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MARY ELLEN RESOURCES LTD.

EBY PROPERTY

EBY TOWNSHIP, ONTARIO

**REVERSE CIRCULATION OVERBURDEN DRILLING
AND HEAVY MINERAL GEOCHEMICAL SAMPLING**

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MINING LANDS SECTION



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SUMMARY

This report details the results of a 62-hole reverse circulation overburden drilling/heavy mineral geochemical sampling program that was conducted by Mary Ellen Resources Ltd. on its Eby property near Kirkland Lake in the Abitibi Greenstone Belt of northeastern Ontario. The drill program was designed to evaluate the gold potential of the Kirkland Lake - Larder Lake Fault in the northwest part of the property and that of green carbonate alteration zones in komatiitic flows south of the fault in the northeast part of the property. The Kerr Addison Mine model, which involves syngenetic precipitation of gold in inter-flow sediments followed by epigenetic remobilization into green carbonate zones, was used as a guide. Drill operating costs, excluding road preparation, averaged \$47.98/metre.

The drill area is underlain mainly by Archean volcanic and sedimentary rocks including, from oldest to youngest: 1) komatiitic volcanics of the Larder Lake Group; 2) tholeiitic volcanics of the Kinojevis Group; and 3) graben sediments of the Timiskaming Group. Basalt predominates but significant areas of komatiite and conglomerate-graywacke are present. Other rock units intersected are granodiorite, syenite and Matachewan diabase. The metamorphic grade is sub-greenschist to lower greenschist facies. In the northwest, shearing associated with the Kirkland Lake - Larder Lake Fault is confined to the conglomerate, and green carbonate is rare. In the northeast, the holes were drilled too far south of the fault to intersect green carbonate. The only significant bedrock anomaly (175 ppb Au) occurs in conglomerate intersected in Hole 13. Slightly elevated levels of arsenic occur along the fault in the northwest.

Overburden depth in the drill holes averages 11.05 metres. All preserved Quaternary strata are of Late Wisconsinan to Holocene age; older strata were unprotected during the Wisconsinan glaciation because the area is topographically elevated. The direction of Late Wisconsinan ice flow was south-southeast. Matheson Till deposited by this ice forms a nearly continuous veneer over the bedrock surface and is largely bedrock-derived, making it an excellent geochemical sampling medium. The till is regionally overlain by glaciolacustrine clay and silt of

Lake Ojibway and is locally overlain or supplanted by sand and gravel of the Highway Esker. Holocene peat bogs locally cap the Late Wisconsinan sediments.

Matheson Till on the Eby property and elsewhere along the Kirkland Lake - Larder Lake Fault has a high gold background that is caused mainly by fine, abraded grains of visible gold. Erratic clustering of these fine gold grains and/or the occasional presence of a coarser gold grain produces many spurious heavy mineral gold anomalies. Thirty-nine of forty till anomalies on the property appear to be of this type. The remaining anomaly in Hole 57 probably represents significant gold dispersion from a known green carbonate zone 200 metres to the north.

The northeast corner of the property, the target area of previous work, appears to have the best gold potential and is the only area in which the Kerr Addison Mine model applies. Follow-up exploration including diamond drilling is recommended to test the green carbonate zone north of the Hole 57 overburden anomaly and the anomalous conglomerate intersected in Hole 13. Detailed reverse circulation drilling should be done over the more northerly green carbonate zones that were not tested in the present program.

2.

INTRODUCTION

2.1

Project Outline

From January 19 to February 03, 1987, Mary Ellen Resources Ltd. of the Kasner Group of Companies conducted a detailed program of reverse circulation overburden drilling for the purpose of heavy mineral geochemical sampling on its Eby property in the Abitibi Greenstone Belt near Kirkland Lake, Ontario (Fig. 1, 2, 3). The Kirkland Lake - Larder Lake Fault, a regionally auriferous structure having a close spatial association with the gold deposits at Matachewan, Kirkland Lake and Larder Lake (Kerr-Addison Mine) in Ontario and at Cadillac, Malartic and Val d'Or in Quebec, transects the northwest portion of the property and the main objective of the drilling was to test the overburden along this fault and subsidiary faults for glacially dispersed gold concentrations indicative of economic bedrock mineralization. The program was also designed to develop a diamond drill target in an area of known sub-economic grade mineralization associated with green carbonate alteration south of the fault in the northeast corner of the property.

Mary Ellen Resources contracted Heath and Sherwood Drilling of Kirkland Lake, Ontario, to perform the drilling and Overburden Drilling Management Limited (ODM) of Nepean, Ontario to manage the program. Geologist G. Shelp logged (Appendix A) and sampled the drill holes and supervised the drilling. Field assistance was provided by David Hurd representing the Kasner Group of Companies.

Sixty-two holes were drilled (Plan 1, in pocket). ODM collected samples from Quaternary till, sand and gravel sections and from short sections (generally 1.5 metres) of the underlying bedrock. One hundred and eighty-one overburden samples and 62 bedrock samples were collected (Table 1).

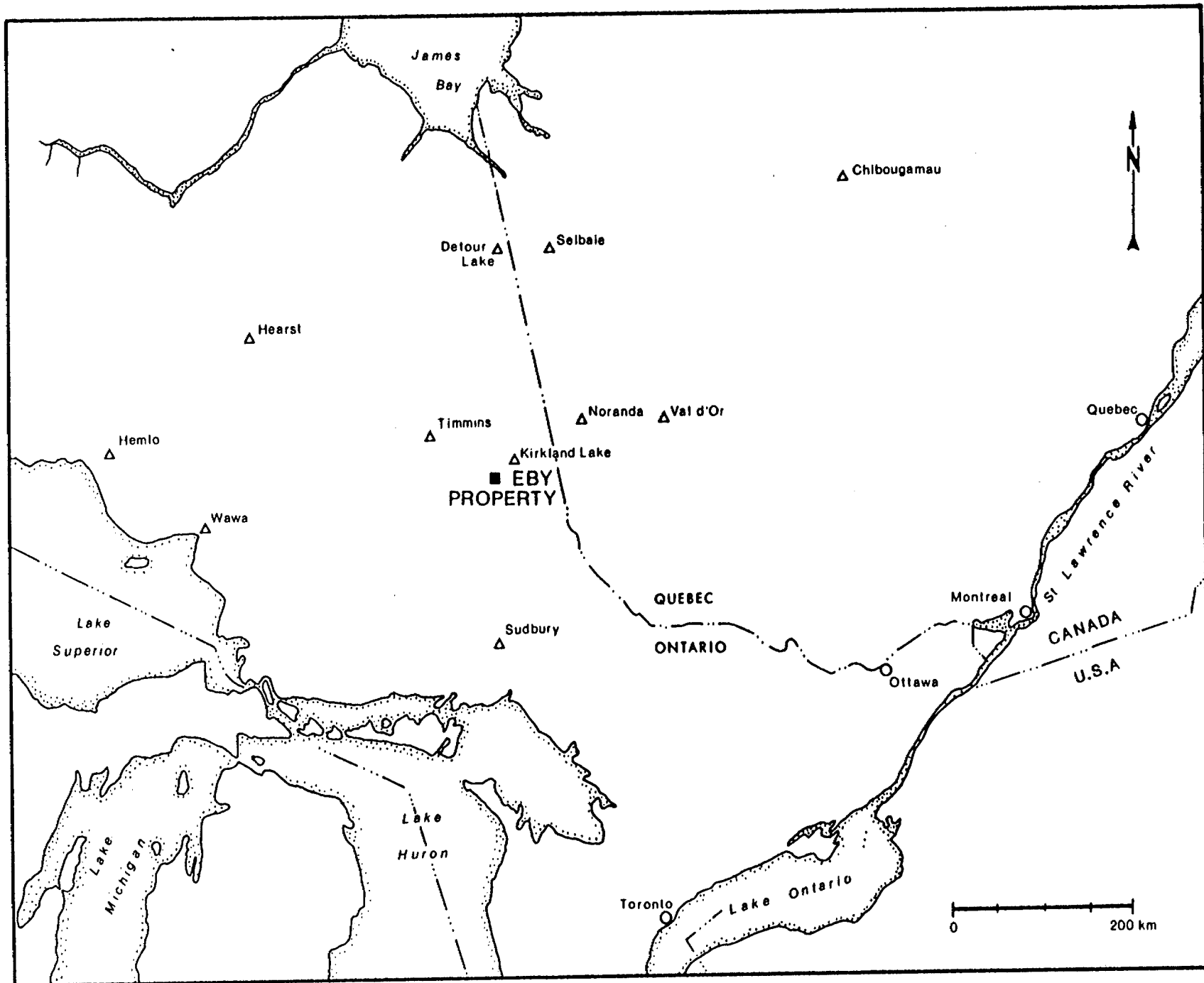


Figure 1 - Eby Property Location Map

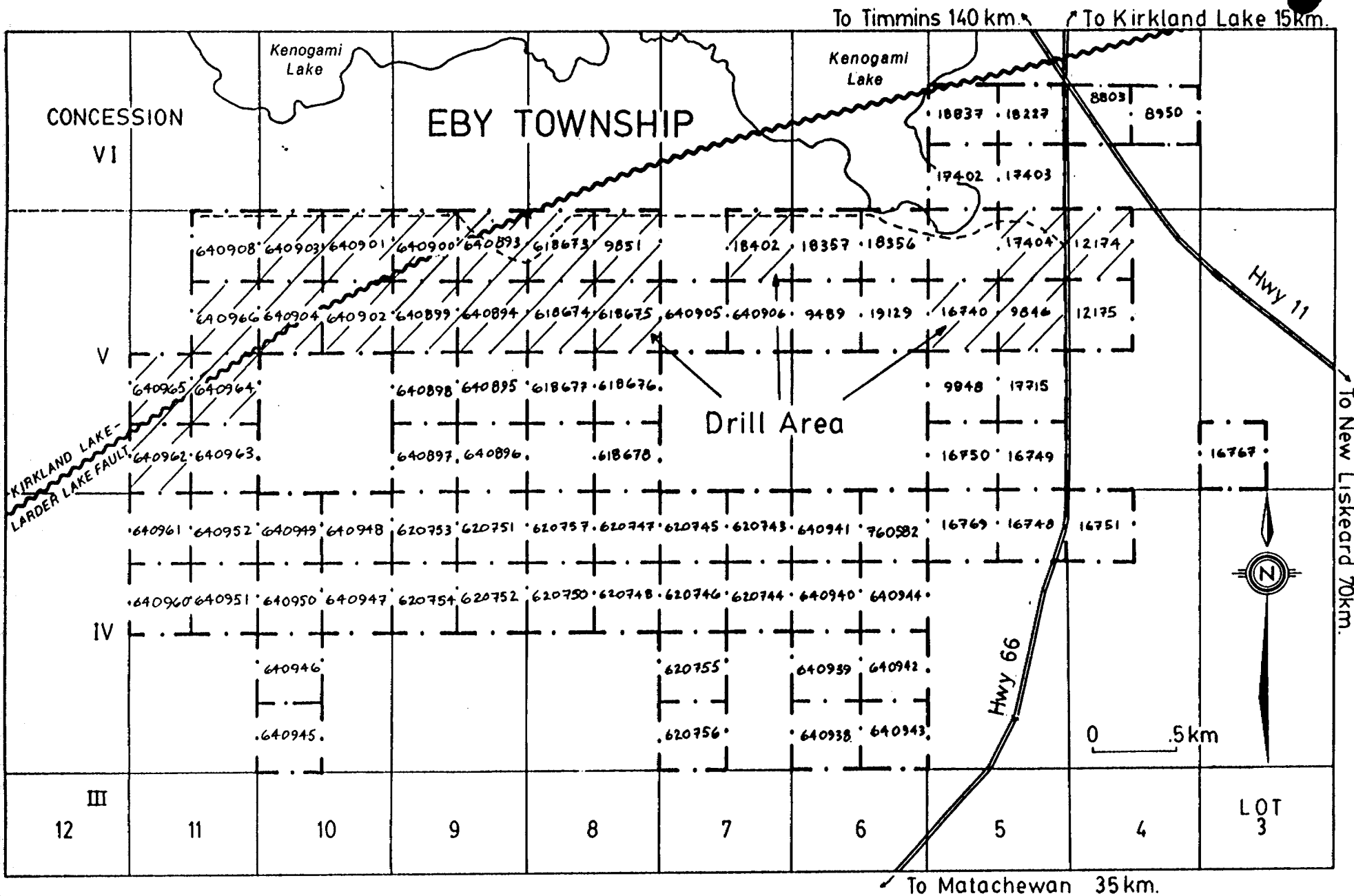


Figure 3 - Claim Map

Hole Number	Coordinates	Metres Drilled		Hole Depth (metres)	Samples Collected	
		Overburden	Bedrock		Overburden	Bedrock
ME-87 01	L 37W; 6+15N	10.0	-	10.0	5	-
02	37W; 4+90N	10.5	1.5	12.0	6	1
03	37W; 3+90N	14.0	1.5	15.5	7	1
04	37W; 2+70N	10.5	1.5	12.0	7	2
05	37W; 1+90N	14.5	1.1	15.6	6	1
06	37W; 0+90N	16.0	1.0	17.0	6	1
07	37W; 0+00N	15.8	1.7	17.5	4	1
08	37W; 1+00S	17.4	1.6	19.0	5	1
09	41W; 2+50N	17.6	1.5	19.1	9	1
10	41W; 1+50N	4.5	2.0	6.5	1	1
11	41W; 0+50N	15.4	1.6	17.0	5	1
12	41W; 0+50S	5.2	1.8	7.0	1	1
13	45W; 2+90N	2.4	1.6	4.0	1	1
14	45W; 1+90N	9.0	1.5	10.5	2	1
15	45W; 0+90N	8.0	1.5	9.5	1	1
16	45W; 0+10S	8.0	1.5	9.5	1	1
17	45W; 1+10S	6.2	1.3	7.5	1	1
18	45W; 2+10S	6.6	2.0	8.6	1	2
19	47W; 0+50S	27.4	1.1	28.5	2	1
20	50W; 0+60S	12.0	-	12.0	3	-
21	50W; 1+60S	13.0	1.5	14.5	3	1
22	50W; 2+60S	11.2	1.3	13.5	2	1
23	50W; 3+60S	11.0	1.5	12.5	1	1
24	54W; 7+25S	5.8	1.7	7.5	1	1
25	54W; 6+00S	6.5	1.0	7.5	2	1
26	54W; 5+00S	8.0	1.5	9.5	1	1
27	54W; 4+00S	12.5	1.5	14.0	6	1
28	54W; 3+25S	5.6	1.4	7.0	2	1
29	41W; 3+50N	11.5	2.0	13.5	7	1
30	33W; 4+00N	8.0	0.5	8.5	4	1
31	L 33W; 2+90N	7.0	1.5	8.5	3	1

Table 1 - Drilling Statistics

Hole Number	Coordinates	Metres Drilled		Hole Depth (metres)	Samples Collected	
		Overburden	Bedrock		Overburden	Bedrock
ME-87- 32	33W; 2+00N	8.0	2.0	10.0	2	1
33	31W; 4+50N	9.0	2.0	11.0	1	1
34	31W; 5+50N	8.8	1.7	10.5	5	1
35	31W; 3+50N	11.8	1.7	13.5	4	1
36	31W; 2+50N	18.0	1.5	19.5	8	1
37	31W; 1+50N	2.7	1.8	4.5	1	1
38	31W; 0+50N	13.5	1.5	15.0	5	1
39	28W; 5+50N	4.6	2.9	7.5	1	2
40	27W; 5+65N	10.0	2.0	12.0	2	2
41	27W; 4+40N	11.3	2.2	13.5	3	1
42	27W; 3+35N	3.0	1.5	4.5	1	1
43	27W; 1+50N	6.6	1.9	8.5	2	1
44	24W; 6+40N	6.4	1.1	7.5	2	1
45	24W; 5+40N	13.0	1.5	14.5	3	1
46	24W; 4+40N	8.4	1.6	10.0	1	1
47	24W; 3+40N	2.0	1.5	3.5	1	1
48	18W; 6+00N	22.0	1.5	23.5	8	1
49	18W; 5+00N	15.4	1.6	17.0	4	1
50	18W; 3+80N	2.5	1.5	4.0	-	1
51	0+25W; 1+00N	22.4	2.1	24.5	9	1
52	7W; 0+50N	3.4	1.1	4.5	1	1
53	6W; 1+10N	5.5	1.1	6.6	-	1
54	5W; 1+50N	12.2	2.8	15.0	5	1
55	4W; 2+25N	5.7	1.8	7.5	1	1
56	3W; 2+75N	10.6	1.5	12.1	1	1
57	2W; 2+25N	18.0	1.5	19.5	1	1
58	1W; 2+40N	14.8	1.6	16.4	1	1
59	1E; 2+75N	29.0	2.5	31.5	2	1
60	2E; 3+50N	19.5	-	19.5	-	-
61	3+50E; 3+50N	11.5	1.5	13.0	1	1
62	2+50E; 3+75N	25.5	-	25.5	-	-
	TOTAL	687.2	93.2	780.4	181	62

Table 1 - Drilling Statistics (cont'd)

Heavy mineral concentrates (Appendix B) were prepared from the overburden samples at ODM's laboratory in Rouyn, Quebec. The bedrock chip samples were logged under a binocular microscope (Appendix D) and their lithologies were related to the established Archean stratigraphy (Plan 1; Fig. 3). Subsamples of the bedrock chips and heavy mineral concentrates were analyzed for gold, arsenic and copper (Appendix E, F). Some of the heavy mineral concentrates were also analyzed for zinc. An independent study consisting of detailed mineralogical logging of the concentrates is being conducted by Jeanette Lourim, consulting geologist.

This report documents all of the work from the drilling program. A detailed analysis of Archean stratigraphy and structure and of Quaternary stratigraphy is included and used as the basis for interpreting the bedrock and heavy mineral gold anomalies.

2.2 Principles of Deep Overburden Geochemistry in Glaciated Terrain

During the Pleistocene epoch of the Quaternary period, the crowns of all ore bodies that subcropped beneath the continental ice sheets of North America were eroded and dispersed down-ice in the glacial debris. The dispersion mechanisms were systematic (Averill, 1978) and the resulting ore "trains" in the overburden are generally long, thin and narrow but most importantly are several hundred times larger than the parent ore bodies. These large trains can be used very effectively to locate the remaining roots of the ore bodies.

