was reached or was significantly lower, indicating dilution.

BULK TESTING 200 LEVEL:

A Report has already been submitted on the bulk testing of the 200L, but for completeness it is appended. The main results are shown below:

<u>1805 NORTH</u>	<u>DRIVE</u> <u>WIDTH (m)</u>	<u>VOL. REMOVED (m<sup>3</sup>)</u>	<u>% WO<sub>3</sub> GRADE (1.4 m STOPE)</u>		
Sample 3	1.91	65.1	0.184	)	)
5	2.02	53.0	0.188	)	)
6	2.66	83.3	0.446	)	)
7	2.70	80.2	0.337	)	)
8	2.53	66.3	0.361	)	0.342)
9	2.68	76.7	0.153	)	)
10	3.05	103.5	0.442	)	)
11	3.20	79.0	0.514	)	)
12	2.85	84.1	0.342	)	)
<u>1805 SOUTH</u>					) 0.297
Sample 1	2.07	168.5	0.247	)	)
2	1.81	60.9	0.070	)	0.185)
4	1.88	58.5	0.134	)	)

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In addition some information can be given for the relevant stope blocks:

200/102 (10m still to come)	0.37%
200/103 (all sampled)	0.32%
200/104 (17m not sampled)	0.20%
200/105 (only 11m sampled)	0.13%

Recent recalculated head grades (as opposed to actual head samples obtained above) are available for sample 13, 14, and 16.

Where possible, the grade of concentrate was used (else an average figure of 57%  $WO_3$  for concentrate obtained) and a 66% recovery was used. These results are tabulated below. If 80% was used grades decrease of about 18%.

<u>SAMPLE NO.</u>	<u>DRIVE WIDTH (m)</u>	<u>VOLUME (m<sup>3</sup>)</u>	<u>TONNES MILLED</u>	<u>KG. CONC. RECOVERED</u>	<u>GRADE (KG/T)</u>	<u>CONC. GDE % <math>WO_3</math></u>	<u>RECALC. HEAD GD</u> 1.4m - % $WO_3$
Sample 13	3.22	106.6	213.2	562	2.64	61.1	.562
Sample 14	3.33	100.9	262.4	605	2.31	56.5	.470
Sample 15	N/A	N/A	N/A	N/A	N/A	N/A	N/A
Sample 16	3.02	86.1	205.8	436	2.12	*	.395

∅ Assumes 66% recovery. If actual recovery is higher this grade is less.

Usually low grade is associated with low recovery and vice versa.

\* Where concentrate grade unknown 57% has been assumed.

These figures show a slight improvement in grade northwards.

#### DILUTION STUDY

Three samples (240/101, 240/104 and 280/102) were checked for amount quartz present in them. The idea was to see if dilution was a noticeable factor as would be indicated by a lower than expected "% qtz" measurement. This was not found to be the case. In all cases there was slightly more quartz found than expected by a ratio calculation of the average vein width for the stope block to the average stope width.

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TABLE 1  
AVERAGE VEIN AND STOPE WIDTHS - OAKLEIGH CREEK

<u>STOPE</u>	<u>CO ORDS</u>	<u>STOPE WIDTH (m)</u>	<u>VEIN WIDTH (cm)</u>
240/101	(1935-1907)	1.71	38.14 $\pm$ 3.90
240/102	(1905-1856)	1.61	42.73 $\pm$ 5.41
240/103	(1854-1804)	1.48	45.00 $\pm$ 8.18
240/104	(1802-1757)	1.27	37.97 $\pm$ 6.07
240/105	(1755-1708)	1.39	33.42 $\pm$ 5.84
280/102	(1904-1858)	1.43	51.43 $\pm$ 6.66
280/103	(1856-1808)	1.53	53.53 $\pm$ 6.12
280/104	(1806-1756)	1.60	41.25 $\pm$ 9.86

Note measurements taken at position of raises were not taken into account in determining average vein widths. The average stope widths are those as at 4/11/1980 and may not be those in effect at time ore broken which was treated in early November.

TABLE 2

<u>STOPE</u>	<u>EXPECTED QUARTZ RATIO</u>	<u>FOUND QUARTZ RATIO</u>
240/101	.3814/1.71 = 22.3%	24.4%
240/104	.3797/1.27 = 29.9%	30.9%
280/102	.5143/1.43 = 36.0%	38.6%

Taking tons broken and measured stope widths as at 4/11/1980 a weighted average for the 240 and 280 levels was obtained. It was 1.51m compared with original feasibility estimate of 1.2m and current planned width of 1.4m. The stope width on 240 level averaged 1.49m and 1.55m on the 280 level.

It is interesting to compare the average vein widths for the three levels as it was originally thought that the vein would widen at depth (since drill hole data suggested this). Values calculated for the levels show the opposite to be the case.

280 level	48.7 cm vein width
240 level	39.4 cm vein width
200 level	35.9 cm vein width.

POINT COUNTING RESULTS (see Appendix II)

The Average Grade (Point counting) for the whole 129 m of drive on 200 level North of cross-cut (actually rise 1805N) was calculated, weighting being by width of vein. This value was then recalculated back to a stoping width of 1.4m. For its entire length, the average width of vein was 37.3 cm. Hence the average grade for point counting of the vein was 2.14%  $WO_3$  or 0.57%  $WO_3$  (1.4m STOPE) and 0.25%  $WO_3$  (3.21m DRIVE). The actual grade from plant testing, samples to 90m mark from 200 level was 0.342%  $WO_3$  (1.4m STOPE) or 68% of bulk sampling results.