Because the dispersion trains originated at the base of the ice, they are either partly or entirely buried by younger, nonanomalous glacial debris. Most trains are confined to the bottom layer of debris deposited during glacial recession-the basal till. In fact, the sampling of glacial overburden for exploration purposes is commonly referred to as "basal till sampling". It is important to note, however, that in areas affected by multiple glaciations the bottom layer of debris in the overburden section may be only the lowermost of several stacked basal tills, and that a dispersion train may occur at any level within any one of the basal till horizons. Consequently, the term "basal till sampling" is not synonymous with the

collection of samples from the base of the overburden section. Moreover, the term is not strictly correct because significant glacial dispersion trains can occur in formations other than basal till.

From the foregoing statements, it can be seen that glacial dispersion and glacial stratigraphy are interdependent. Consequently, the effectiveness of overburden sampling as an exploration method is related to the ability of the sampling equipment to deliver stratigraphic information from the unconsolidated glacial deposits. In areas of deep overburden including most of the Abitibi Greenstone Belt in northwestern Quebec, drills must be used. Most drills have been designed to sample bedrock and are unsuitable for overburden exploration, but in the last fifteen years rotasonic coring rigs and reverse circulation rotary rigs have been developed to sample the overburden as well as the bedrock. Both drills provide accurate stratigraphic information throughout the hole and also deliver large samples that compensate for the natural inhomogeneity of glacial debris.

Reverse circulation rotary rigs are much more widely used in the Abitibi than are rotasonic coring rigs. They employ dual-tube rods and a tricone bit with the outer rod tube acting as a casing to contain the drill water for recirculation and to prevent contamination of samples by material caving from overlying sections. Air and water are injected at high pressure through the annulus between the outer and inner rods to deliver a continuous sample of the entire overburden section through the small inner rod (Fig. 4). The sample is disturbed but returns to surface instantly, and the precise positions of stratigraphic contacts can be identified. Full sample recovery is possible in all formations regardless of porosity or consistency, although sample loss due to blow-out commonly occurs in the first 1 to 3 metres of the hole until a sediment seal is made around the outer rod.

Reverse circulation holes are normally extended 1.5 metres into bedrock. Cuttings of maximum 1 cm size are obtained. These cuttings are used to determine the bedrock stratigraphy, structure and geochemistry and are also compared to the till clasts to help determine ice flow directions and glacial dispersion patterns.

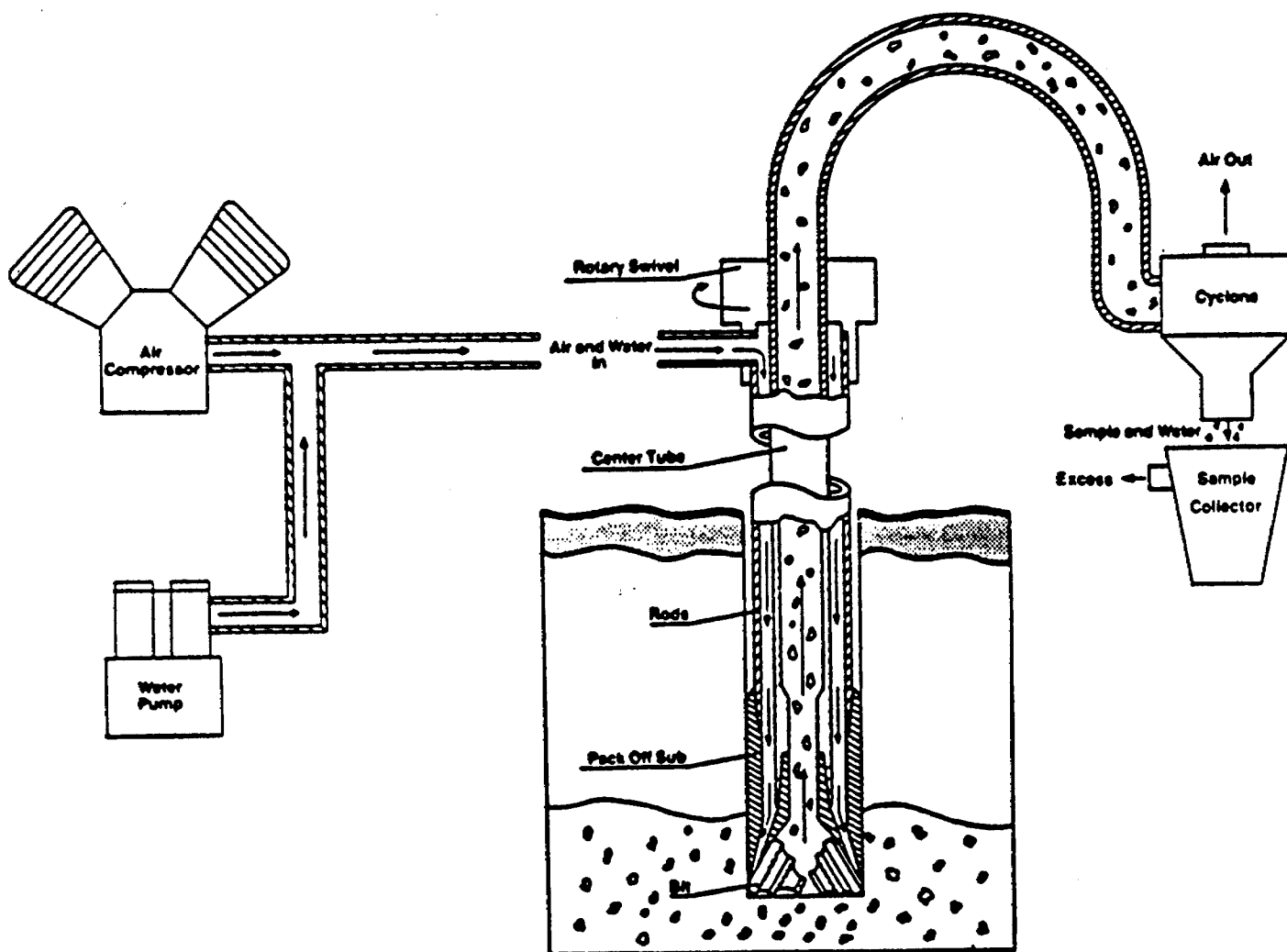


Figure 4 - Schematic of a Typical Reverse Circulation Drilling System

Most of the glacial overburden in Canada is fresh, and metals in the overburden occur in primary, mechanically dispersed minerals rather than in secondary chemical concentrations. While ore mineral dispersion trains are very large, they are also weak due to dilution by glacial transport and are difficult to identify from a normal "soil" analysis of the fine fraction of the samples. Consequently, heavy mineral concentrates are prepared to amplify the primary anomalies, and analysis of the fines is normally reserved for areas where significant post-glacial oxidation is evident. The heavy mineral concentrates are very sensitive, and special care must be taken to avoid the introduction of contaminants into the samples. On gold exploration programs, it is advantageous to separate and examine any free gold particles because most gold anomalies in heavy mineral concentrates are caused by background nugget grains that are of no interest.

2.3 Property Description and Access

The Eby property is located 16 km west of Kirkland Lake (Fig. 1) in Lots 4 to 11, Concessions 4, 5 and 6 of Eby Township (Fig. 3). The property lies south and southeast of Kenogami Lake and is represented by an irregularly shaped claim block comprised of 58 unpatented and 26 patented mining claims totalling approximately 3,480 acres (Tables 2 and 3).

Highway 66 passes in a north-south direction through the eastern end of the property. East-west logging roads provide good access to the drill area which is in the northern third of the property. Tractor roads were cleared to the drill sites.

2.4 Physiography and Vegetation

The Eby property lies within the southern part of the Abitibi Upland physiographic region (Bostock, 1968). The central portion of the Abitibi Upland is represented by a north-sloping clay belt region that was covered by Lake Ojibway

<u>Claim No.</u>	<u>Twp.</u>	<u>Conc.</u>	<u>Lot</u>	<u>Acres</u>	<u>Claim No.</u>	<u>Twp.</u>	<u>Conc.</u>	<u>Lot</u>	<u>Acres</u>
618673	Eby	5	8	40	640902	Eby	5	10	40
618674	Eby	5	8	40	640903	Eby	5	10	40
618675	Eby	5	8	40	640904	Eby	5	10	40
618676	Eby	5	8	40	640905	Eby	5	7	40
618677	Eby	5	8	40	640906	Eby	5	7	40
618678	Eby	5	8	40	640908	Eby	5	11	40
620743	Eby	4	7	40	640938	Eby	4	6	40
620744	Eby	4	7	40	640939	Eby	4	6	40
620745	Eby	4	7	40	640940	Eby	4	6	40
620746	Eby	4	7	40	640941	Eby	4	6	40
620747	Eby	4	8	40	640942	Eby	4	6	40
620748	Eby	4	8	40	640943	Eby	4	6	40
620750	Eby	4	8	40	640944	Eby	4	6	40
620751	Eby	4	9	40	640945	Eby	4	10	40
620752	Eby	4	9	40	640946	Eby	4	10	40
620753	Eby	4	9	40	640947	Eby	4	10	40
620754	Eby	4	9	40	640948	Eby	4	10	40
620755	Eby	4	7	40	640949	Eby	4	10	40
620756	Eby	4	7	40	640950	Eby	4	10	40
620757	Eby	4	8	40	640951	Eby	4	11	40
640893	Eby	5	9	40	640952	Eby	4	11	40
640894	Eby	5	9	40	640960	Eby	4	11	40
640895	Eby	5	9	40	640961	Eby	4	11	40
640896	Eby	5	9	40	640962	Eby	5	11	40
640897	Eby	5	9	40	640963	Eby	5	11	40
640898	Eby	5	9	40	640964	Eby	5	11	40
640899	Eby	5	9	40	640965	Eby	5	11	40
640900	Eby	5	9	40	640966	Eby	5	11	40
640901	Eby	5	10	40	760582	Eby	4	6	40

Table 2 - List of Mary Ellen Mining
Claims, Eby Property

<u>Claim No.</u>	<u>Township</u>	<u>Concession</u>	<u>Lot</u>	<u>Acres</u>
18402	Eby	5	7	40
18357	Eby	5	6	40
9489	Eby	5	6	40
18837	Eby	6	5	40
18227	Eby	6	5	40
8803	Eby	6	4	40
8950	Eby	6	4	40
12174	Eby	5	4	40
12175	Eby	5	4	40
17402	Eby	6	5	40
17403	Eby	6	5	40
17404	Eby	5	5	40
19129	Eby	5	6	40
16740	Eby	5	5	40
9846	Eby	5	5	40
9848	Eby	5	5	40
17715	Eby	5	5	40
16750	Eby	5	5	40
16749	Eby	5	5	40
16767	Eby	5	3	40
16748	Eby	4	5	40
16751	Eby	4	4	40
9851	Eby	5	8	40
16769	Eby	4	5	40
18356	Eby	5	6	40

Table 3 - List of Optioned Mining Claims, Eby Property

10,000 years ago during Late Wisconsinan ice withdrawal. The southern boundary of the clay belt is the Hudson Bay - St. Lawrence River drainage divide. The Eby property is 5-10 km south of the divide and Ojibway clay plain and is 20-25 km north of a smaller clay plain deposited in glacial Lake Barlow. Lakes Barlow and Ojibway were briefly joined over the drainage divide (Vincent and Hardy, 1979), and during this period all bedrock valleys on the Eby property were infilled with glaciolacustrine sediments. As a result, overburden thickness in the drill area is extremely variable, ranging from 2 to 29.0 metres with an average of 11.05 metres.

All creeks and swamps in the drill area drain into the Blanche River system which flows southward into Lake Timiskaming, the largest remnant of glacial Lake Barlow. The property is generally flat with an approximate elevation of 320 metres ASL. Relief of 10 to 15 metres occurs on bedrock outcrops that are clustered in the south-central and northeastern parts of the property (Plan 1).

2.5

Previous Work

Mary Ellen Resources acquired the Eby property in January, 1982, to cover a potential source area for gold anomalies detected during a 1980-1981 reverse circulation drilling program conducted by the Ontario Geological Survey (Routledge et al, 1981).

Prior to Mary Ellen's acquisition of the property, the geology of Eby Township was mapped in detail by Lovell (1972). His report also includes a summary of previous exploration. A synopsis of the work pertaining to the Mary Ellen property has been compiled by Stewart Carmichael, geologist for the Kasner Group of Companies. This work was focussed on several green carbonate zones lying immediately south of the Kirkland Lake - Larder Lake Fault on the six claims east of Kenogami Lake (Fig. 3). Some work was also done along the presumed southwestern extension of the fault on the south side of the lake. The most encouraging gold intersections, 0.702 oz/ton over 10 feet and 0.64 oz/ton over 2 feet, were obtained by Beaucoeur Mines in diamond drill holes on a green carbonate

zone on Claim 18227. Green carbonate is a favourable gold indicator in the region, particularly at the Kerr Addison Mine at Larder Lake.

In 1979, the Ontario Geological Survey conducted a regional airborne electromagnetic and magnetic survey (Pitcher 1979) as part of the Kirkland Lake Incentive Program (KLIP). Four programs of reverse-circulation drilling followed. Three of the KLIP holes (80-24, 81-01 and 81-03) were drilled on the Eby property (Routledge, 1981). Anomalous gold was detected in the heavy mineral fraction of samples from Hole 80-24 and in the minus 63 micron fraction of samples from Hole 81-03.

In February 1983, Mary Ellen Resources conducted an airborne E.M. and magnetometer survey which was successful in verifying the presence of the Kirkland Lake - Larder Lake Fault. Subsequent geological mapping (Carmicheal, 1985) revealed the presence of interflow carbonate sediments and iron formation similar to those hosting much of the gold at the Kerr Addison Mine (Jensen and Hinse, 1979). In 1985, line cutting, detailed geological mapping, ground geophysics and humus sampling were carried out. In November of the same year, Mary Ellen Resources undertook a 9 hole (4008 feet) diamond drill program on Claims 18227 and 17403 in the green carbonate area to test an I.P. anomaly and a zone where anomalous gold values had been reported previously. Five of the nine holes intersected sub-ore grade mineralization.

3.

DRILLING AND SAMPLING

3.1

Drill Hole Pattern

Twelve of the sixty-two reverse circulation holes were drilled on the southern edge of the green carbonate area at the southeast corner of Kenogami Lake. These holes were positioned on one east-west traverse, roughly perpendicular to the 165 degree azimuth (Baker 1980) of Late Wisconsinan ice movement. The balance of the holes were drilled south of the lake along the projection of the Kirkland Lake -

Larder Lake Fault and were positioned on north-south sections to cross the fault at intervals of 200 to 500 metres. The hole separation along all traverses was 100 metres.

The above drilling pattern should provide good overburden exploration coverage because most known gold dispersion trains (Table 4) are 300 to 1000 metres long and 100 to 400 metres wide. It should also provide good bedrock exploration coverage because the traverses cross rather than follow the bedrock strata and structures.

3.2

Drilling Equipment

Heath and Sherwood's drill rig employed an Acker MP drill head with a 3 metre feed cylinder. The drill, together with all its ancilliary equipment including air compressor, water pump and logging and sampling facilities, was unitized and enclosed on the bed of a Nodwell Model FN-160 tracked carrier for all-terrain mobility and all-weather operation.

The rig employed an air compressor with a rated capacity of 300 c.f.m. at 160 p.s.i. and a water pump having a capacity of 20 g.p.m. at 600 p.s.i. pressure. Water flow was normally restricted to 4-5 g.p.m. to improve recovery of fines. The rig was equipped with a 110 volt generator and Cool White fluorescent fixtures that simulate natural sunlight for accurate sample logging. All equipment except the air compressor and Nodwell carrier was operated hydrostatically from a central diesel engine.

The rig carried twenty-two 10-foot drill rods. The holes were logged in metres using the approximate conversion factor of 3 metres to 10 feet. This resulted in the logged hole depth being 1.6 percent less than true depth.

Heath and Sherwood supported the drill rig with a Bombadier B-15 Muskeg tractor equipped with a 400-gallon, exhaust-heated water tank. Road clearing was done by Glen Kasner of the Kasner Group of Companies using a Caterpillar D7 bulldozer.

PROVINCE	GOLD DEPOSIT	TRAIN LENGTH ¹ (m)	
		TRACED	EST. TOTAL
Saskatchewan	Lake "X" ²	300	300
Saskatchewan	Star Lake	300	800
Saskatchewan	Lake "Y"	500	1000
Saskatchewan	Waddy Lake ²	600	2000
Ontario	McCool	300	400
Quebec	Cooke Mine ³	800	1000
Quebec	Golden Pond West	300	400 ⁴
Quebec	Golden Pond	400	500 ⁴
Quebec	Golden Pond East	100	1000

- 1 - Based on minimum 10 gold grains of similar size and shape per 8 kg sample for free gold trains and on coincident high gold and base metal assays for invisible gold trains
- 2 - Deposit oriented parallel to glacial ice advance
- 3 - Invisible gold deposit
- 4 - Train foreshortened by erosion in last ice advance

Table 4 - Heavy Mineral Gold Dispersion Trains Identified by Overburden Drilling Management Limited Laboratory

3.3

Drill Performance

Drilling on the Eby property started on January 19, 1987 and was completed on February 3, 1987, for a total of 16 drilling days. The drill usually operated on one 10-hour shift per day but the shift was lengthened or shortened at the discretion of the field geologist. A broken axle and transmission problems on the FN-160 Nodwell resulted in two and a half days of down-time. Minor delays were caused by track problems on the muskeg tractor and by the replacement of a fuel injector on the Nodwell.

Sixty-two holes were drilled for a total of 687.2 metres of overburden and 93.2 metres of bedrock (Table 1). All holes except Holes 01, 20, 60 and 62 reached bedrock. Production averaged 48.8 metres per day or 6.0 metres per operating hour. Chargeable (productive) drill hours amounted to 115 and mechanical downtime to 29 hours or 25.2 percent. Drilling costs, excluding road clearing, averaged \$47.98/metre (\$14.63/foot).

3.4

Logging and Sampling

Overburden samples were collected in two 20 litre buckets coupled with a plastic tube. This procedure ensures a quiet settling environment thus reducing the loss of fines encountered if only one bucket is used and allowed to overflow. Most of the clay is still lost but a recent research study made by ODM (Dimock, 1985) showed that sand loss is insignificant and silt loss is reduced to 40 percent compared to 72 percent with the one-bucket system. Interestingly, fine gold is lost in direct proportion to fine minerals of low specific gravity such as quartz and feldspar because the flake shape rather than high density of fine gold is the primary factor controlling the rate of settling. Further research conducted by ODM (Kurina, 1986) on various inlet/outlet attachments on the second bucket showed an additional 33 percent of fine material could be retained by utilizing a horizontally curved inlet tube for spiral flow and a vertical stack skimmer on the outlet. The two-bucket system with the modified flow configuration was employed on the Mary Ellen program.

ODM employed a 10-mesh (1700 micron) screen over the first bucket to separate and discard the majority of rock cuttings and thereby increase the proportion of matrix material needed to identify and trace dispersion trains. The +10 mesh rock cuttings were constantly monitored to discern any variations which could give clues to overburden stratigraphy, or for any clasts indicative of an environment suitable for gold or base metal mineralization. Approximately 20 percent of the cuttings were kept for future reference. The degree of sorting of the -10 mesh matrix was monitored to differentiate till from sand and gravel.

Till units were sampled continuously using an average sample interval of 1.5 metres. Glaciofluvial and related sand and gravel were sampled over longer, 3 to 6 metre intervals because they are far-travelled and thus generally ineffective for mineral tracing. Glaciolacustrine clay and silt were not sampled because they are of no exploration value.

In the field, both the overburden and bedrock samples were assigned a number denoting the company (ME), the year (87), the position of the hole in the drilling sequence and the position of the sample in the drill hole. Thus a designation such as ME-87-35-03 indicates the third sample collected from the thirty-fifth drill hole.

Following collection, the overburden samples were reduced to 7-9 kilograms with an aluminum scoop, packed in heavy plastic bags and shipped in 20-litre metal pails to the ODM processing laboratory in Rouyn, Québec.

3.5

Sample Processing

ODM's processing procedures for the overburden samples are illustrated in the flow sheet of Figure 5 and may be summarized as follows:

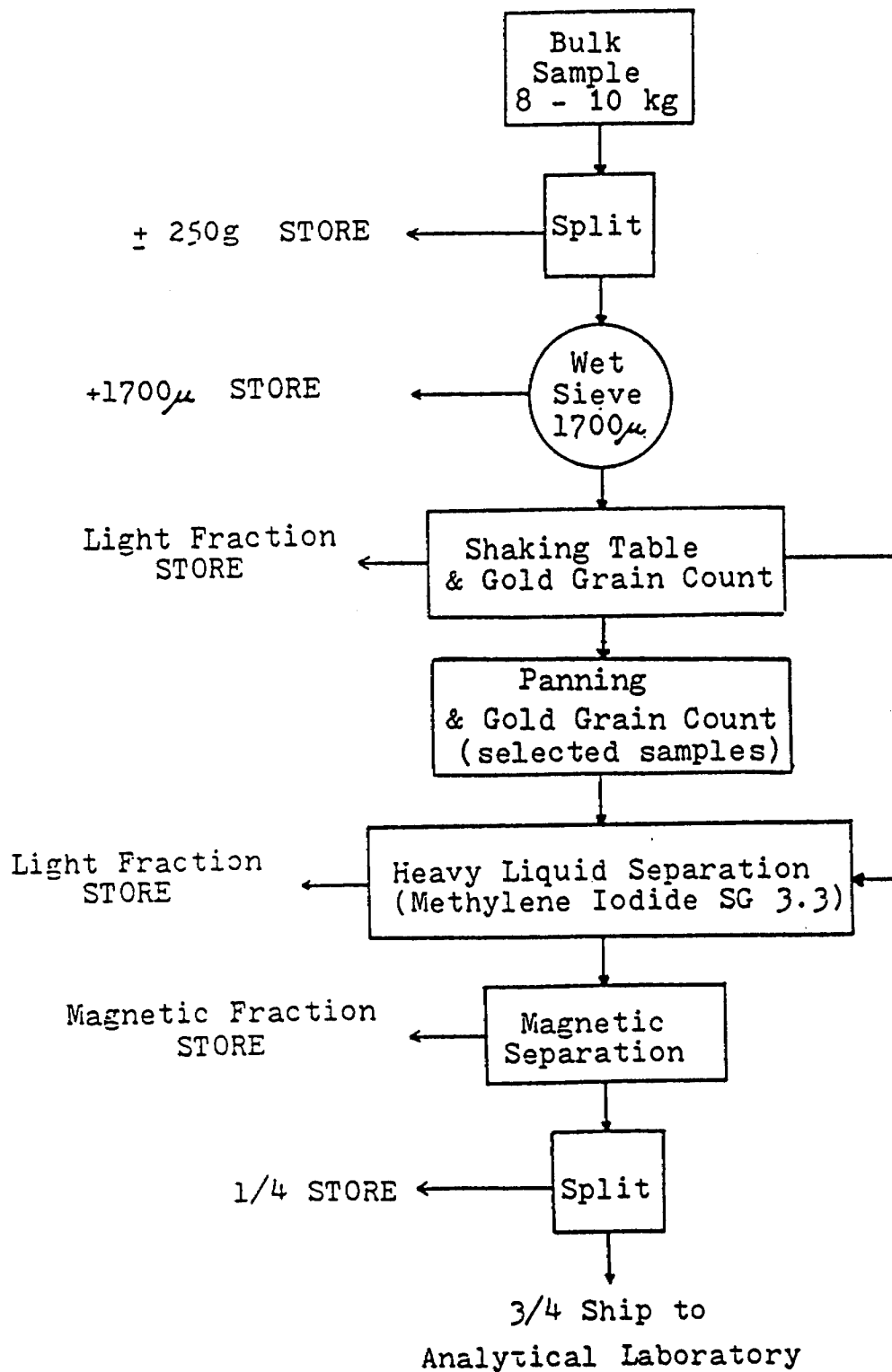


Figure 5 - Sample Processing Flow Sheet

First, a 250 gram character sample is extracted from the bulk sample using a tube-type sampler. The character sample is dried and stored for future reference. On some gold programs, its minus 250 mesh fraction is separated and analyzed to check for occluded gold that may not be recovered in the heavy mineral concentrates.

The remainder of the bulk sample is weighed wet and is sieved at 1700 microns (10 mesh). The +1700 micron clasts are weighed wet and the -1700 micron matrix is processed on a shaking table to obtain a preconcentrate. The table concentrate and all fractions obtained from it are weighed dry. The Mary Ellen sample weights are listed in Appendix B.

ODM has developed technology for evaluating free gold anomalies as the samples are being tabled. The use of special feeders and table adjustments causes many gold grains to separate from the other heavy minerals and follow individual paths across the table. These grains are picked from the deck, placed under a binocular microscope, measured to obtain an estimate of their contribution to the eventual assay of the concentrate (Table 5), and classified as delicate, irregular or abraded (Fig. 6) to determine their approximate distance of glacial transport. Photomicrographs (35 mm slides) are taken if more than 10 gold grains are present.

Magnetite, with a Specific Gravity of 5.2, is the heaviest of the common minerals and normally forms the top mineral band on the table above garnet and epidote/pyroxene. Common flake gold coarser than 125 microns separates completely from the magnetite and is readily counted. Fine gold, thick gold and delicate gold travel with the magnetite due to size and shape effects, and only 10 to 20 percent of such grains can be sighted on the table. Gold particles can also be obscured by pyrite which tends to cross the table in the gold path if it forms more than 10 percent of the concentrate. However, ODM has developed a special panning technique to recover the hidden particles together with some copper, lead and arsenic pathfinder minerals. ODM normally pans samples in which two or more gold particles are sighted on the table as well as samples with high pyrite concentrations or any delicate gold. The Mary Ellen table and pan gold counts are listed in Appendix C.

<u>Size Classification</u>	<u>Flake Diameter (microns)</u>	<u>ppb Au</u>
Very Fine	50	10
"	100	100
Fine	150	330
"	200	760
Medium	300	2,400
"	400	5,400
"	500	10,000
Coarse	600	16,200
"	700	24,000
"	800	33,300
"	900	43,700
"	1,000	55,000
Very Coarse	1,000+	55,000+

Table 5 - Geochemical Contribution of One Gold Grain to a Fifteen Gram Sample

DELICATE

0-100 m ice transport.
Primary crystal faces, pitted leaf
surfaces & ragged leaf edges intact.



IRREGULAR

100-1000 m ice transport.
Gross primary shape
and pitted surface
intact.



IRREGULAR

Curled leaf variety.



ABRADED

1000+ m ice transport.
Large primary leaf
reduced to smaller
flakes with polished
surfaces.



ABRADED

Spindled leaf variety.



ROUNDED

1000+ m ice + stream transport.
Polished equidimensional grains.

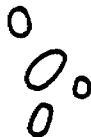


Figure 6 - Effects of Glacial Transport on Gold Particle Size and Shape
(Developed by Overburden Drilling Management Ltd.)

The table and pan concentrates and any gold grains are recombined and the concentrate is dried. A heavy liquid separation in methylene iodide (Specific Gravity 3.3) is then performed. The light fraction (S.G. less than 3.3) is stored and the heavy fraction undergoes a magnetic separation to remove drill steel and magnetite. The Mary Ellen magnetic separates were checked to ensure that they contained not more than five percent pyrrhotite.

3.6 Sample Analysis

The heavy mineral concentrates were analyzed by Assayers (Ontario) Ltd. of Toronto. A 1-gram subsample was used for Cu-Zn-As analysis and the balance of the concentrate was used for gold analysis. No pulping was required. Gold was analyzed using fire assay preparation and atomic absorption finish. Base metals (Cu and Zn) and As were analyzed by inductively coupled plasma (I.C.P.).

The bedrock samples were analyzed by Swastika Laboratories Ltd. of Kirkland Lake. Subsamples of the bedrock chips were homogenized by pulping in a shatterbox and were analyzed for gold by the fire assay method with atomic absorption finish. Copper and arsenic were analyzed using atomic absorption and colourimetry, respectively.

4. BEDROCK GEOLOGY

4.1 Regional Geology

The Eby property is in the southwestern portion of the Abitibi Greenstone Belt. The Abitibi Belt is of Archean age (2,700-2,750 million years old) and comprises repeated komatiitic through tholeiitic to calc-alkalic cycles of lavas, volcanoclastics, porphyries and layered basic-ultrabasic intrusions with coeval clastic sedimentary rocks and intrusives of potassium poor dioritic to tonalitic composition. These rocks have been complexly deformed and metamorphosed to

the sub-greenschist or greenschist facies and intruded by late kinematic granodiorite and monzonite plutons (Gariépy, Allègre, Lajoie, 1984).

The volcanic rocks in the southwestern part of the Abitibi Belt are preserved in an east plunging synclinorium between the Lake Abitibi Batholith in the north and the Round Lake Batholith in the south (Jensen, 1985). The Eby property is on the south limb of the synclinorium (Fig. 2) where two main cycles of volcanism are recognized. The property is underlain by komatiitic and tholeiitic volcanics of the Larder Lake Group and Kinojevis Groups respectively, and by unconformably overlying sediments of the Timiskaming Group. All three groups belong to the upper volcanic cycle, Cycle II. Metamorphic grade is sub-greenschist.

In the Kirkland Lake area, the south limb of the synclinorium is cut by a large scale, east-west trending structure known as the Kirkland Lake - Larder Lake Fault. Following peak metamorphism of the Abitibi Belt 2,650 ± years ago, the Kirkland Lake - Larder Lake Fault acted as a major hydrothermal conduit which resulted in the deposition of significant gold mineralization at Matachewan, Kirkland Lake and Larder Lake in Ontario and at Cadillac, Malartic and Val d'Or in Quebec. This fault transects the northwestern part of the Eby property and also passes through or near the northeast corner of the property where it crosses komatiite flows of the Larder Lake Group, resulting in the formation of considerable "green carbonate" (Fe/Mg carbonate mixed with fuchsite). A similar relationship occurs at the Kerr-Addison Mine and other gold deposits in the Larder Lake area.

Although it is known mainly as a late-Abitibi gold conduit, the Kirkland Lake - Larder Lake Fault was also active at other stages of the evolution of the Abitibi Belt, most notably during Cycle II when it was a basin-marginal subsidence zone (Jensen, 1985) and at the culmination of Cycle II when it received graben sediments of the Timiskaming Group.

Lovell (1972) mapped a number of small intrusive bodies of several lithologies on or near the Eby property. In order of decreasing age, these intrusives are:

1. Na-rich trondhjemite and quartz diorite satellites of the Round Lake batholith which lies 5 km south of the property and is probably coeval with the Cycle II volcanics (Jensen, 1985).
2. K-rich syenite, granodiorite and quartz monzonite satellites of the Otto stock which lies 7 km to the southeast and is probably coeval with the sediments of the Timiskaming Group.
3. North-south trending diabase dikes of the Matachewan swarm which are of Late Archean to Early Proterozoic age (2,485 million years, Fahrig and Wanless, 1963; 2,690 million years, Gates and Hurley, 1973).

Northeast and west of the property, the Abitibi Belt is covered by Proterozoic sediments of the Cobalt Group. These sediments, which include extensive tillite and varved sequences of glacial origin, are cut by diabase dikes of Keweenawan age (1400 ± million years).

4.2 Eby Bedrock Logging Procedures

A binocular microscopic log of all bedrock samples was prepared (Appendix D) to confirm and amplify field descriptions with the objective of producing an accurate stratigraphic map. Particular attention was paid to primary features, and the rocks were assigned genetic names such as intermediate volcanics rather than metamorphic names such as chlorite-carbonate schist.

Reasonably accurate measurements of primary mineralogy, structure, texture, degree of metamorphism and alteration can be made from chip samples with a binocular microscope, but inherent limitations are present. These limitations include:

1. Inability to differentiate gray plagioclase from pale gray-brown and gray-green pyroxene where the grain size is less than 0.1 mm as in many volcanic rocks. This often precludes differentiation of intermediate volcanics from mafic volcanics in the Abitibi belt as extensive areas have undergone only sub-greenschist facies metamorphism resulting in the preservation of primary pyroxene. In greenschist facies areas where pyroxene has been largely converted to darker green amphibole and chlorite, intermediate and mafic units can be differentiated.
2. Inability to determine bedding thickness or fragment size where the dimensions of the beds or fragments are greater than the 1 cm diameter of the coarsest drill cuttings.
3. Inability to recognize tops in bedded sections.
4. Difficulty in differentiating certain primary structures such as pillow selvages from secondary veins and shears.
5. Necessity of inferring gross mineralogy of aphanitic samples from rock colour and hardness.

4.3 Bedrock Lithology of the Reverse Circulation Drill Holes

The following six rock units were intersected in the Eby reverse circulation drill holes:

- 1) Ultramafic volcanics (komatiitic flows)
- 2) Mafic volcanics (basaltic flows)
- 3) Metasediments (conglomerate, graywacke)
- 4) Granodiorite
- 5) Syenite
- 6) Diabase

The distribution of these rock units (Plan 1) is very similar to that mapped by Carmichael (1985). Basalt is the dominant rock type, occurring in 66 percent of the holes, followed by komatiite, conglomerate and the intrusive rocks. The main mafic/ultramafic contact occurs in the northeast corner of the property and represents the boundary between the Kinojevis and Larder Lake Groups. The contact between the mafic volcanics and sediments roughly coincides with the Kirkland Lake - Larder Lake Fault in the northwestern part of the property and represents the boundary between the Kinojevis and Timiskaming Groups. Small intrusive bodies including granodiorite, syenite and Matachewan diabase are randomly distributed on the property.

The low grade of regional metamorphism and the intense shearing and hydrothermal alteration along the Kirkland Lake - Larder Lake Fault described by previous workers are all evident in the drill holes. Basalt samples south of the fault are massive to weakly foliated and contain either primary pyroxene or pale green chloritic pseudomorphs of pyroxene. Conglomerate samples along the fault invariably show some degree of foliation, schistosity, shearing, brecciation or quartz-Fe/Mg carbonate veining.

4.3.1 Ultramafic Volcanics (Unit 1)

Ultramafic volcanics were intersected in six of the 58 drill holes in which a bedrock sample was obtained. All six intersections occur in the northeast corner of the property in the area previously mapped as the Larder Lake Group. The rocks range in colour from a mottled dark green and white to dark green and have an average grain size of 0.2 mm. The original massive structure is present only in Hole 57 where a distinct pyroxene spinifex texture has been preserved. The remaining ultramafic samples are well foliated to schistose, resulting in a masking of the original grain size and texture of the rocks.

Two distinct mineralogical phases of komatiite are present; peridotitic and pyroxenitic komatiite. According to Jensen (1985), such variations are normal within a single komatiitic flow. The peridotitic komatiite samples (e.g. Hole 54) consists of 60-75 percent talc, 5-25 percent Fe/Mg carbonate, 5-10 percent chlorite and 7-10 percent magnetite (a product of the breakdown of primary olivine). In contrast, the pyroxenitic komatiite samples (e.g. Hole 58), consist of 20-60 percent talc, 30-40 percent chlorite, 5-25 percent Fe/Mg carbonate and less than 0.1 percent magnetite. An exception is the spinifex-textured sample of Hole 57 which consists of 30-50 percent acicular pyroxene phenocrysts (1-5 mm in size) shot through a groundmass composed of 70-80 percent pyroxene, 20 percent plagioclase, and 3 percent magnetite.

The Fe/Mg carbonate in the komatiite is probably a product of the breakdown of olivine and pyroxene during dewatering of the lava as the samples do not contain any indicators of hydrothermal alteration such as fuchsite, pyrite, tourmaline, quartz-carbonate veins or bleached zones.

4.3.2 Mafic Volcanics (Unit 2)

Mafic volcanics were intersected in 40 drill holes. The primary fabric of most samples is typical of basaltic flows, ranging almost randomly from aphanitic to medium/coarse-grained (0.2-0.5 mm) and massive. Samples having a grain size of less than 0.1 mm often contain 1 to 5 percent unstretched amygdules that are variably filled with quartz, chlorite, calcite or pyrite. Hyaloclastic (flow-top breccia) structure is rare (Hole 17) and many samples contain both coarse and fine-grained phases suggesting small-scale chilling more consistent with pillowed flows than with thick, texturally-zoned flows.

Most of the aphanitic samples are pale to medium green in colour and consist of a hypocrystalline mixture of glass and pyroxene or plagioclase needles. Hyaloclastite fragments, where recognized, are coarser-grained (0.05-0.1 mm). Cooling cracks are often evident in the glass, and amygdules may be present. Sometimes the feldspar or pyroxene needles in the glass are concentrated in variolitic clusters.

Coarse-grained massive basalt, representing the lower or central portions of flows, is a darker green colour becoming speckled green and white in very coarse-grained samples. Some samples contain 1 to 10 percent pyroxene phenocrysts, locally with ophitic habit, and others are sub-diabasic but most have a hypidiomorphic (equigranular, interlocking) texture. The gross texture of the coarsest samples resembles that of gabbros. In detail, however, the pyroxene and plagioclase are seen to occur in the elongate lath form that characterizes basalt flows.