(Samples 3, 5, 6, 7, 8, 9, 10, 11, 12). If only point counting values for equivalent samples plant tested were taken (i.e. to sample 90 North) point counting grade was 2.26%  $WO_3$  for vein, 0.60%  $WO_3$  (1.4 m STOPE), and 0.26%  $WO_3$  (3.21m DRIVE) showing that bulk sampling obtained 176% of bulk sampling results or grades 57% of that predicted.

In the South, point counting failed to record any grade (0%  $WO_3$ ) for the 60m measured. However, bulk testing (samples 1,2,4) yielded 0.185%  $WO_3$  (1.4m STOPE).

Over the whole 190m long section of 200m measured the point counting suggested a grade of 1.67%  $WO_3$  (VEIN), 0.43%  $WO_3$  (1.4m STOPE) and 0.18%  $WO_3$  (3.29 m DRIVE). For comparison the grade obtained by bulk sampling was 0.297%  $WO_3$  (1.4 m STOPE) i.e. point counting obtained 145% of bulk sampling values or grades 69% of that predicted.

It is instructive to note that to 15.12.80, the metallurgist calculated 40060 kg saleable concentrates (70%  $WO_3$  usually) produced from a wet treated tonnage of 13681 tonnes i.e. a recovered grade of 0.293%  $WO_3$  for the mine.

In the 1805 North programme the following data were measured:

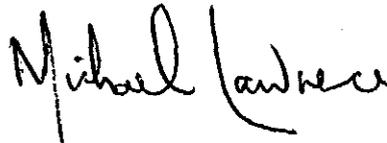
Mean Drive Width	3.21 m (141 measurements)
Mean Vein Width	37.3 cm (139 measurements)
Point Counting Grade	2.14% $WO_3$ for Vein
Point Counting Grade	0.57% $WO_3$ for 1.4m Stope (whole 130 m)
* Point Counting Grade	0.60% $WO_3$ (initial 90 m)
* Bulk Testing Grade	0.342% $WO_3$ for 1.4m stope (initial 90 m)

In the 1805 South programme the following data were measured:

- Mean Drive Width 3.51 m (51 measurements)
- Mean Vein Width 31.4 cm (46 measurements)
- \* Point Counting Grade 0%  $W_3$  for vein
- Point Counting Grade 0%  $W_3$  for 1.4 m Stope (whole 60 m)
- \* Bulk Testing Grade 0.185%  $W_3$  for 1.4 m Stope (whole 60 m)

Total 200L programme (190 m)

- Mean Drive Width 3.29 m (192 measurements)
- Mean Vein Width 35.9 cm (185 measurements)
- Point Counting Grade 1.67%  $W_3$  for Vein
- \* Point Counting Grade 0.43%  $W_3$  for 1.4 m STOPE (whole 190m)
- \* Bulk Testing Grade 0.297%  $W_3$  for 1.4 m STOPE (whole 190 m)

  
M. J. LAWRENCE.

1805 NORTH					
SAMPLE NO.		WIDTH WOLFRAMITE	WIDTH QUARTZ	WIDTH VEIN	% W <sub>3</sub>
1		25	305	330	13.92
2				365	
3				350	
4				330	
5		5	340	345	2.94
6	Average	9	381	390	4.61
7	Vein Width			420	
8	34.9 cm			300	
9				340	
10				350	
11				360	
12				330	
12 + 710 mm				335	
13				335	
14				340	
15	Av.			350	
16	Vein Width			345	
17	34.4 cm			340	
18		45	315	360	21.36
19		7	363	370	3.81
20				325	
20 + 890 mm				340	
21		15	325	340	8.52
22				390	
23	Av. Vein			385	
24	Width			340	
25	37.5 cm			370	
26				380	
27		50	320	370	22.76
28		28	362	390	13.27
29				380	
30		20	380	400	9.57
31				400	
31 + 120				400	

1805 NORTH					
SAMPLE NO.		<u>WIDTH WOLFRAMITE</u>	<u>WIDTH QUARTZ</u>	<u>WIDTH VEIN</u>	<u>% WO<sub>3</sub></u>
32				390	
33				390	
34				390	
35	Ave.	95	325	420	33.77
36	Vein Width			380	
37				410	
38	39.9 cm			390	
39				410	
40				390	
41				410	
41 + 210				410	
42				415	
43				400	
44				440	
45				390	
46	Av. Vein	11	359	370	5.88
47	Width			410	
48	39.4			380	
49				430	
50				390	
50 + 890				300	
51				300	
52				370	
53	Average			400	
54	Vein			380	
55	Width			350	
56				370	
57				360	
58				355	
59				345	
59 + 900				310	

1805 NORTH SAMPLE NO.	WIDTH WOLFRAMITE	WIDTH QUARTZ	WIDTH VEIN	% WO <sub>3</sub>
60			310	
61			375	
62			345	
63 Average			340	
64 Width			345	
65 Vein			330	
66 34.8 cm			335	
67			340	
68			355	
69			360	
70			370	
71			370	
71 + 420			370	
72			355	
73			400	
74 Average			400	
75 Width	12	408	420	5.66
76 Vein			405	
77 39.10 cm	10	415	425	4.70
78	12	388	400	5.93
79	34	321	355	17.08
80			375	
81			355	
82 Average			400	
83 Width			345	
84 Vein			325	
85 37.3 cm	10	405	415	4.81
86	37	308	345	18.80
87			350	
88			420	
89	10	380	390	5.10
90			385	