Mineralogically the basalt consists of 40 to 70 percent pyroxene with the only other major mineral being plagioclase. In the aphanitic samples, the pyroxene and plagioclase could be differentiated only in hyaloclastite fragments, if at all (of course, in hyaloclastites, the matrix would have the same composition as the fragments), and it was often necessary to infer the composition on the basis of colour and diagnostic basaltic textures such as variolites. Quartz was seen in only a few samples. Leucoxene is the most common accessory mineral, occurring in concentrations of 1 to 5 percent in many of the coarse-grained samples. A few samples contain 2 to 10 percent magnetite. These samples occur in Holes 48, 49, 52 and 53 located along the Kinojevis/Larder Lake contact (Plan 1) and represent a normal transition from komatiitic to tholeiitic volcanics. Most samples contain trace to 0.3 percent pyrite, or less commonly pyrrhotite.

Three colour phases of pyroxene can sometimes be seen in the more massive samples. In order of decreasing abundance they are pale green, pale brown and dark green. The dark green variety is the most likely to form phenocrysts and, in deformed samples, is the first to be altered to chlorite while the pale brown variety is the most stable. The chlorite generally pseudomorphs the pyroxene in colour as well as habit and is recognized only by its softness. Chloritization of pyroxene is accompanied by albitization of plagioclase which results in the formation of 1 to 5 percent disseminated calcite, and up to 20 percent visible saussurite flecks.

Basalt intersections along the Kirkland Lake - Larder Lake Fault show little evidence of deformation and hydrothermal alteration whereas the conglomerate to the north is sheared and altered. In a few isolated intersections, the brittle basalt has been brecciated, bleached and replaced by a stockwork of carbonate or quartz-carbonate veins, locally with strong silicification (Hole 30). Most of the carbonate reacts either moderately or very slowly with dilute hydrochloric acid, indicating a composition ranging from dolomite to ankerite. The total percentage of disseminated plus vein Fe/Mg carbonate is contoured on Plan 1.

4.3.3 Metasediments (Unit 3)

Metasediments were intersected in eight holes along the Kirkland Lake - Larder Lake Fault in the northwest corner of the property. These metasediments belong to the Timiskaming Group, and are slightly younger than all other volcanic and sedimentary rocks of the Abitibi greenstone belt (Jensen, 1985). Their confinement to the fault zone indicates that this fault, which is known mainly for its role as an enormous hydrothermal conduit during the waning stages of folding and metamorphism, was also a major graben structure during sedimentation.

Because the sediments are proximal to the fault, most of the samples are highly deformed and altered. However, the preservation of primary structures and textures is sufficient to allow recognition of the following subunits:

3 a - conglomerate

3 b - graywacke

Conglomerate predominates and was intersected in seven of the eight drill holes. The rock is varicoloured with the matrix exhibiting a pale gray-green colour and the clasts a bleached (pale) green or red colour. The predominant green colour of these rocks is a reflection of the samples being composed largely of locally derived tholeiitic basalt belonging to the Kinojevis Group. The bleaching of these basalts most likely occurred during sedimentation.

The conglomerate samples have a gravelly texture which varies from clast supported with up to 90 percent clasts to matrix supported with greater than 90 percent coarse sand. The matrix generally consists of 85-90 percent volcanic lithics, 5-10 percent quartz, 1-5 percent chlorite and a trace of jasper and fuchsite. Sericite is locally present (up to 20 percent) in the more deformed samples. The clasts component is composed mainly of basaltic rock fragments with varying amounts of granodiorite and "feldspar porphyry". The granodiorite clasts generally represent less than 10 percent of the sample but rise to 50 percent in Holes 14 and 21 which were drilled near a small granodiorite stock. This suggests that the stock is older than the sediments. Hematization which most likely occurred during diagenesis has imparted a pervasive red stain to the conglomerate making clast lithology identification more difficult. The "feldspar porphyry" clasts tend to be particularly susceptible to hematization and may actually be K-rich porphyritic trachytes. This lithology is common in the conglomerate east of Kirkland Lake where Timiskaming sedimentation was accompanied by alkalic volcanism. Hydrothermal alteration has added up to 10 percent Fe/Mg carbonate, a trace of fuchsite and trace to 0.5 percent pyrite to the conglomerate.

Greywacke was encountered only in Hole 16 where the character of the sediment is largely masked by severe deformation, buff-yellow bleaching and 40 percent quartz-carbonate veining. The least deformed chips consist of 70 percent pale green, sand-sized (0.2-0.5 mm) volcanic lithics (the same as those present in the conglomerate), 10-15 percent quartz sand and 15 percent matrix chlorite. Fe/Mg carbonate constitutes 10 percent of the sample. Only a trace of pyrite is present.

4.3.4 Granodiorite (Unit 4)

Granodiorite was intersected in Holes 18, 19 and 40 located immediately south of the fault zone in the northwestern part of the property. The Hole 18 and 19 intersections appear to define a small stock while the Hole 40 intersection probably represents a dike.

The granodiorite is pale pink to buff in colour and porphyritic exhibiting a grain size of 0.05-0.2 mm for the groundmass and 1.0-3.0 mm for phenocrysts. Structurally, the rock is massive to weakly foliated in Holes 18 and 40 and strongly sheared in Hole 19 where the groundmass and many of the feldspar phenocrysts have been reduced to a buff sericitic mylonite.

Mineralogically, the samples consist of 20-50 percent feldspar phenocrysts in a groundmass composed of 60-70 percent pink to gray feldspar, 20 percent chlorite and 10-20 percent quartz. Slow-reacting Fe/Mg carbonate constitutes 3-15 percent of the groundmass and also occurs as disseminations and fracture fillings. Sulphides are present as disseminated pyrite or as pyrite associated with quartz-carbonate veinlets and constitute less than 1 percent of the sample.

The name "granodiorite" is used very loosely here as much of the feldspar is pink suggesting a quartz monzonite or trondhjemite composition. A trondhjemite composition would help explain the presence of "granodiorite" clasts in the Timiskaming conglomerate as trondhjemites in the area tend to be associated with the older plutonism that produced the Round Lake batholith.

4.3.5 Syenite (Unit 5)

Syenite was intersected in Holes 51 and 56 in the northeast corner of the property where it appears to form small dikes cutting the Larder Lake komatiites. The syenite in Hole 56 the syenite is brick red in colour and exhibits a porphyritic texture with a grain size of 0.1-0.15 mm for the groundmass and 1.0-1.5 mm for phenocrysts. Xenoliths of komatiitic affinity (talc-chlorite schists), 1-5 mm in size, constitute one percent of the rock. The groundmass exhibits an equigranular, interlocking texture and is moderately to well foliated.

Mineralogically, Sample 51-10 consists of approximately 70 percent feldspar phenocryst whereas sample 56-03 contains two percent chloritized biotite phenocrysts. The groundmass in both samples is composed of 50-80 percent feldspar, 10-30 percent chloritized biotite, 5-10 percent quartz and 5-10 percent specular hematite which has imparted a pervasive red stain to the feldspar.

The syenite contains up to 8 percent disseminated calcite that may represent contamination from komatiite xenoliths as the feldspar in Kirkland Lake syenites is an albitic perthite (Lovell, 1972) that would produce little calcite if altered. The samples contain up to 1 percent magnetite and 1 percent disseminated cubic pyrite.

4.3.6 Diabase (Unit 6)

One diabase dike was intersected in Hole 07 and a second dike was intersected in Holes 25 and 26. Both dikes are in the northwestern corner of the property and are assumed to be of the north-south trending Matachewan variety.

The diabase samples are massive and are dark green-black in colour with the exception of local pink weathering of plagioclase which is characteristic of the lithology (Lovell, 1967). The texture varies from equigranular to diabasic to slightly ophitic, and the grain size is generally 0.2 to 1.2 mm with the coarsest grains being pyroxene crystals that poikilitically enclose smaller plagioclase crystals to produce the ophitic texture.

Because the diabase dikes post-date regional metamorphism, they are unaltered except for local, fracture-controlled epidote veining, chloritization of pyroxene and saussuritization/carbonatization of plagioclase. The major minerals are plagioclase (40 percent), pyroxene (40-45 percent), chlorite (5-15 percent) and quartz (1-2 percent). Most of the pyroxene is pale green in colour but some is pale brown and has a schiller lustre. Accessory minerals include 1 to 5 percent magnetite, 1 percent leucoxene and 2 percent rutile (Hole 26 only).

4.4

Bedrock Geochemistry

All bedrock chip samples from the reverse circulation drilling program were analyzed for copper, arsenic and gold. The analytical results are presented in Appendix E.

Copper results are all less than 200 ppm. Most of the higher values (100 to 200 ppm) occur in mafic volcanics or diabase because the pyroxene that is a major constituent of these rocks is an effective scavenger of copper.

All gold assays greater than or equal to 10 ppb and all arsenic values greater than or equal to 50 ppb are shown on Plan 1. Anomalous gold values occur only in Holes 13 and 58, with 175 ppb (average value of two analyses) and 10 ppb respectively. Neither sample contains elevated arsenic or copper values. The host is conglomerate in Hole 13 and carbonatized komatiite in Hole 58.

Elevated arsenic (50-73 ppm) is generally restricted to the fault zone and the southern edge of the Larder Lake komatiites. Similar concentrations of arsenic have been encountered by ODM elsewhere along the Kirkland Lake - Larder Lake Fault and along the other regional auriferous structures of the Abitibi Belt--the Porcupine-Destor Fault and the Casa-Berard Break.

5. OVERBURDEN GEOLOGY





5.1 Quaternary History and Stratigraphy of the Abitibi Region




The Quaternary geology of the Abitibi region, as determined by ODM from thousands of drill holes and scanty literature, is summarized in Figure 7 and Table 6. Tills from three major glaciations and sediments from two interglacial periods are present.

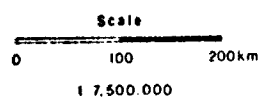
The oldest till was deposited by ice moving southward from Hudson Bay -- possibly 1 million years ago in Kansan time -- and is enriched in clasts of Proterozoic sandstone and Paleozoic limestone. This till is so rarely preserved that it is of no significance in exploration. The next till (Lower Till) was deposited by ice moving southwestward from Nouveau Quebec in Illinoian time more than 125,000 years ago. It is preserved in many buried valleys and contains the dispersion trains from any mineralization in these valleys. The youngest till was



LEGEND

- SOURCE ROCKS**
-  Paleozoic limestone
 -  Proterozoic sandstone
 -  Abitibi Belt volcanics
 -  Archean granite

- ICE MOVEMENT**
-  Wisconsinan
 -  Illinoian
 -  Kansan (?)



Abitibi Quaternary Stratigraphy

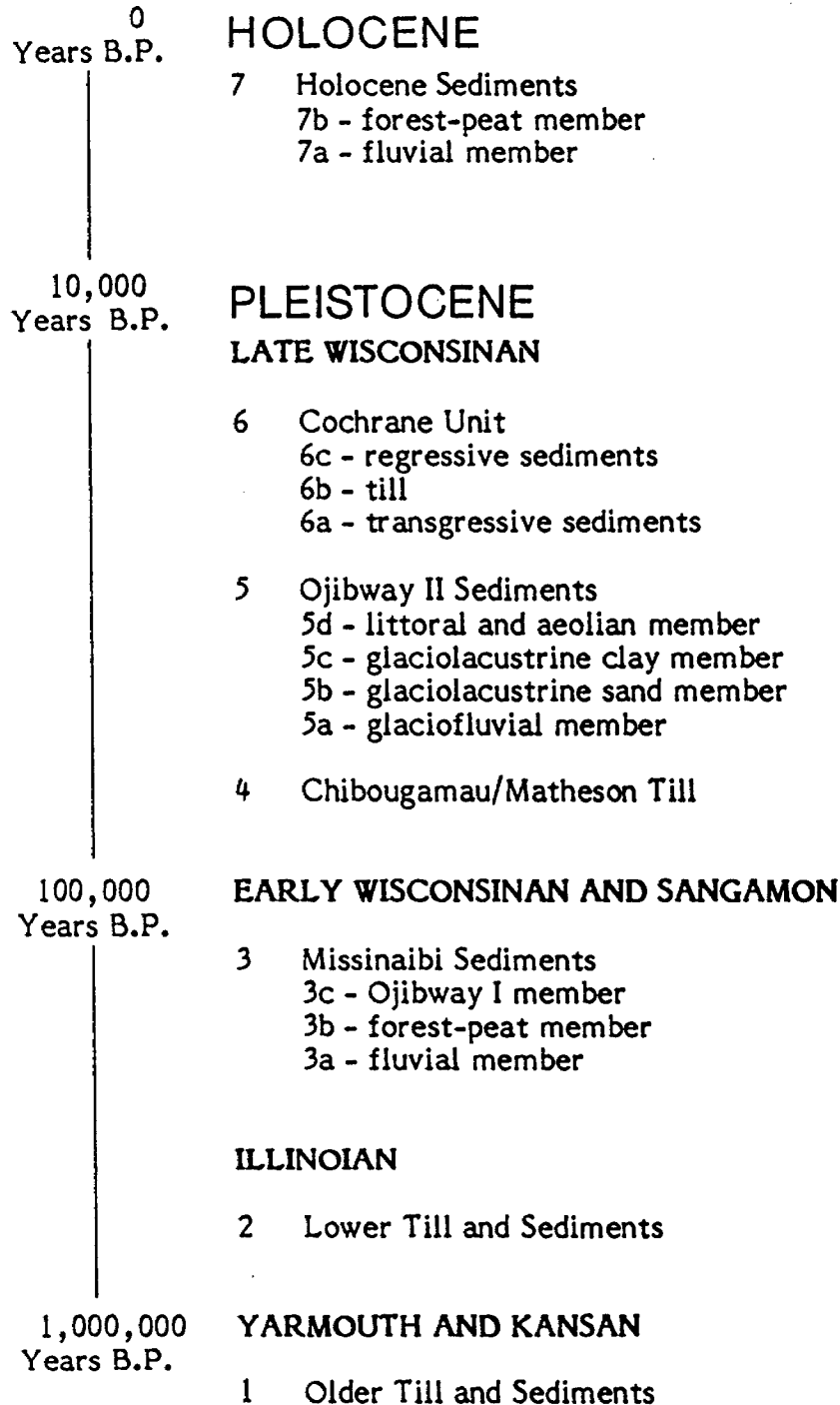


Table 6 - Table of Quaternary Formations for the Abitibi Region

deposited 10,000 years ago by Late Wisconsinan ice that has split into a southeast-moving Matheson/Cochrane lobe west of Val d'Or-Matagami and a southwest-moving Chibougamau lobe east of Val d'Or-Matagami. The esker-like Harricana Moraine was deposited at the contact between the two ice lobes.

In Yarmouth and Sangamon time immediately following the Kansan and Illinoian glaciations, respectively, interglacial sediments including soil profiles and northward-transported fluvial gravels were deposited on the Kansan and Illinoian tills. The gravels consist mostly of recycled till debris, are oxidized, and often contain wood fragments.

In Early Wisconsinan time 100,000 years ago and in Late Wisconsinan time 10,000 years ago, the region was flooded by glacial Lakes Ojibway I and II respectively, and varved clay, silt and fine sand sheets up to 30 metres thick were deposited. The Ojibway I sediments coarsen upward because they were deposited from a transgressive ice sheet. They were overridden by the thick Wisconsinan glacier and are indurated, dry and platy whereas the Ojibway II sediments were deposited from regressive ice, fine upward and are soft. Glaciofluvial esker/delta sands and gravels were deposited by the meltwater rivers that fed both lakes.

The final glacial event in the Abitibi was a minor southeastward re-advance of the thin Cochrane ice lobe into the north part of Lake Ojibway II, depositing Cochrane Till which consists mainly of clay recycled from the soft lake bed. When the Cochrane ice melted, Lake Ojibway II drained catastrophically, exposing the Late Wisconsinan eskers which were subjected to considerable erosion by wave and wind action until they became stabilized by vegetation.

5.2

Quaternary Geology of the Eby Property

During Late Wisconsinan time, the Eby Property was situated in a relatively short-lived, shallow body of water which joined glacial lakes Barlow and Ojibway II. Water depth probably varied between 5-30 metres with bedrock ridges on the property forming small islands in the lake (Vincent and Hardy, 1979).

The formation of Lake Barlow south of the drainage divide was a freak of nature resulting from the construction of a moraine dam in the Lake Timiskaming area. It is highly improbable that similar conditions existed in Early Wisconsinan time; therefore the Eby property probably was not covered by Lake Ojibway I and no clay layer was deposited to protect the pre-Wisconsinan deposits from erosion during the Wisconsinan glaciation. Consequently, only deposits of Late Wisconsinan age were intersected in the drilling. These units are described in detail below and are shown in section in Figures 8 to 14. For the sake of simplicity, the Late Wisconsinan sediments occurring on the property will be referred to as Lake Ojibway II sediments and not as Lake Barlow - Ojibway sediments.

5.2.1 Matheson Till (Abitibi Unit 4)

During the wasting of the Wisconsinan ice sheet, a nearly continuous layer of Matheson Till was deposited on the bedrock surface on the Eby property. The till was absent in only six of the 62 drill holes, five of which are along the axis of a north-south trending esker-delta complex occurring along Highway 66 (Plan 1) and known as the Highway Esker (Baker, 1980). Two of the five holes were abandoned during penetration of delta sands and consequently it is not known if till overlies bedrock in these holes.

The thickness of the till layer is generally less than 4 metres in elevated areas and between 8 and 17 metres in buried bedrock valleys which trend roughly east-northeast parallel to the Kirkland Lake - Larder Lake Fault and transverse to Wisconsinan ice movement, and thus tended to entrap debris as the ice melted. The thickest section was intersected in Hole 09 which was drilled off the southern end of a drumlin (Section L-L', Fig. 14). Drumlins generally occur in areas of flat lying stratigraphy. Flat lying Proterozoic sediments of the Cobalt Group occur just north of the property.

LEGEND

Aotibi Quaternary Stratigraphy

0
Years B.P. **HOLOCENE**
7 Holocene Sediments
7a - forest-peat member
7b - fluvial member

11,000
Years B.P. **PLEISTOCENE**
LATE WISCONSINAN
6 Chubuganai Unit
6a - regressive sediments
6b - till
6c - transgressive sediments
5 Outwash II Sediments
5a - littoral and aeolian member
5b - glaciolacustrine clay member
5c - glaciolacustrine sand member
5d - glaciostuvial member
4 Chubuganai/Matheson Till

109,000
Years B.P. **EARLY WISCONSINAN AND SANGAMON**
3 Missisquoi Sediments
3a - Outwash I member
3b - forest-peat member
3c - fluvial member

ILLINOIAN
2 Lower Till and Sediments

1,000,000
Years B.P. **YARMOUTH AND KANSAN**
1 Older Till and Sediments

Sediment Varieties

- P Peat
- C Clay, silt
- S Sand
- G Gravel
- ST Sand-silt till; clay subordinate
- CT Clay till

Symbols

- Quaternary/bedrock unconformity
- Interglacial unconformity
- Quaternary unit boundary
- Quaternary sub-unit boundary

Geochemistry

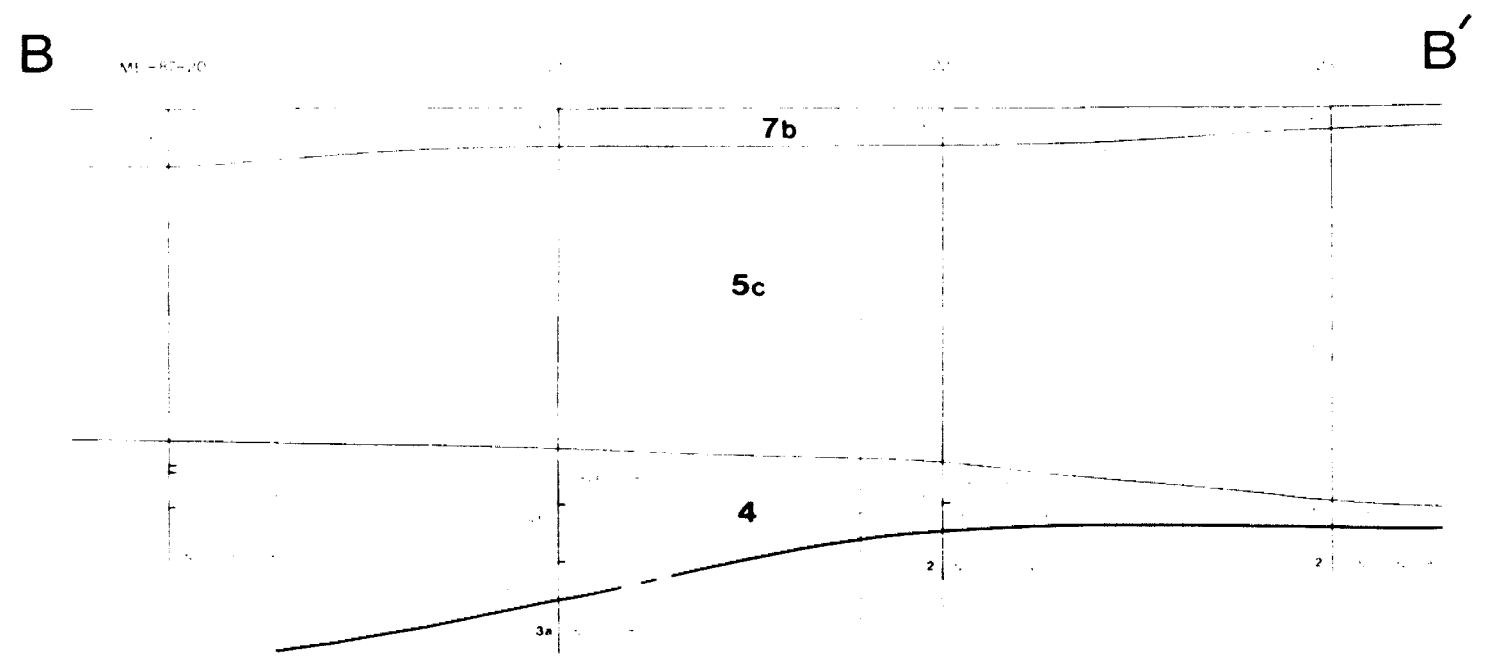
ST 10s 10 17 30
Sand-silt till interval with
19s ppm Au, 10 ppm Cu, 17 ppm
As, and 10 ppm Zn (where
measured) in non-magnetic
heavy mineral fraction (S.G.
greater than 1.0)

Bedrock Lithology

- 6 Devonian
- 5 Silurian
- 4 Ordovician
- 3 Missisquoi Group (sandstone, siltstone, shale, limestone)
- 2 Metasediments (shale, siltstone, sandstone)

Scale

0 100 200 300 400 500



A'

B

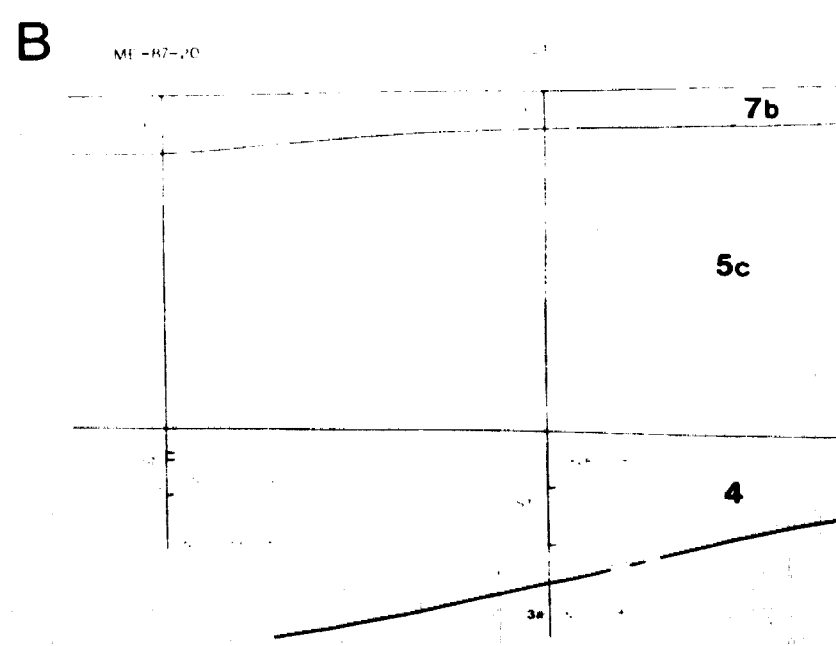
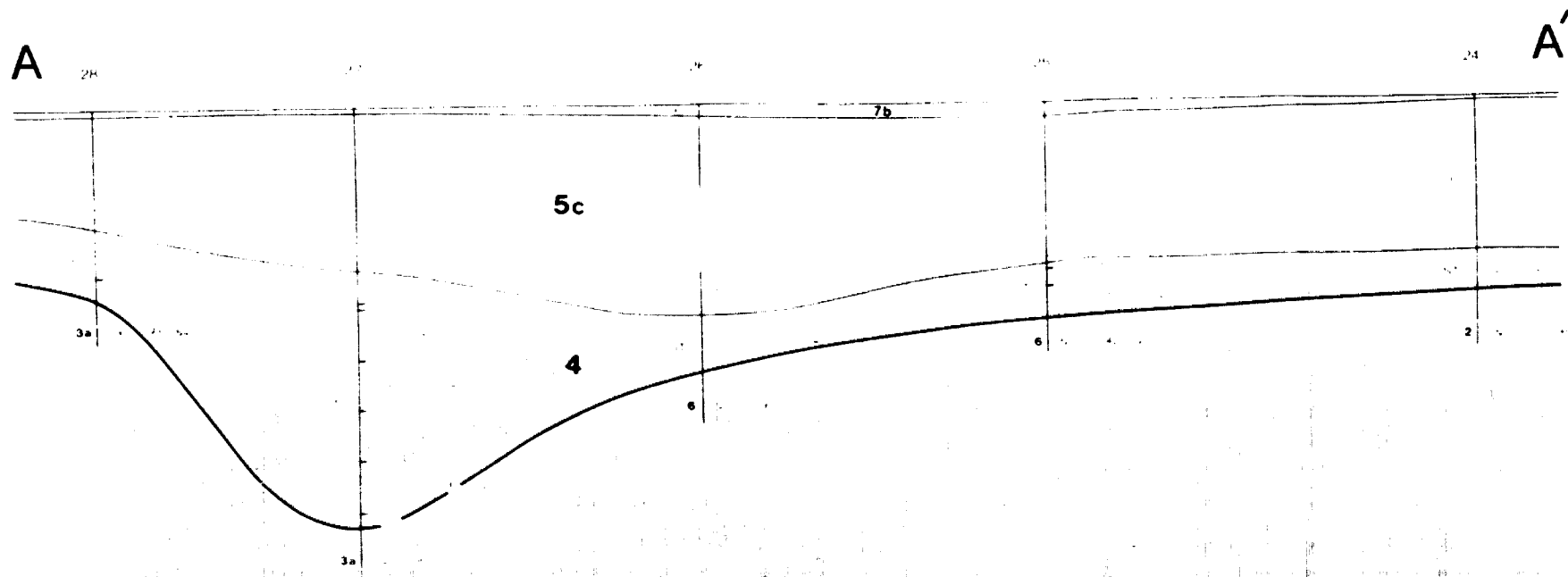
B'

MARY ELLEN RESOURCES LTD.

EBY PROPERTY

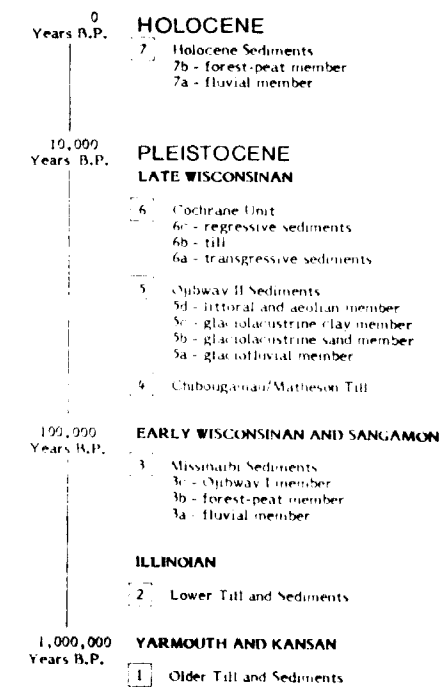
ETC. 740

A-A' and B-B'



LEGEND

Abitibi Quaternary Stratigraphy



Sediment Varieties

P	Peat
C	Clay, silt
S	Sand
G	Gravel
ST	Sand-silt till; clay subordinate
CT	Clay till

Symbols

—	Quaternary/bedrock unconformity
~	Interglacial unconformity
—	Quaternary unit boundary
- - -	Quaternary sub-unit boundary

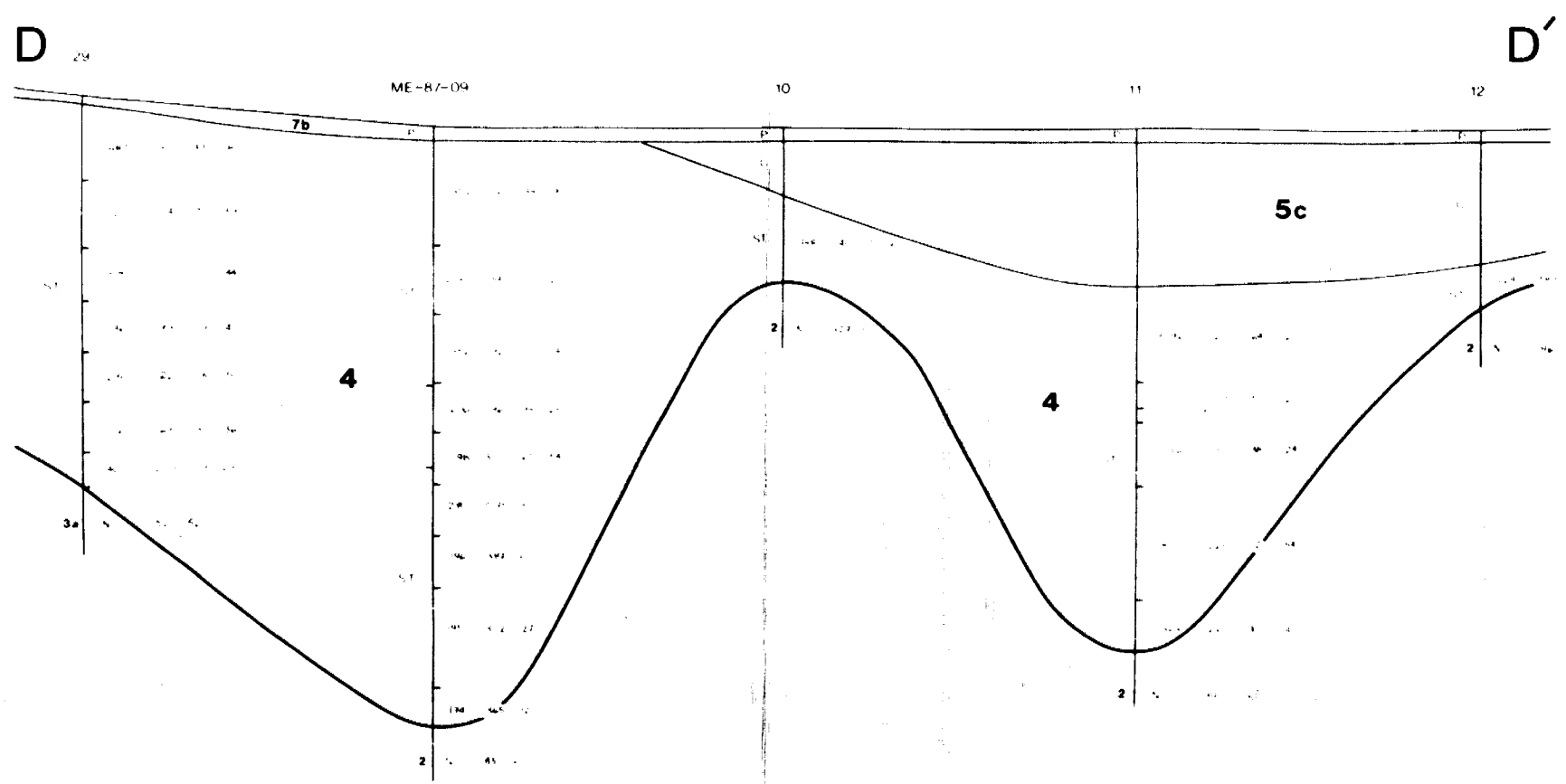
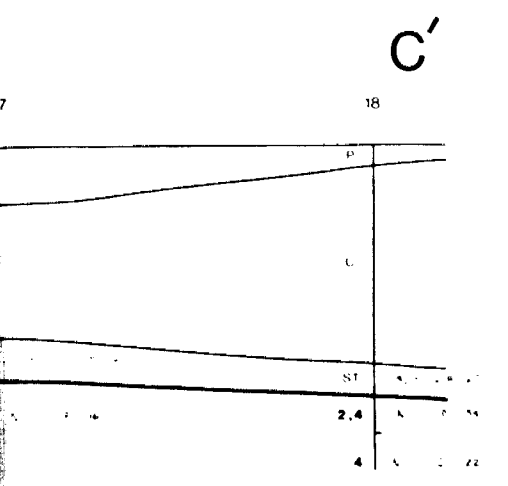
Geochemistry

ST 198 30 17 30 Sand-silt till interval with 198 ppb Au, 50 ppm Cu, 17 ppm As, and 30 ppm Zn (where measured) in non-magnetic heavy mineral fraction (S.G. greater than 3.3)

Bedrock Lithology

6	Diabase
5	Syenite
4	Granodiorite
3	Metasediments (conglomerate (3a); graywacke (3b))
2	Mafic volcanics (basalt)
1	Ultramafic volcanics (komatiite flows)

Scale
 HOR. 1:2,000 VERT. 1:200

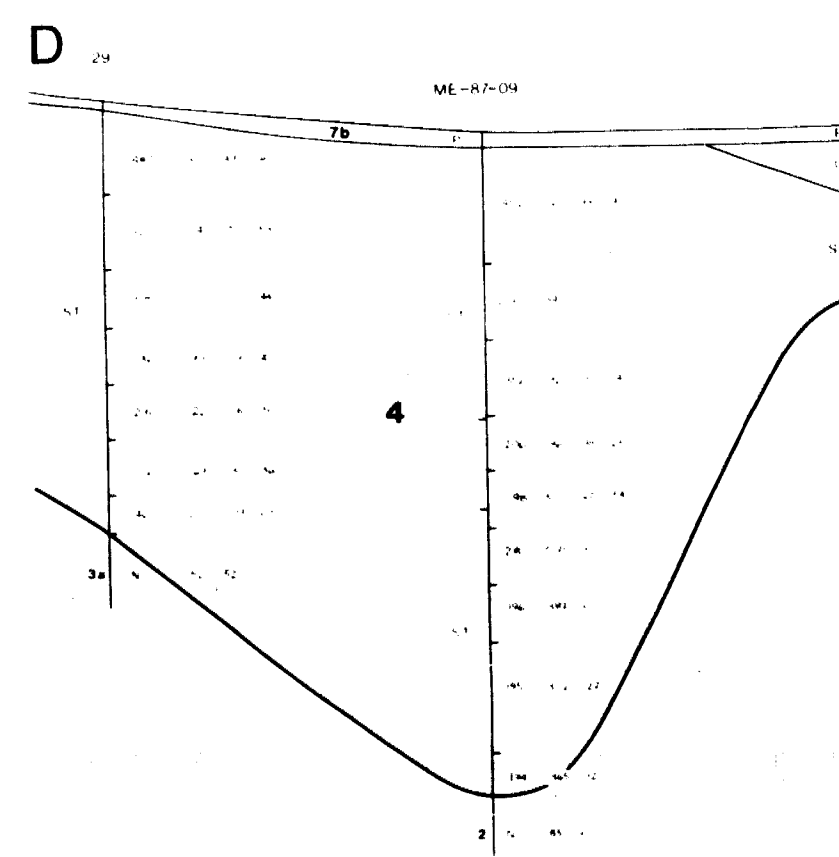
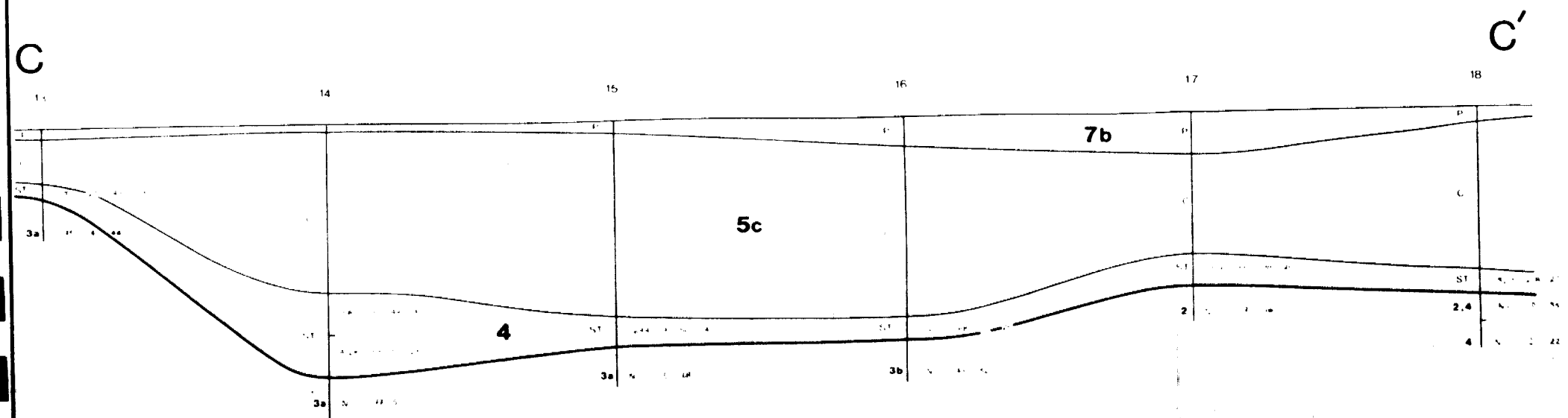


MARY ELLEN RESOURCES LTD.