<u>1805 NORTH</u>					
<u>SAMPLE NO.</u>		<u>WIDTH WOLFRAMITE</u>	<u>WIDTH QUARTZ</u>	<u>WIDTH VEIN</u>	<u>% WO<sub>3</sub></u>
91				420	
92				375	
93				390	
94	Average			430	
95	Vein	15	415	430	6.84
96	Width			450	
97	42.2 cm			375	
98				395	
99		20	520	540	7.24
100				420	
101				420	
102				405	
103		20	400	420	9.15
104	Average	5	405	410	2.48
105	Vein	70	340	410	21.62
106	Width	60	365	425	23.58
107	36.9 cm			390	
108				385	
109				375	
110				380	
111				230	
111 + 200				230	
112				250	
113	Average			325	
114	Vein	5	440	445	2.30
115	Width			430	
116	40.9 cm			425	
117		10	325	335	5.90
118				360	
119				375	
119 + 800				380	

<u>1805 NORTH</u> <u>SAMPLE NO.</u>	<u>WIDTH WOLFRAMITE</u>	<u>WIDTH QUARTZ</u>	<u>WIDTH VEIN</u>	<u>% WO<sub>3</sub></u>
120			380	
121			420	
122			340	
123 Average			385	
124 Vein			420	
125 Width			390	
126 40.9 cm			350	
127			355	
128			370	
129			350	
129 + 350			350	

<u>1805 SOUTH</u> <u>SAMPLE NO.</u>	<u>WIDTH WOLFRAMITE</u>	<u>WIDTH QUARTZ</u>	<u>WIDTH VEIN</u>	<u>% WO<sub>3</sub></u>
0			340	
1			300	
2			285	
3 Av. Vein			305	
4 Width			320	
5 31.8 cm			300	
6			380	
7			330	
8 - 23			Unmeasured	
24			310	
25			305	
26			300	
27			315	
28			340	
29			350	
30			340	
31			340	
32			275	
33			330	
34			365	
35			350	
36			310	
37			250	
38			220	
39			365	
40			335	
40 + 600			340	

<u>1805 SOUTH</u> <u>SAMPLE NO.</u>	<u>WIDTH WOLFRAMITE</u>	<u>WIDTH QUARTZ</u>	<u>WIDTH VEIN</u>	<u>% WO<sub>3</sub></u>
41			340	
42			360	
43 Av. Vein			340	
44 Width			395	
45 34.7 cm			360	
46			310	
47			-	
48			-	
49			-	
50			330	
50 + 600			350	
51			360	
52			345	
53 Average			290	
54 Vein			220	
55 Width			210	
56 28.1 cm			165	
57			280	
58			265	
59			285	
60			300	
60 + 600			300	

200 LEVEL DEVELOPMENT

## — Interim Report and Revision of Earlier Work —

Interim reporting of results from the 200 Level development drive has been updated to include sample lots 7 to 12, all of which continue on the north heading. Unfortunately, a revision of dilution factors obtained for sample lots 1 to 6 has been necessary. Originally the drive width had been measured to the nearest 10cm; in order to increase the accuracy these widths have been remeasured to the nearest 1cm and the dilution factors recalculated. Sample lots 1 to 6 were taken on a 1.8m drive which was subsequently stripped to 2.6m to provide easier access. Correction factors for the widths of sample lots 1 to 6 were calculated from values obtained for samples 7 to 12. A summary of this is shown in the appendix.

Further to this the analysis of sample No.6, previously reported, has since been proven incorrect and has been amended. Improper grinding of this sample had resulted in coarse material being presented to XRF and hence the low  $WO_3$  values obtained.

TONNES : 366  
 DRIVE LENGTH : 24.6m  
 MEAN WIDTH : 2.07m  
 VOL. REMOVED : 168.5m<sup>3</sup>

Belt Sample : 340kg ( 1x 34cm section from belt every 20 min. )  
 Dry Weight : 325kg  
 % H<sub>2</sub>O : 4.4%

### Sample Preparation

The entire sample was crushed to <9.0mm and riffled to 17.0kg. This was then pulverised to <1.0mm and 3x 200g samples were riffled from this for final analysis. Each 200g sample was ground in a ring mill to ensure complete homogeneity before splitting for final analysis. Duplicate analyses for WO<sub>3</sub> were performed by A.L.S. using methods XRF-2 and 104 ( A.A.S. ). i.e. results were average of two samples for each method.

### Results

( duplicate WO <sub>3</sub> )	<u>WO<sub>3</sub></u> (XRF2)	<u>WO<sub>3</sub></u> (104)	<u>Sn</u> (XRF2)	<u>As</u> (XRF2)	<u>S</u> (155)
A	.16	.18	135ppm	220ppm	.33
B	.15	.16	140 "	250 "	.33
C	.17	.18	140 "	250 "	.33

Mean WO<sub>3</sub> = 0.167 %

### Statistical Accuracy

Based on a top size of 40mm and a liberation size of 1.0mm, Gy's basic sampling equation states that error committed in sampling the millfeed was 0.085. In other words the relative error in sampling is ±8.5 % of the sample assay. Further to this, another error is committed in reducing the sample size after crushing. Thus, for a 17.0kg sample with a top size of 9.0mm, the committed error is 0.060 - or ± 6.0 % relative.