EBY PROPERTY

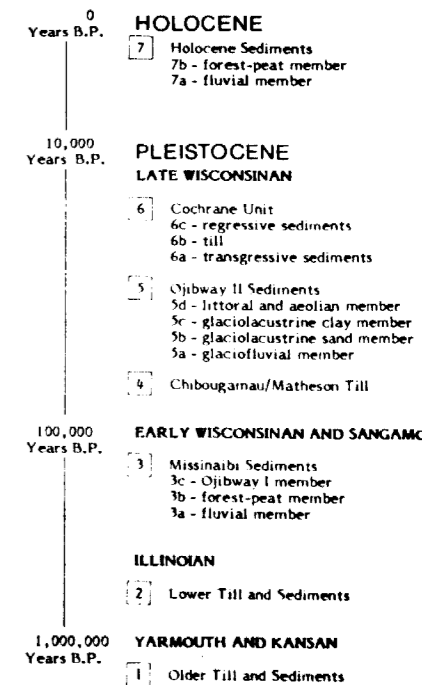
Eby Twp

C-C' and D-D'



LEGEND

Abitibi Quaternary Stratigraphy



Sediment Varieties

- P Peat
- C Clay, silt
- S Sand
- G Gravel
- ST Sand-silt till; clay subordinate
- CT Clay till

Symbols

- Quaternary/bedrock unconformity
- Interglacial unconformity
- Quaternary unit boundary
- Quaternary sub-unit boundary

Geochemistry

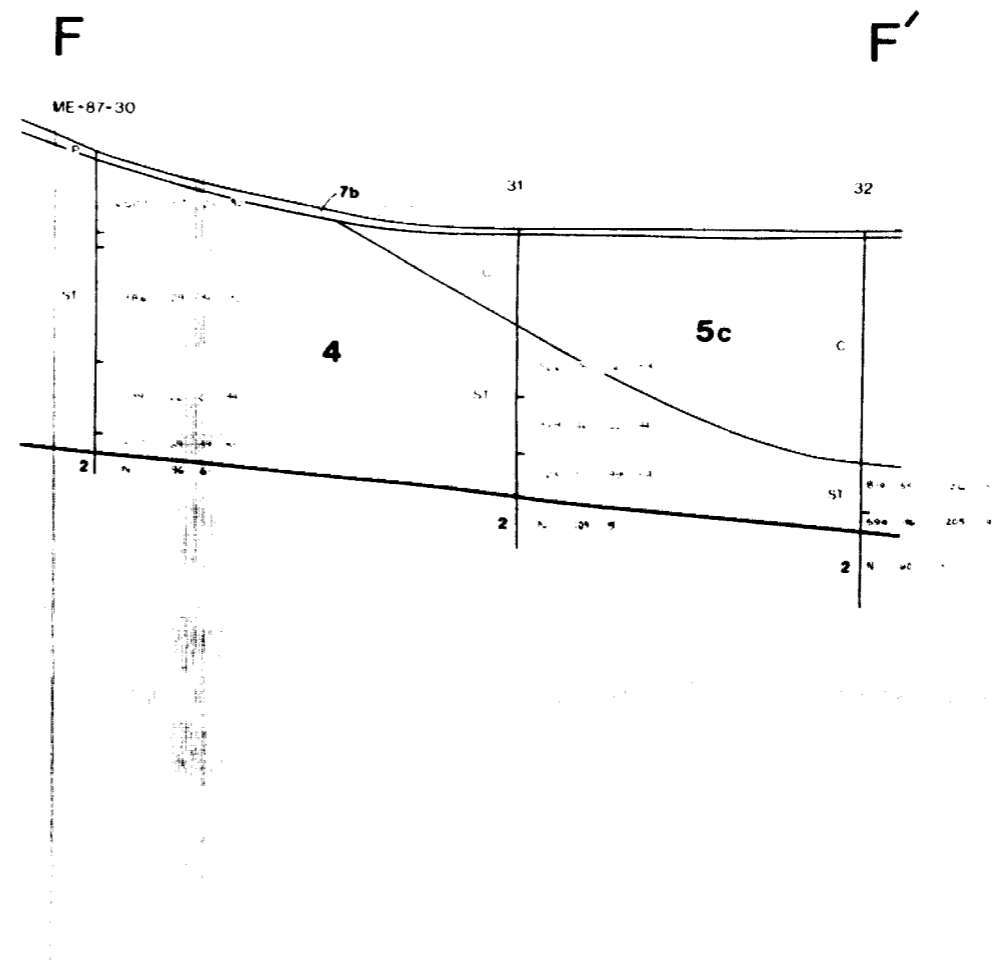
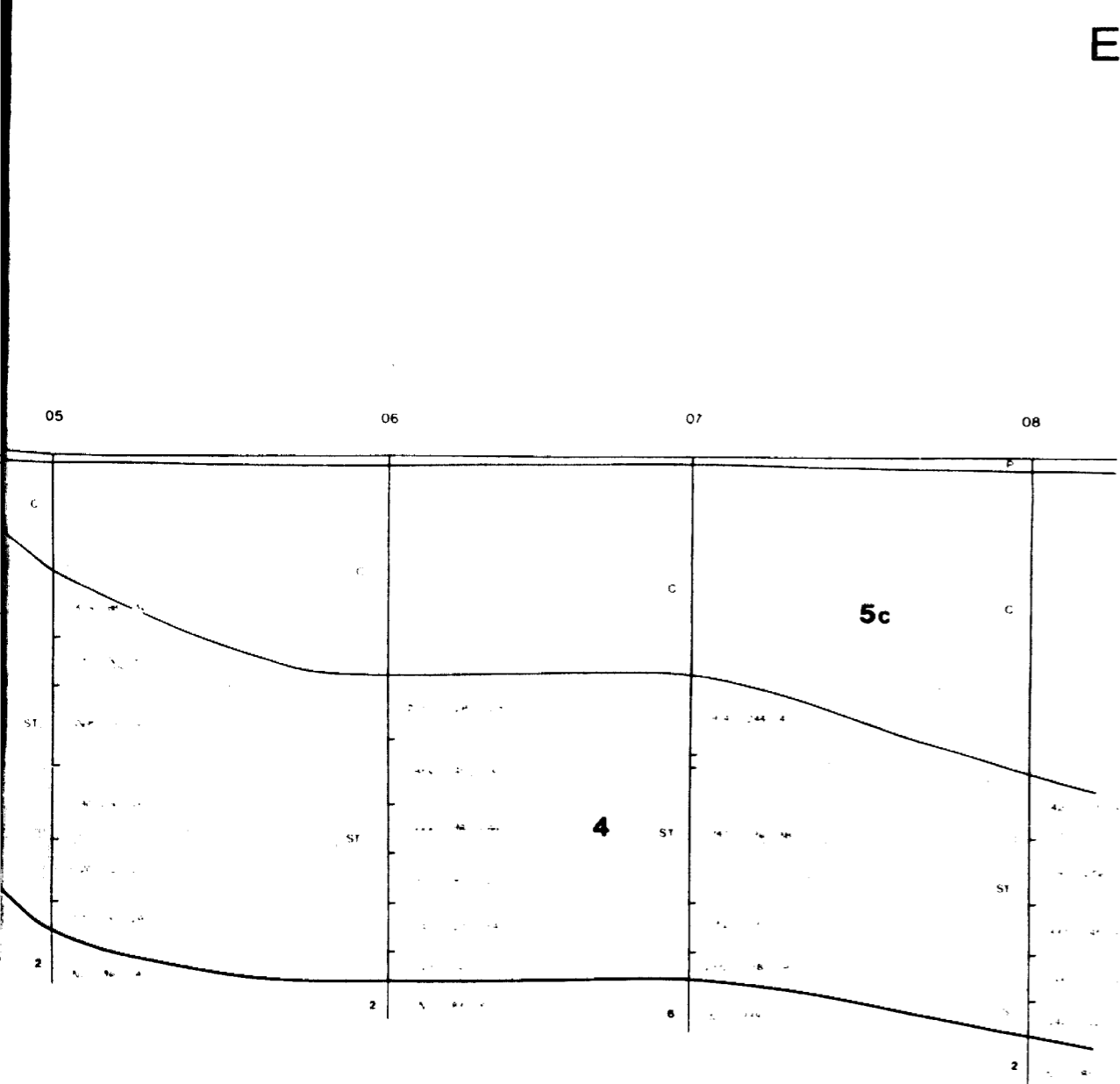
ST : 98 30 17 30
 Sand-silt till interval with
 198 ppb Au, 30 ppm Cu, 17 ppm
 As, and 30 ppm Zn (where
 measured) in non-magnetic
 heavy mineral fraction (S.G.
 greater than 3.3)

Bedrock Lithology

- 6 Diabase
- 5 Syenite
- 4 Granodiorite
- 3 Metasediments (conglomerate (3a); graywacke (3b))
- 2 Mafic volcanics (basalt)
- 1 Ultramafic volcanics (komatiite flows)

Scale

HOR. 1:2,000 VERT. 1:200



MARY ELLEN RESOURCES LTD.

EBY PROPERTY

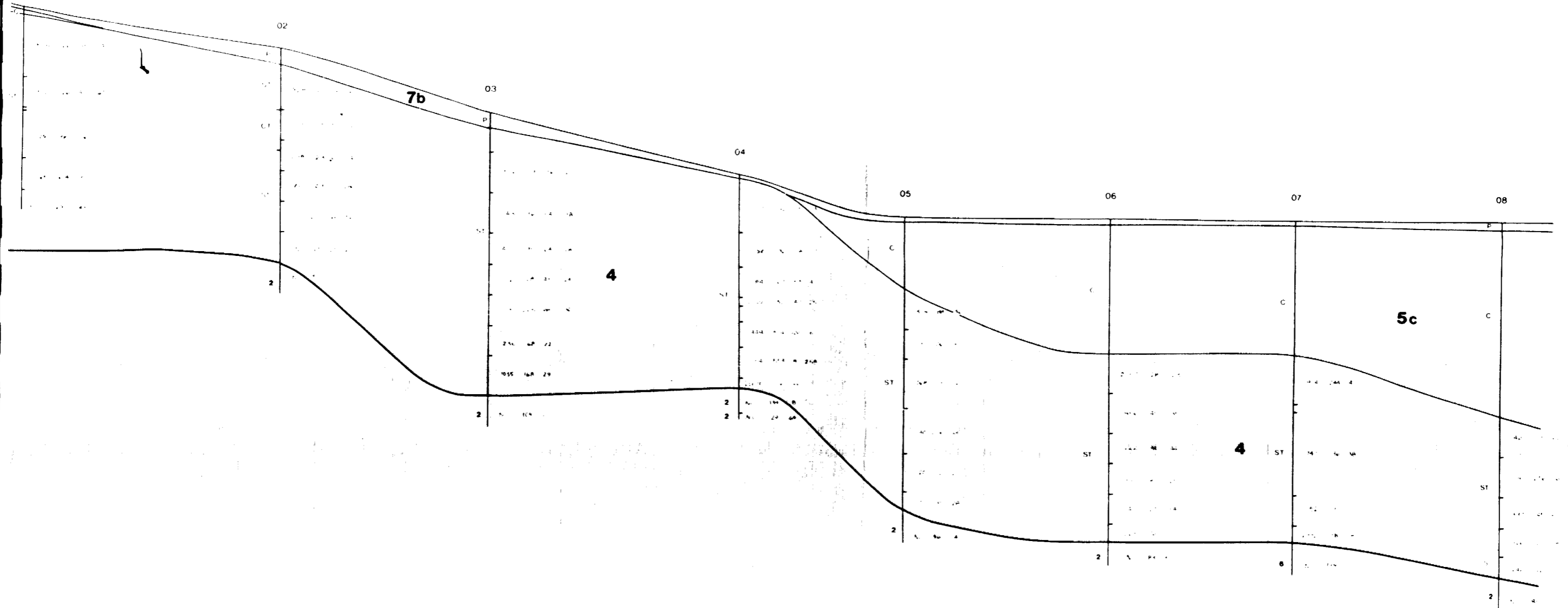
Eby Twp.

E-E' and F-F'

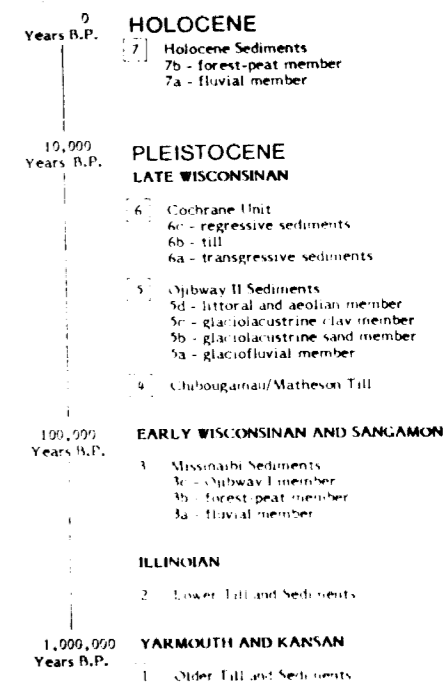
E

ME-87-01

E'



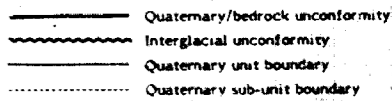
Abitibi Quaternary Stratigraphy



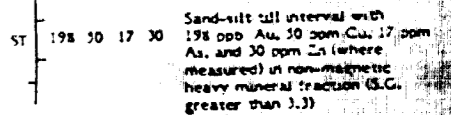
Sediment Varieties

P	Peat
C	Clay, silt
S	Sand
G	Gravel
ST	Sand-silt till; clay subordinate
CT	Clay till

Symbols



Geochemistry



Bedrock Lithology

- 6] Diabase
- 5] Syenite
- 4] Granodiorite
- 3] Metasediments (conglomerate (3a); graywacke (3b))
- 2] Mafic volcanics (basalt)
- 1] Ultramafic volcanics (komatiite flows)

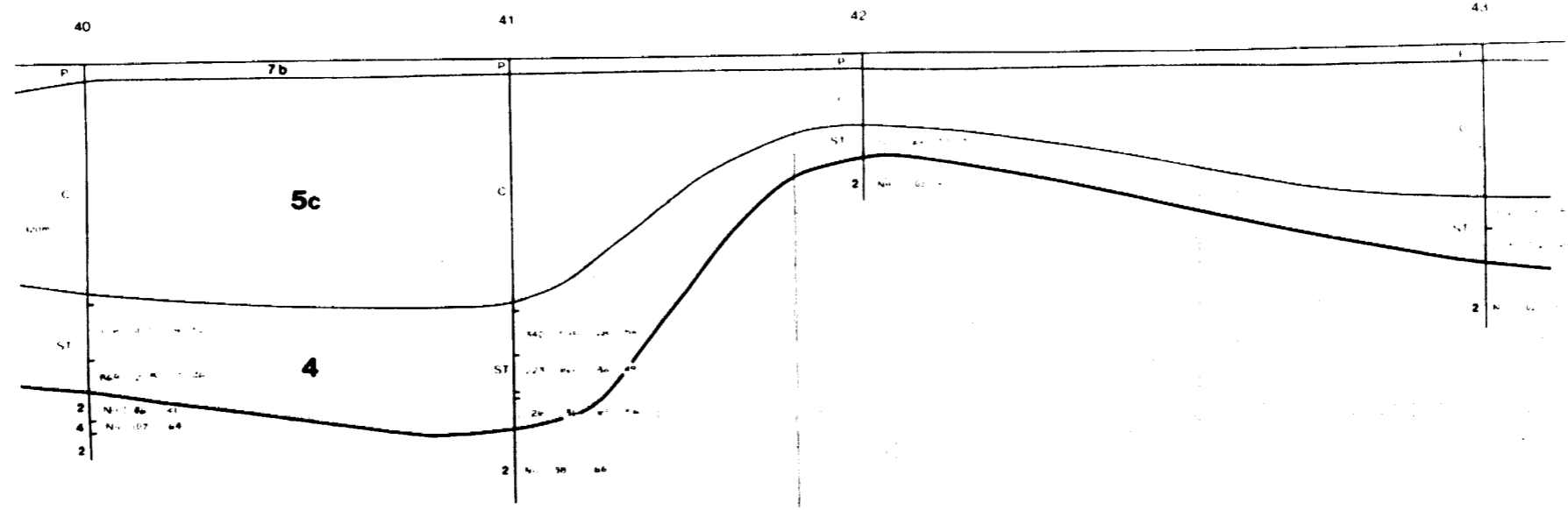
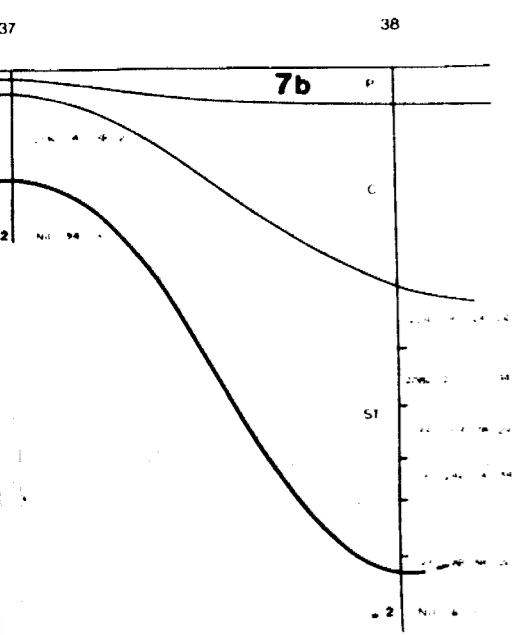
Scale

H. OR. 1:2,000 VERT. 1:200

G'

H

H'