The sum sampling error committed is :

$$s = \sqrt{(0.085)^2 + (0.060)^2}$$

$$= \underline{0.104} \quad \text{- or } \underline{10.4 \% \text{ relative.}}$$

### Ore Dilution

With a planned stope width of 1.4m, the mean width of this section ( 2.07m ) represents a dilution factor of 47.9%. Thus, material

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recovered from this section by normal stoping methods could be expected to run at  $0.247 \pm 0.026$  %  $WO_3$  .

200 DRIVE SAMPLE 2

TONNES: 137.6  
 DRIVE LENGTH : 10.1m  
 MEAN WIDTH : 1.81m  
 VOL. REMOVED : 60.9m<sup>3</sup>  
 Belt Sample : 350kg (2x 700mm cuts from belt every 20 min. )  
 No. Taken : 56  
 Dry Weight : 336kg  
 % H<sub>2</sub>O : 4.0 %

Size Analysis

+30mm : 63.5kg = 18.9 %  
 -30/+20mm : 76.0kg = 22.6 %  
 -20mm : 196.5kg = 58.5 %

Sample Preparation

All samples were crushed to <9.0mm and reduced to the following weights :

+30mm 15.5kg  
 -30/+20mm 16.0kg  
 -20mm 18.5kg

These were then pulverised to <1.0mm and riffled to duplicate samples of 200g each. Analyses were performed by A.L.S. using method XRF-2 for WO<sub>3</sub>, Sn & As and method 155 for S.

Results

		<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
+30mm	A	.01	.01	45ppm	<.10	.15
	B	.01	.01	40 "	<.10	.19
-30/+20mm	A	.03	.03	50 "	<.10	.19
	B	.03	.03	55 "	<.10	.18
-20mm	A	.07	.08	55 "	<.10	.20
	B	.08	.08	70 "	<.10	.20

Weighted Mean WO<sub>3</sub> = 0.054 %

Statistical Accuracy

According to Gy's basic sampling equation, the statistical error committed in sampling the mill feed was 0.140. In reducing the sample size after crushing, committed error was 0.095. The sum sampling error was :

$$\begin{aligned}
 s &= \sqrt{(0.14)^2 + (0.095)^2} \\
 &= \underline{0.169} \quad - \text{ or } \underline{16.9 \% \text{ relative}}
 \end{aligned}$$

Ore Dilution

The mean width of this section ( 1.81m ) represents a dilution factor of 29.3 % when compared to the planned stope width of 1.4m . Thus, by normal stoping methods this section could be expected to yield  $0.070 \pm 0.012$  %  $WO_3$  .

200 DRIVE SAMPLE 3

TONNES : 236.1  
 DRIVE LENGTH : 11.6m  
 MEAN WIDTH : 1.91m  
 VOL.REMOVED : 65.1m<sup>3</sup>

Belt Sample : N/R ( 2x 700mm cuts from belt every 30 min. )  
 No. Taken : 69  
 Dry Weight : 447kg  
 % H<sub>2</sub>O : N/R

Sample Preparation

Sample crushed to <10mm and reduced to 14.0kg. This was then pulverised to <1.0mm and duplicate samples of 200g riffled out for final analysis.

Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.15	.15	120ppm	<.10	.27
B	.12	.12	120 "	<.10	.23

Mean WO<sub>3</sub> = 0.135 %

Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.075 and in reducing the crushed sample, 0.035 .

The sum sampling error was :

$$s = \sqrt{(0.075)^2 + (0.035)^2}$$

$$= \underline{0.0828} \quad \text{- or } \underline{8.28 \% \text{ relative}}$$

Ore Dilution

The mean width of this section ( 1.91m ) represents a dilution factor of 36.4 % when compared to the planned stope width of 1.4m. By normal stoping methods this section should produce 0.184 ± 0.015 % WO<sub>3</sub> .

TONNES : 157.5  
 DRIVE LENGTH : 10.0m  
 MEAN WIDTH : 1.88m  
 VOL. REMOVED : 58.5m<sup>3</sup>

Belt Sample : 410kg ( 2x 700mm cuts from belt every 30 min. )  
 No. Taken : 75  
 Dry Weight : 391kg  
 % H<sub>2</sub>O : 4.6 %

#### Sample Preparation

Sample was crushed to <10mm and reduced to 23kg .  
 This was then pulverised to <1.0mm and duplicate samples of 200g  
 riffled out for final analysis .

#### Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.11	.10	115ppm	<.10	.31
B	.10	.09	115 "	<.10	.29

$$\underline{\text{Mean WO}_3 = 0.10 \%}$$

#### Statistical Accuracy

Statistical error committed in sampling the mill  
 feed was 0.10 and in reducing the crushed sample, 0.060 .

The sum sampling error was :

$$s = \sqrt{(0.10)^2 + (0.060)^2}$$

$$= \underline{0.117} \text{ - or } \underline{11.7 \% \text{ relative}}$$

#### Ore Dilution

The mean width of this section (1.88m) represents a  
 dilution factor of 34.3 % when compared to the planned stope width of  
 1.4m . By normal stoping methods the expected grade from this section  
 would be 0.134 ± 0.016 % WO<sub>3</sub> .

TONNES : 122.3  
 DRIVE LENGTH : 8.2m  
 MEAN WIDTH : 2.02m  
 VOL REMOVED : 53.0m<sup>3</sup>

Belt Sample : N/R ( 2x 700mm cuts from belt every 30 min. )  
 No. taken : 26  
 Dry Weight : 116kg  
 % H<sub>2</sub>O : N/R

### Sample Preparation

Sample was crushed to <10mm and reduced to 16.5kg. After pulverising to <1.0mm duplicate samples of 200g were riffled out for final analysis.

### Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.14	.14	185ppm	<.10	.30
B	.12	.12	210 "	<.10	.33

$$\underline{\text{Mean WO}_3 = 0.130 \%}$$

### Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.175 and in reducing the crushed sample, 0.070 .

The sum sampling error was :

$$s = \sqrt{(0.175)^2 + (0.070)^2}$$

$$= \underline{0.188} \text{ - or } \underline{18.8 \% \text{ relative}}$$

### Ore Dilution

The mean width of this section (2.02m) represents a dilution factor of 44.3 % when compared to the planned stope width of 1.4m . By normal stoping methods this section should produce 0.188 ± 0.035 % WO<sub>3</sub> .

021

200 DRIVE SAMPLE 6 - AMMENDED

TONNES : 256.9  
 DRIVE LENGTH : 10.2m  
 MEAN WIDTH : 2.66m  
 VOL. REMOVED : 83.3m<sup>3</sup>  
 Belt Sample : 569.5kg ( 2x 700mm cuts from belt every 30 min. )  
 No. Taken : 84  
 Dry Weight : 551.25kg  
 % H<sub>2</sub>O : 3.2 %

Sample Preparation

The mill feed sample was crushed to <10mm and reduced to 25.5kg . This was then pulverised to <1.0mm and reduced to two 150g samples for final analysis.

Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.25	.25	60ppm	<.10	.13
B	.22	.22	55 "	<.10	.14

$$\underline{\text{Mean WO}_3 = 0.235 \%}$$

Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.055 and in reducing the crushed sample, 0.045 .

The sum sampling error was :

$$\begin{aligned}
 s &= \sqrt{(0.055)^2 + (0.045)^2} \\
 &= \underline{0.071} \text{ - or } \underline{7.1 \% \text{ relative}} .
 \end{aligned}$$

Ore Dilution

The mean width of this section (2.66m ) represents a dilution factor of 90.0 % when compared to the planned stope width of 1.4m . By normal stoping methods this section should produce 0.446 ± 0.032 % WO<sub>3</sub> .

022

200 DRIVE SAMPLE 7

TONNES : 154  
 DRIVE LENGTH : 10.1m  
 MEAN WIDTH : 2.70m ;  
 VOL. REMOVED : 80.2m<sup>3</sup>  
  
 Belt Sample : 365.25kg ( 2x 700mm cuts from belt every 30 min. )  
 No. Taken : 36  
 Dry Weight : 353.5kg  
 % H<sub>2</sub>O : 3.2 %

Sample Preparation

Sample was crushed to <10mm and reduced to 23kg .  
 After pulverising to <1.0mm, duplicate samples of approx. 150g were riffled out for final analysis .

Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.18	.18	85ppm	<.10	.17
B	.17	.17	85 "	<.10	.21

$$\underline{\text{Mean WO}_3 = 0.175 \%}$$

Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.080 and in reducing the crushed sample, 0.055 .

The sum sampling error was :

$$\begin{aligned}
 s &= \sqrt{(0.080)^2 + (0.055)^2} \\
 &= \underline{0.097} \text{ - or } \underline{9.7 \% \text{ relative}} .
 \end{aligned}$$

Ore Dilution

The mean width of this section (2.70m) represents a dilution factor of 92.9% when compared to the planned stope width of 1.4m. By normal stoping methods this section should produce

0.337 ± 0.033 % WO<sub>3</sub> .

023

200 DRIVE SAMPLE 8

TONNES : 254.6  
 DRIVE LENGTH : 8.8m  
 MEAN WIDTH : 2.53m  
 VOL. REMOVED : 66.3m<sup>3</sup>

Belt Sample : N/A ( 2x 700mm cuts from belt every 30 min. )  
 No. Taken : 76  
 Dry Weight : 567kg  
 % H<sub>2</sub>O : N/A

Sample Preparation

The mill feed sample was crushed to <10mm and reduced to 23kg. This was then pulverised to <1.0mm and reduced to two 150g samples for final analysis.

Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.21	.21	165ppm	<.10	.18
B	.19	.19	155 "	<.10	.16

Mean WO<sub>3</sub> = 0.20 %

Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.060 and in reducing the crushed sample, 0.050 .

The sum sampling error was :

$$s = \sqrt{(0.060)^2 + (0.050)^2}$$

$$= 0.078 \text{ - or } 7.8 \% \text{ relative .}$$

Ore Dilution

The mean width of this section (2.53m) represents a dilution factor of 80.7 % when compared to the planned stope width of 1.4m . By normal stoping methods this section should produce 0.361 ± 0.028 % WO<sub>3</sub> .