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EBY PROPERTY

Eby Twp

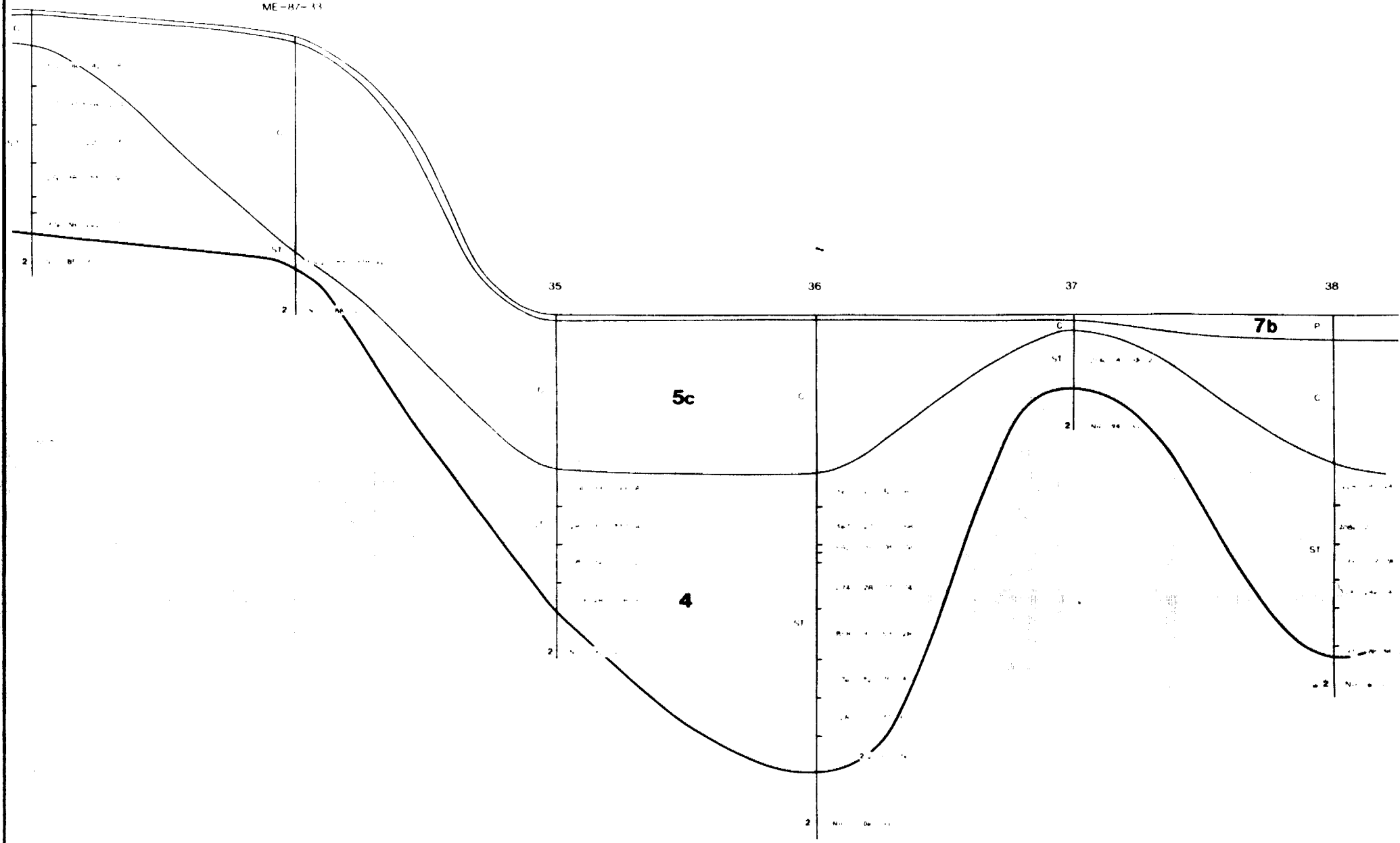
G-G' and H-H'

BY OVERBURDEN DRILLING MANAGEMENT LIMITED JUNE 1987

G

34

ME-87-33

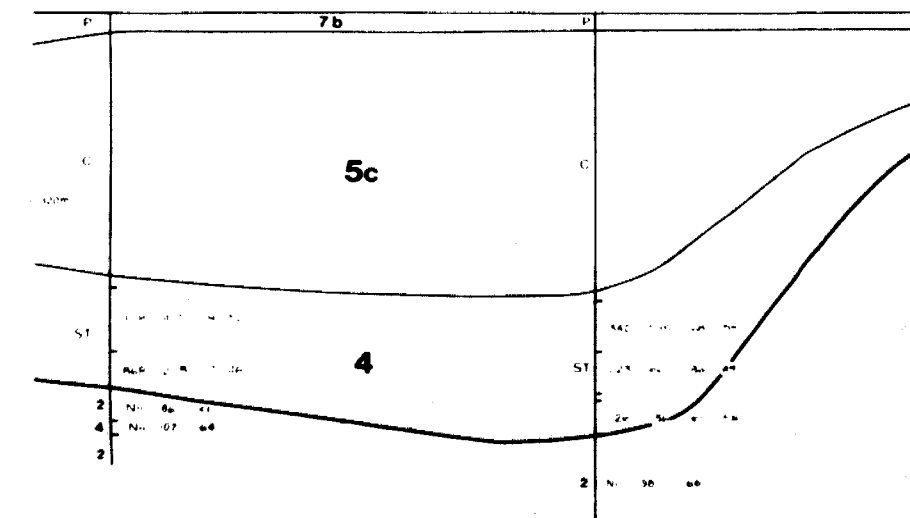


G'

H

40

41



LEGEND

Antibi Quaternary Stratigraphy

- Years B.P.
- HOLOCENE**
- 1 - Holocene Sediments
 - 2a - forest-peat member
 - 2b - fluvial member
- 19,000 Years B.P.
- PLEISTOCENE**
- LATE WISCONSINAN**
- 6 - Cochrane Unit
 - 6a - regressive sediments
 - 6b - till
 - 6c - transgressive sediments
 - 5 - Outwash II Sediments
 - 5d - littoral and aeolian member
 - 5c - glaciolacustrine clay member
 - 5b - glaciolacustrine sand member
 - 5a - glaciolacustrine member
 - 4 - Chibougamau/Matheson Till
- 100,000 Years B.P.
- EARLY WISCONSINAN AND SANGAMON**
- 3 - Mississauga Sediments
 - 3a - Outwash I member
 - 3b - forest-peat member
 - 3c - fluvial member
- ILLINOIAN**
- 2 - Lower Till and Sediments
- 1,000,000 Years B.P.
- YARMOUTH AND KANSAN**
- 1 - Older Till and Sediments

Sediment Varieties

- P - Peat
- C - Clay, silt
- S - Sand
- G - Gravel
- ST - Sand-silt till; clay subordinate
- CT - Clay till