024

200 DRIVE SAMPLE 9

TONNES : 301.7  
 DRIVE LENGTH : 9.9m  
 MEAN WIDTH : 2.68m  
 VOL. REMOVED : 76.7m<sup>3</sup>  
 Belt Sample : N/A ( 2x 700mm cuts from belt every 30 min. )  
 No. Taken : 72  
 Dry weight : 556  
 % H<sub>2</sub>O : N/A

Sample Preparation

The mill feed sample was crushed to <10mm and reduced to 21kg . This was then pulverised to <1.0mm and reduced to two 150g samples for final analysis .

Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.08	.08	175ppm	<.10	.17
B	.08	.08	130 "	<.10	.15

$$\underline{\text{Mean WO}_3 = 0.080 \%}$$

Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.090 and in reducing the crushed sample 0.080 .

The sum sampling error was :

$$\begin{aligned}
 s &= \sqrt{(0.090)^2 + (0.080)^2} \\
 &= \underline{0.120} \text{ - or } \underline{12.0 \% \text{ relative}} .
 \end{aligned}$$

Ore Dilution

The mean width of this section (2.68m) represents a dilution factor of 91.4 % when compared to the planned stope width of 1.4m . By normal stoping methods this section should produce 0.153 ± 0.018 % WO<sub>3</sub> .

200 DRIVE SAMPLE 10

TONNES : 165.8  
 DRIVE LENGTH : 11.5m  
 MEAN WIDTH : 3.05m  
 VOL. REMOVED : 103.5m<sup>3</sup>

Belt Sample : N/A ( 2x 700mm cuts from belt every 30 min. )  
 No. Taken : 48  
 Dry Weight : 321kg  
 % H<sub>2</sub>O : N/A

Sample Preparation

The mill feed was crushed to <10mm and reduced to 26kg . This was then pulverised to <1.0mm and reduced to two 150g samples for final analysis .

Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.21	.20	105ppm	<.10	.19
B	.20	.20	130 "	<.10	.16

$$\underline{\text{Mean WO}_3 = 0.203 \%}$$

Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.075 and in reducing the crushed sample, 0.050 .

The sum sampling error was :

$$s = \sqrt{(0.075)^2 + (0.050)^2}$$

$$= \underline{0.090} \text{ - or } \underline{9.0 \% \text{ relative} .}$$

Ore Dilution

The mean width of this section (3.05m) represents a dilution factor of 117.9% when compared to the planned stope width of 1.4m . By normal stoping methods this section should produce 0.442 ± 0.040 % WO<sub>3</sub> .

026

200 DRIVE SAMPLE 11

TONNES : 198.2  
 DRIVE LENGTH : 8.6m  
 MEAN WIDTH : 3.20m  
 VOL. REMOVED : 79.0m<sup>3</sup>

Belt Sample : 267kg ( 2x 700mm cuts from belt every 30 min )  
 No. Taken : 50  
 Dry Weight : 260kg  
 % H<sub>2</sub>O : 2.7 %

Sample Preparation

The mill feed sample was crushed to <10mm and reduced to 25kg . This was then pulverised to <1.0mm and reduced to two 150g samples .

Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.24	.23	85ppm	<.10	.17
B	.22	.21	85 "	<.10	.18

$$\underline{\text{Mean } \text{WO}_3 = 0.225 \%}$$

Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.080 and in reducing the crushed sample, 0.045 .

The sum sampling error was :

$$s = \sqrt{(0.080)^2 + (0.045)^2}$$

$$= \underline{0.092} \text{ - or } \underline{9.2 \% \text{ relative}} .$$

Ore Dilution

The mean width of this section ( 3.20m) represents a dilution factor of 128.6 % when compared to the planned stope width of 1.4m . By normal stoping methods this section should produce 0.514 ± 0.047 % WO<sub>3</sub> .

200 DRIVE SAMPLE 12

TONNES : 215.8  
 DRIVE LENGTH : 10.0m  
 MEAN WIDTH : 2.85m  
 VOL. REMOVED : 84.1m<sup>3</sup>

Belt Sample : N/A ( 2x 700mm cuts from belt every 30 min. )  
 No. Taken : 78  
 Dry Weight : 556kg  
 % H<sub>2</sub>O : N/A

Sample Preparation

The mill feed sample was crushed to <10mm and reduced to 28kg . This was then pulverised to <1.0mm and reduced to two 150g samples for final analysis .

Results

	<u>WO<sub>3</sub></u>	<u>WO<sub>3</sub></u>	<u>Sn</u>	<u>As</u>	<u>S</u>
A	.17	.17	120ppm	<.10	.20
B	.17	.16	120 "	<.10	.18

$$\underline{\text{Mean WO}_3 = 0.168 \%}$$

Statistical Accuracy

Statistical error committed in sampling the mill feed was 0.065 and in reducing the crushed sample, 0.055 .

The, sum sampling error was :

$$s = \sqrt{(0.065)^2 + (0.055)^2}$$

$$= \underline{0.085} \text{ - or } \underline{8.5 \% \text{ relative}}$$

Ore Dilution

The mean width of this section (2.85m) represents a dilution factor of 103.6% when compared to the planned stope width of 1.4m . By normal stoping methods this section should produce 0.342 ± 0.029 % WO<sub>3</sub> .

SUMMARY

Results for the various sample lots have been summarised and related to the proposed development blocks for the 200 Level. Calculations for this are shown in the appendix, together with diagrams of the sample lots / block positions. (1.4m mining width)

200 / 102	weighted mean $WO_3$ = 0.369 % ( 10m of this section yet to be sampled )
200 / 103	weighted mean $WO_3$ = 0.315 %
200 / 104	weighted mean $WO_3$ = 0.201 % ( 17m access not sampled )
200 / 105	weighted mean $WO_3$ = 0.130 % ( only 10.6m developed due to poor grade )

A.W.Nicholls



Plant Metallurgist

c.c. G.E.  
M.J.L.  
T.A.M.  
J.Mc.  
J.W.B.

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APPENDIX

CALCULATION - WIDTH CORRECTION FOR SAMPLE LOTS 1 to 6

030

<u>LOT No.</u>	<u>200 LEVEL ORIGINAL</u>				<u>200 LEVEL REVISED</u>		<u>WIDTH VARIATION</u>		
	<u>H(Av.)</u>	<u>W(Av.)</u>	<u>L</u>	<u>VOL.</u>	<u>W(Av.)</u>	<u>VOL.</u>	<u>Δ W</u>	<u>% VAR.</u>	<u>UNIT VAR.</u>
12	2.95m	3.10m	10.0m	91.4m <sup>3</sup>	2.85m	84.1m <sup>3</sup>	-0.25m	-8.0645	-1.6492
11	2.87	2.66	8.6	65.6	3.20	79.0	+0.54	+20.3007	+3.5703
10	2.95	2.63	11.5	89.2	3.05	103.5	+0.42	+15.9696	+3.7556
9	2.89	2.57	9.9	73.5	2.68	76.7	+0.11	+4.2802	+0.8665
8	2.98	2.64	8.8	69.2	2.52	66.1	-0.12	-4.5454	-0.8180
7	2.94	2.92	10.1	86.7	2.70	80.2	-0.22	-7.5342	-1.5561
6	3.07	2.56	10.2	80.2	2.66*	83.3		TOTAL	<u>+4.1691</u>
5	3.20	1.94	8.2	50.9	2.02*	53.0			
4	3.11	1.81	10.0	56.3	1.88*	58.5			
3	2.94	1.83	11.6	62.4	1.91*	65.1			
2	3.33	1.74	10.1	58.5	1.81*	60.9			
1	3.31	1.99	24.6	162.0	2.07*	168.5			

- OR A POSITIVE VARIATION  
OF 4.169 %

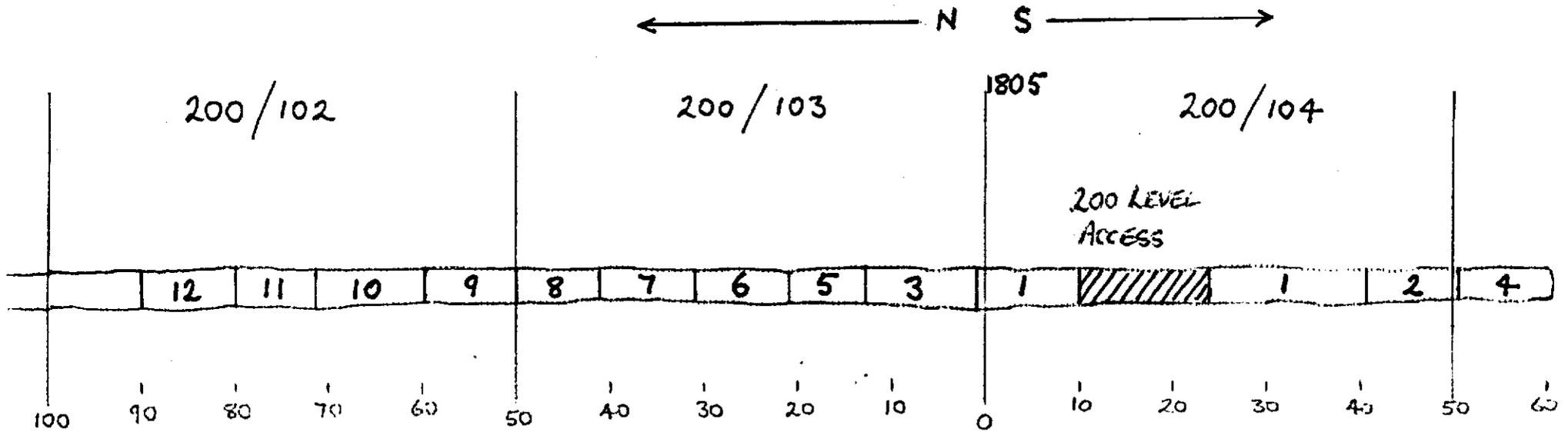
\* Calculated from mean width variation  
in samples 7 to 12 .

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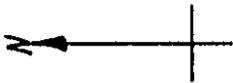
	LOT No.	%WO <sub>3</sub>	W(Av.)	H(Av.)	L(In Block)	V(In Block)	VxWO <sub>3</sub>		
200/102	12	.342	2.85	2.95	10.0	84.1	28.762	$\text{MEAN } \text{WO}_3 = \frac{\text{VxWO}_3 \text{TOT}}{\text{VTOT}}$ $= \frac{126.850}{343.3}$ $= 0.369 \% \text{WO}_3$	
	11	.514	3.20	2.87	8.6	79.0	40.606		
	10	.442	3.05	2.95	11.5	103.5	45.747		
	9	.153	2.68	2.89	9.9	76.7	11.735		
					TOTAL	343.3	TOTAL	126.850	
200/103	8	.361	2.53	2.98	8.8	66.3	23.934	$\text{MEAN } \text{WO}_3 = \frac{111.907}{355.4}$ $= 0.315 \% \text{WO}_3$	
	7	.337	2.70	2.94	10.1	80.2	27.027		
	6	.446	2.66	3.07	10.2	83.3	37.152		
	5	.188	2.02	3.20	8.2	53.0	9.964		
	3	.184	1.91	2.94	11.6	65.1	11.978		
	1	.247	2.07	3.31	1.1	7.5	1.852		
					TOTAL	355.4	TOTAL	111.907	
200/104	1	.247	2.07	3.31	23.5	161.0	39.767	$\text{MEAN } \text{WO}_3 = \frac{43.778}{218.3}$ $= 0.201 \% \text{WO}_3$	
	2	.070	1.81	3.33	9.5	57.3	4.011		
					TOTAL	218.3	TOTAL	43.778	
200/105	2	.070	1.81	3.33	0.6	3.6	0.252	$\text{MEAN } \text{WO}_3 = \frac{8.091}{62.1}$ $= 0.130 \% \text{WO}_3$	
	4	.134	1.88	3.11	10.0	58.5	7.839		
					TOTAL	62.1	TOTAL	8.091	

200 LEVEL SAMPLE / BLOCK LOCATIONS

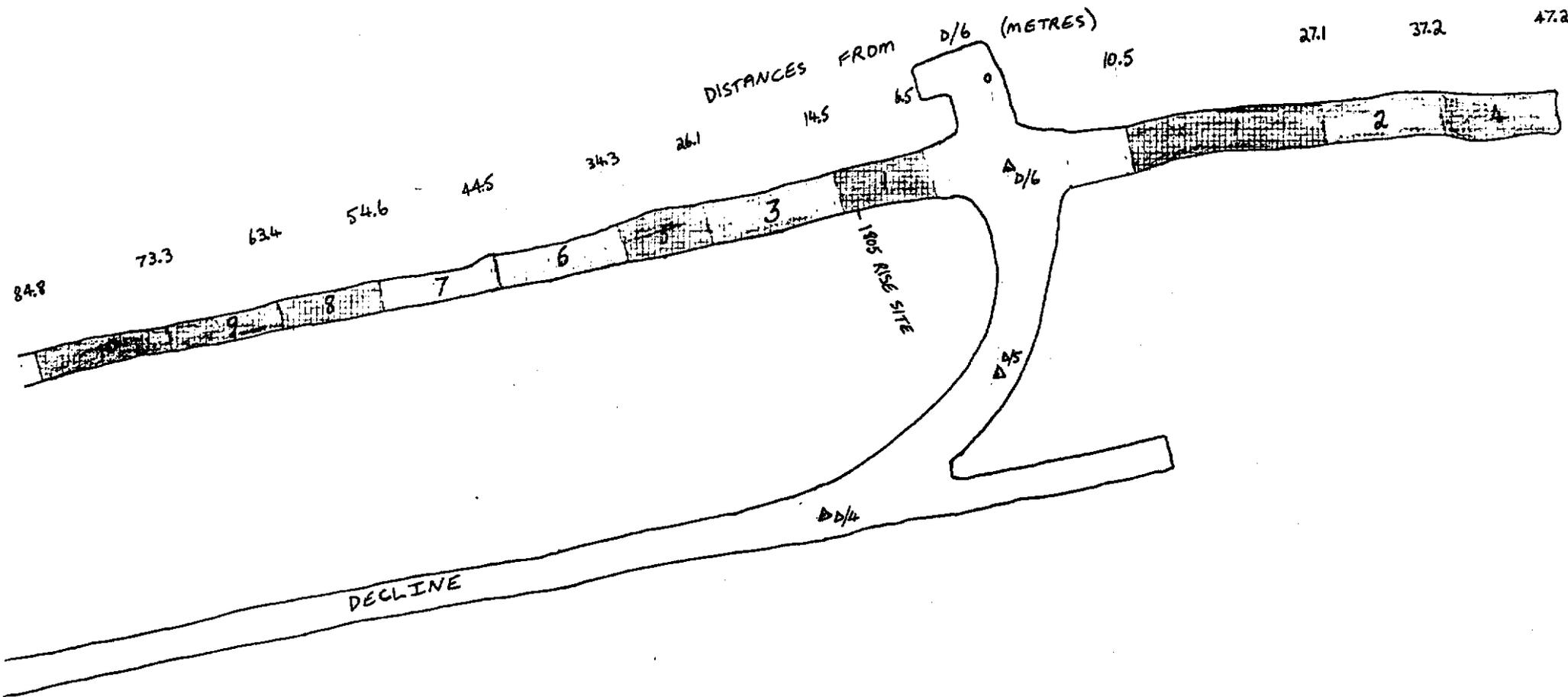
032



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033



Location Plan for 200L Bulk Samples.

Drawn: 13.10.80

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