Symbols

- Quaternary/bedrock unconformity
- ~~~~~ Interglacial unconformity
- Quaternary unit boundary
- Quaternary sub-unit boundary

Geochemistry

- ST 198 10 17 30
- Sand-silt till intervals with 15% silt, 10% clay, 10% sand, 10% gravel, 10% peat, 10% organic matter, 10% heavy minerals, 10% S.G. greater than 2.00

Bedrock Lithology

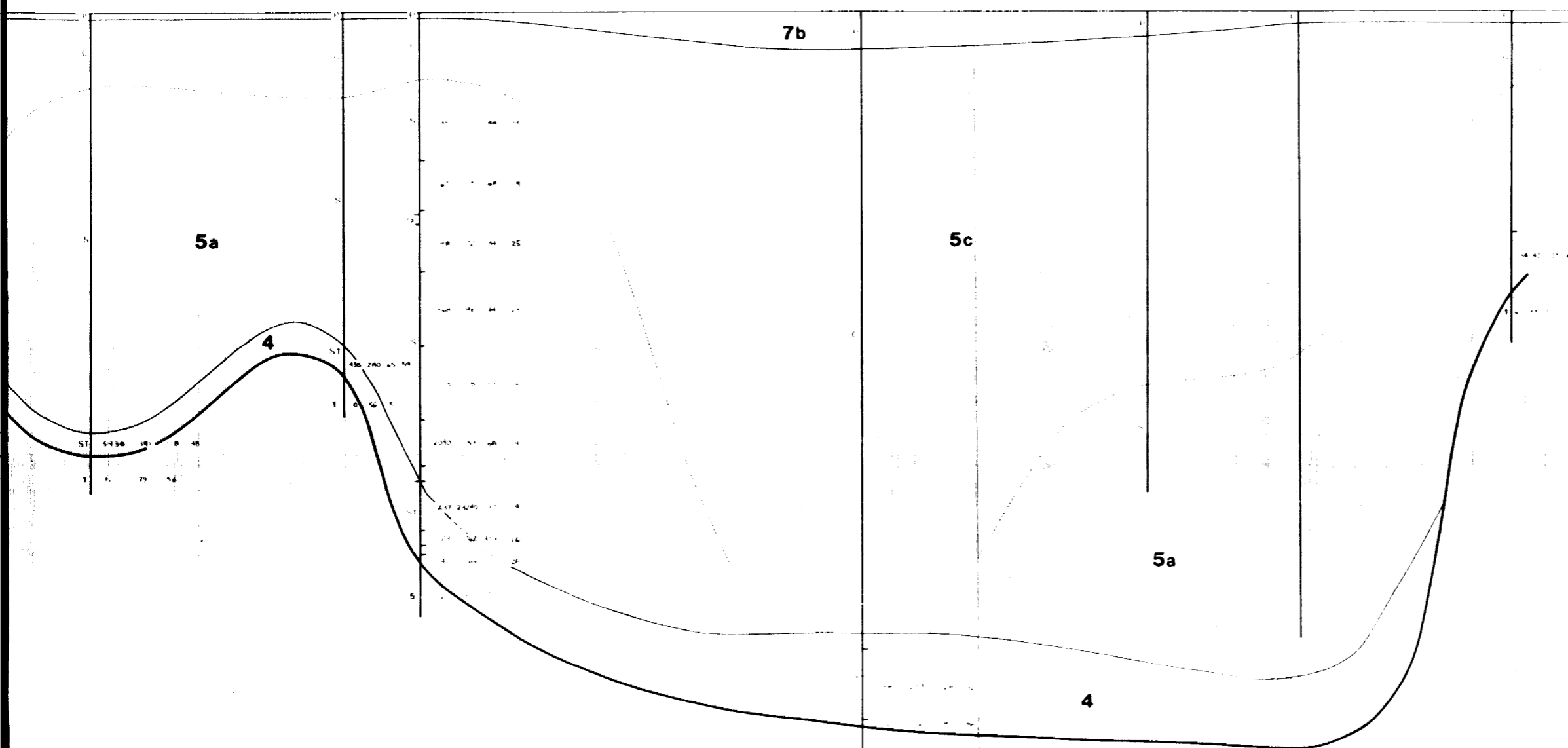
- 6 - Diabase
- 5 - Granite
- 4 - Gneiss
- 3 - Metasediments (Zirconite, Barite, Graphite, etc.)
- 2 - Metavolcanic (Basalt)
- 1 - Intrusive (Granite, Diorite, etc.)

Scale

1:50,000 HORIZONTAL

K'

57 58 ME-87-51 59 60 61 62



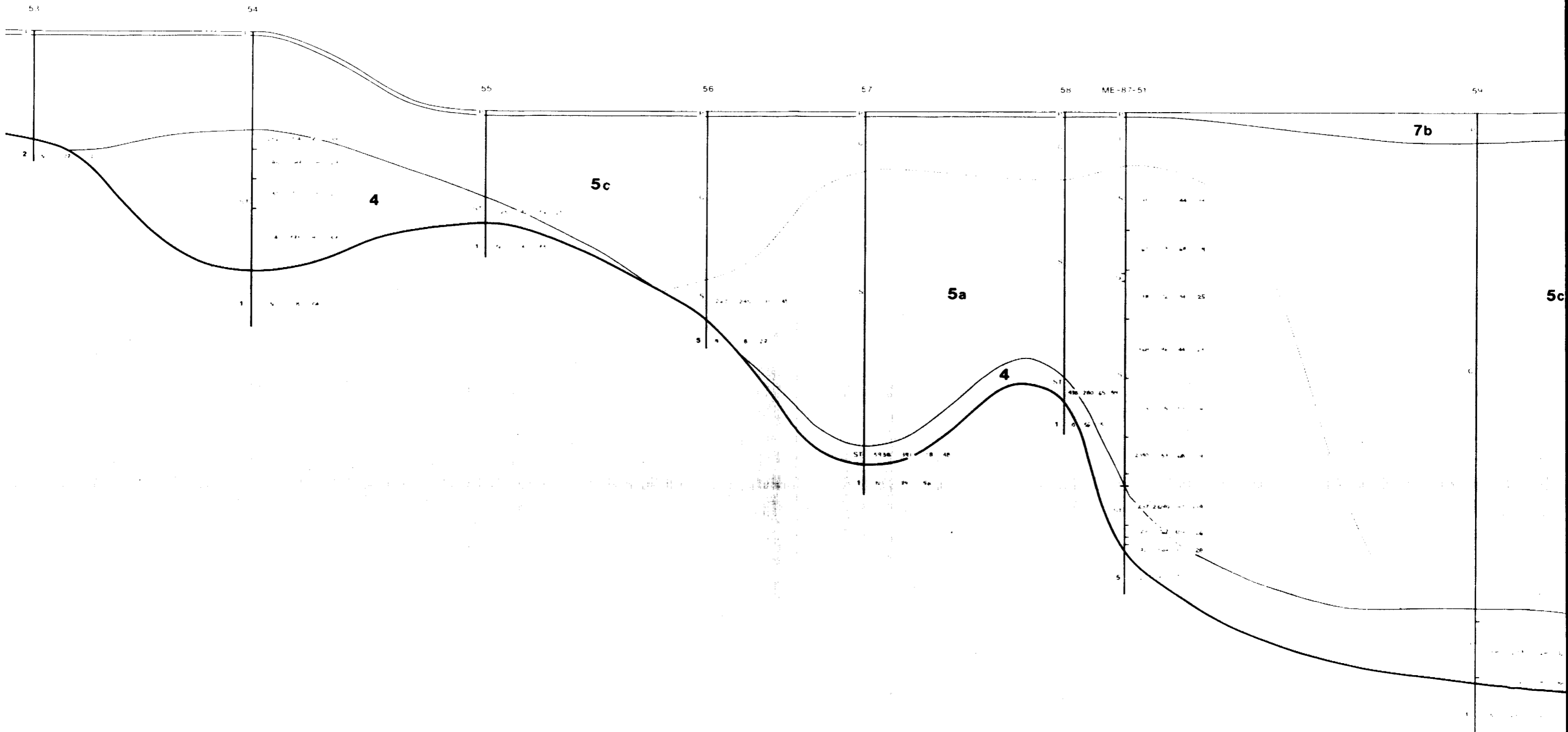
MARY ELLEN RESOURCES LTD

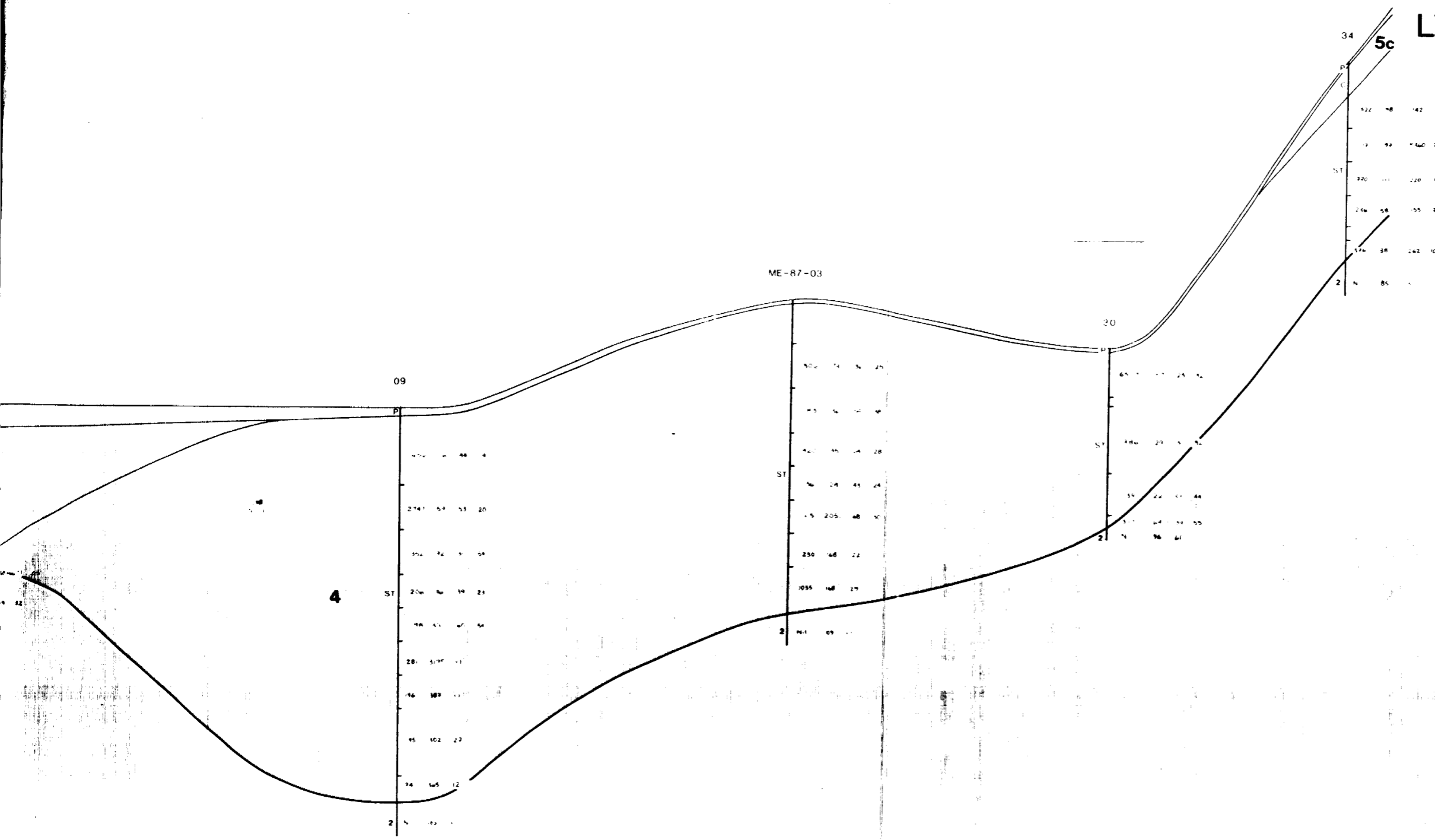
EBY PROPERTY

Ec. Two

K-K'

K





LEGEND
Abitibi Quaternary Stratigraphy

- HoloCENE**
 7 Holocene Sediments
 7b - forest-peat member
 7a - fluvial member
- 10,000 Years B.P.**
PLEISTOCENE
LATE WISCONSINAN
 6 Cochrane Unit
 6c - regressive sediments
 6b - till
 6a - transgressive sediments
 5 Ojibway II Sediments
 5d - littoral and aeolian member
 5c - glaciolacustrine clay member
 5b - glaciolacustrine sand member
 5a - glaciofluvial member
 4 Chibougamau/Matheson Till
- 100,000 Years B.P.**
EARLY WISCONSINAN AND SANGAMON
 3 Missinaibi Sediments
 3c - Ojibway I member
 3b - forest-peat member
 3a - fluvial member
- ILLINOIAN**
 2 Lower Till and Sediments
- 1,000,000 Years B.P.**
YARMOUTH AND KANSAN
 1 Older Till and Sediments

Sediment Varieties
 P Peat
 C Clay, silt
 S Sand
 G Gravel
 ST Sand-silt till; clay subordinate
 CT Clay till

Symbols
 Quaternary/bedrock unconformity
 Interglacial unconformity
 Quaternary unit boundary
 Quaternary sub-unit boundary

Geochemistry
 ST 198 30 17 30
 Sand-silt till interval with
 198 ppm Au, 50 ppm Cu, 17 ppm
 As, and 30 ppm Zn (where
 measured) in non-magnetic
 heavy mineral fraction (S.G.
 greater than 3.0)

Bedrock Lithology
 6 Diabase
 5 Svecite
 4 Granodiorite
 3 Metasediments (subordinate) (3a) granitic (3b)
 2 Mafic volcanic (basalt)
 1 Ultramafic volcanic (mafic gabbro)

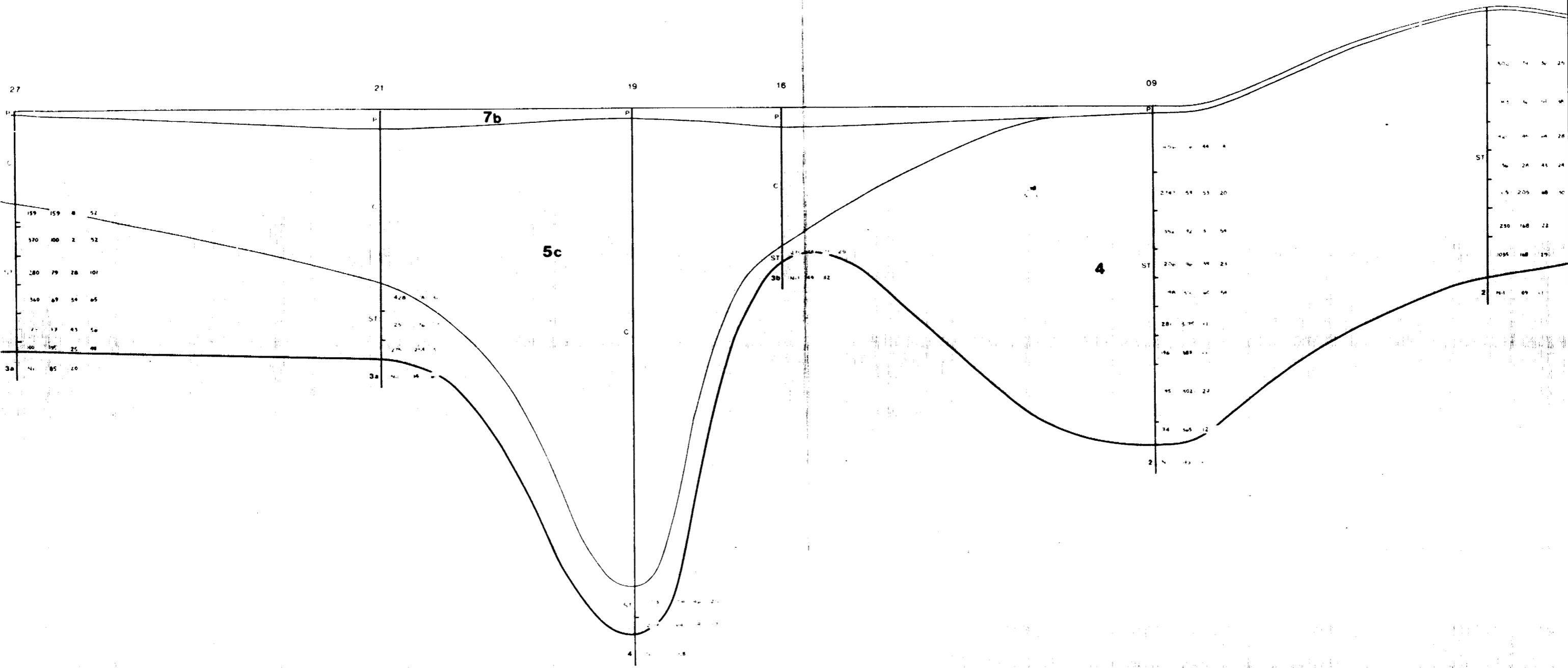
Scale
 H. 1:15,000 V. 1:250

MARY ELLEN RESOURCES LTD.

EBY PROPERTY

Eby Twp

L-L



LEGEND

Abitibi Quaternary Stratigraphy

- HOLOCENE**
- Years B.P. 7
 7 - Holocene Sediments
 7b - forest-peat member
 7a - fluvial member
- 10,000
 Years B.P. **PLEISTOCENE**
LATE WISCONSINAN
- 6 - Cochrane Unit
 6c - regressive sediments
 6b - till
 6a - transgressive sediments
- 5 - Ojibway II Sediments
 5d - littoral and aeolian member
 5c - glaciolacustrine clay member
 5b - glaciolacustrine sand member
 5a - glaciofluvial member
- 4 - Chibougamau/Matheson Till
- 100,000
 Years B.P. **EARLY WISCONSINAN AND SANGAMON**
- 3 - Missinabi Sediments
 3c - Ojibway I member
 3b - forest-peat member
 3a - fluvial member
- ILLINOIAN**
- 2 - Lower Till and Sediments
- 1,000,000
 Years B.P. **YARMOUTH AND KANSAN**
- 1 - Older Till and Sediments

Sediment Varieties

- P Peat
 C Clay, silt
 S Sand
 G Gravel
 ST Sand-silt till; clay subordinate
 CT Clay till

Symbols

- Quaternary/bedrock unconformity
 Interglacial unconformity
 Quaternary unit boundary
 Quaternary sub-unit boundary

Geochemistry

- ST 198 50 17 30
 Sand-silt till interval with
 198 ppm Au, 50 ppm Cu, 17 ppm
 As, and 30 ppm Zn, where
 measured in non-magnetic
 heavy mineral fraction (S.G.
 greater than 1.3)

Bedrock Lithology

6. Diabase
 5. Syenite
 4. Granodiorite
 3. Metasediments (conglomerate (3a); graywacke (3b))
 2. Mafic volcanics (basalt)
 1. Ultramafic volcanics (komatiite flows)

Scale

HORIZ. 1:2,000 VERT. 1:200

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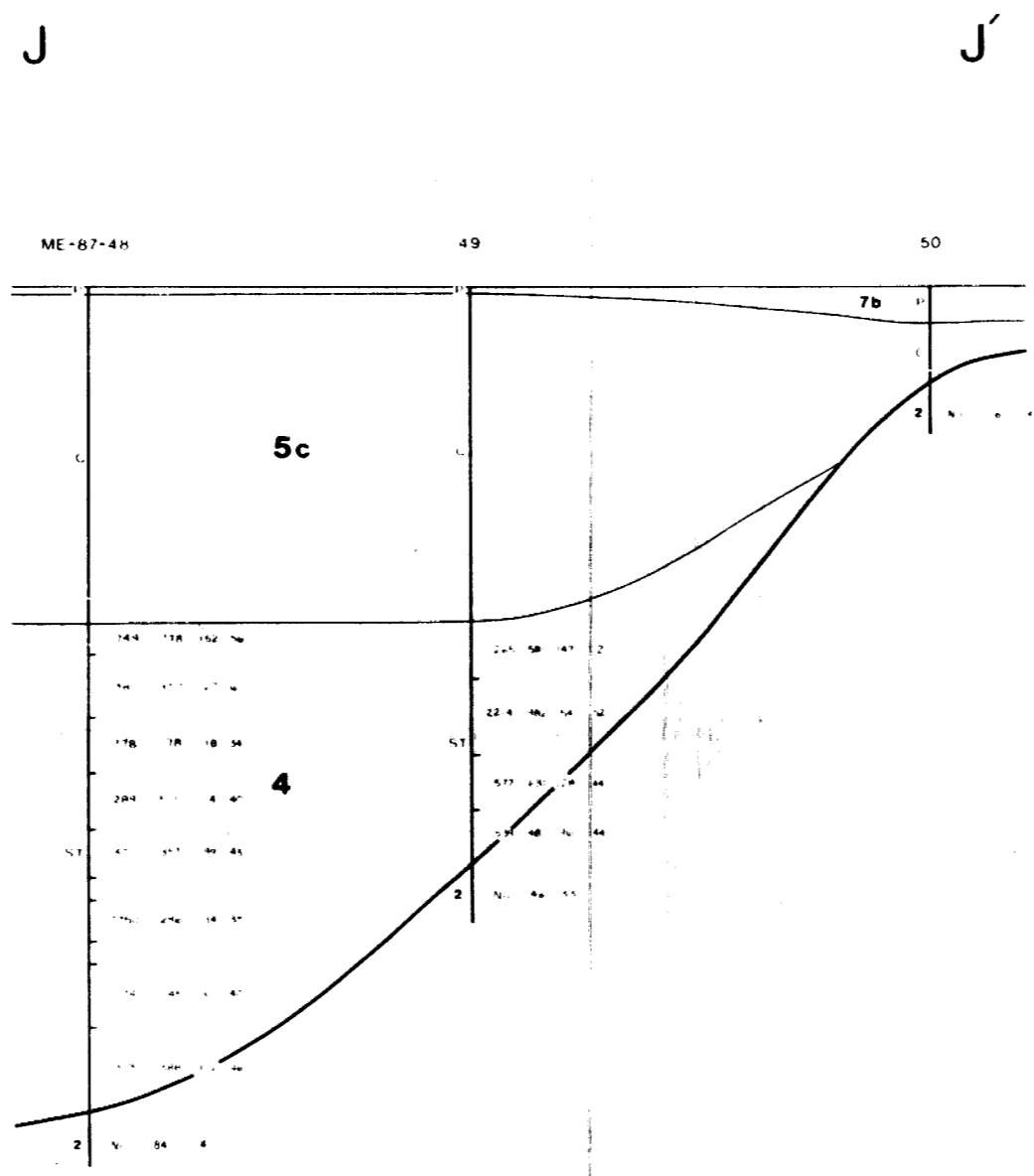
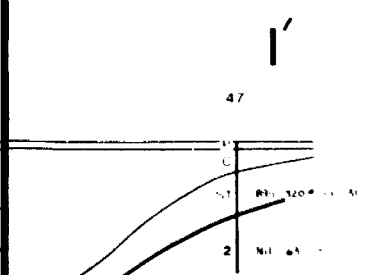
EBY PROPERTY

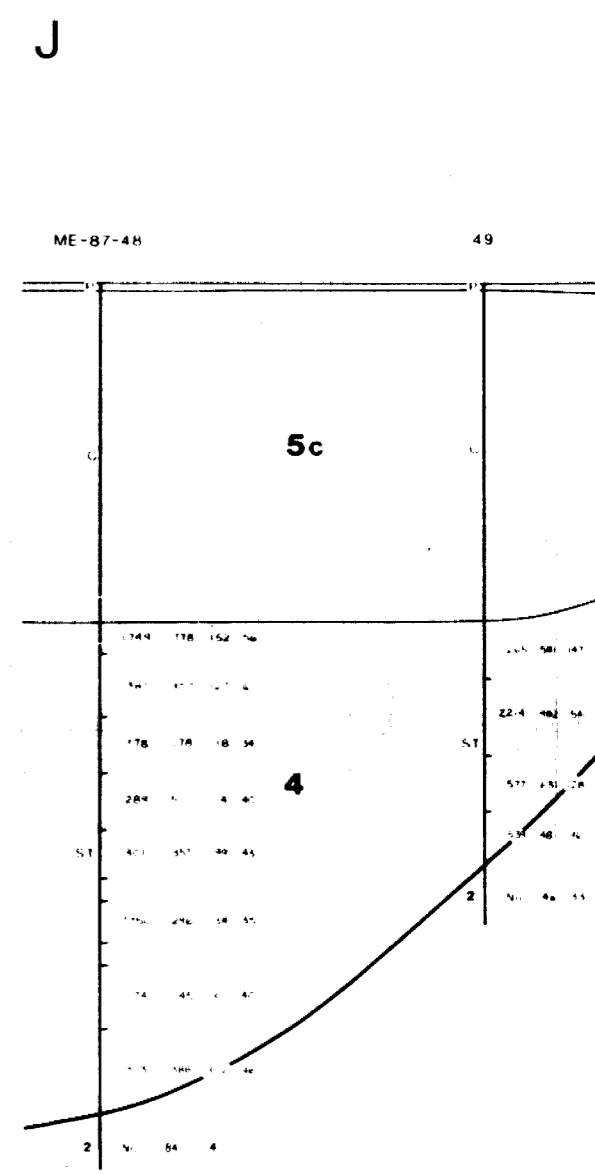
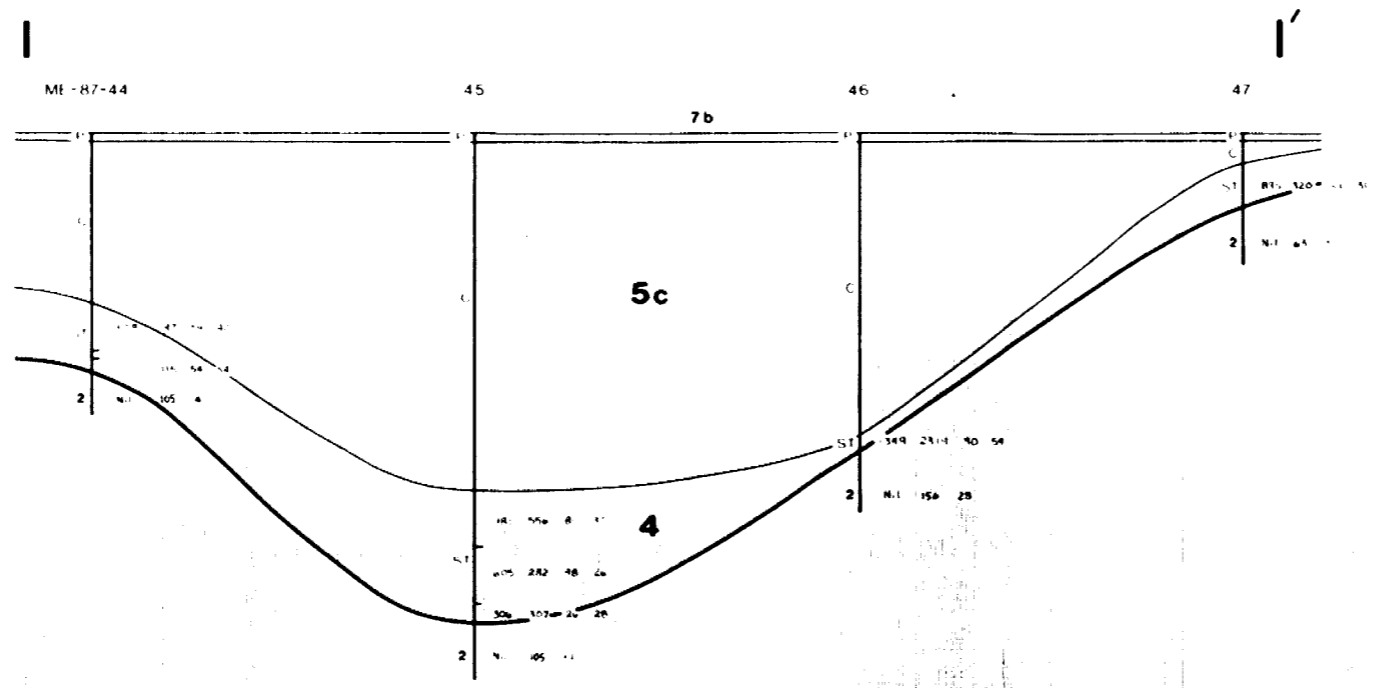
Eby Twp

Geological Map

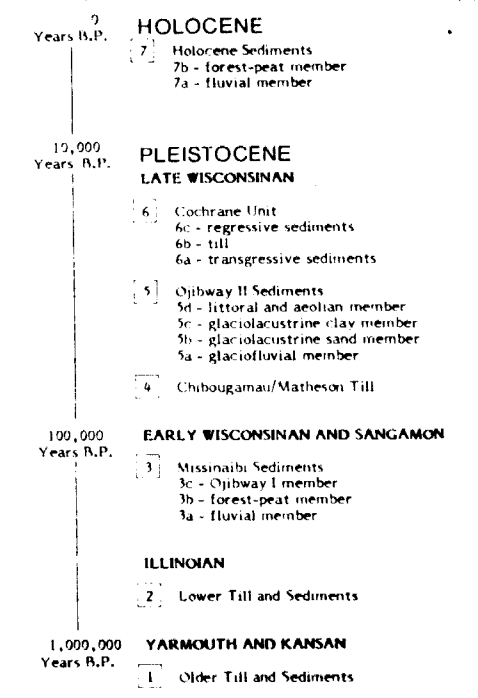
I-I' and J-J'

BY OVERBURDEN DRILLING MANAGEMENT LIMITED JUNE 1987





Abitibi Quaternary Stratigraphy



Sediment Varieties

- P Peat
- C Clay, silt
- S Sand
- G Gravel
- ST Sand-silt till; clay subordinate
- CT Clay till

Symbols

- Quaternary/bedrock unconformity
- ~~~~~ Interglacial unconformity
- Quaternary unit boundary
- Quaternary sub-unit boundary

Geochemistry

ST 198 30 17 30
 Sand-silt till interval with
 198 ppb Au, 30 ppm Cu, 17 ppm
 As, and 30 ppm Zn (where
 measured) in non-magnetic
 heavy mineral fraction (S.G.
 greater than 3.3)

Bedrock Lithology

- 6 Diabase
- 5 Syenite
- 4 Granodiorite
- 3 Metasediments (conglomerate (3a); graywacke (3b))
- 2 Mafic volcanics (basalt)
- 1 Ultramafic volcanics (komatiite flows)

Scale

HOR. 1:2,000 VERT. 1:200

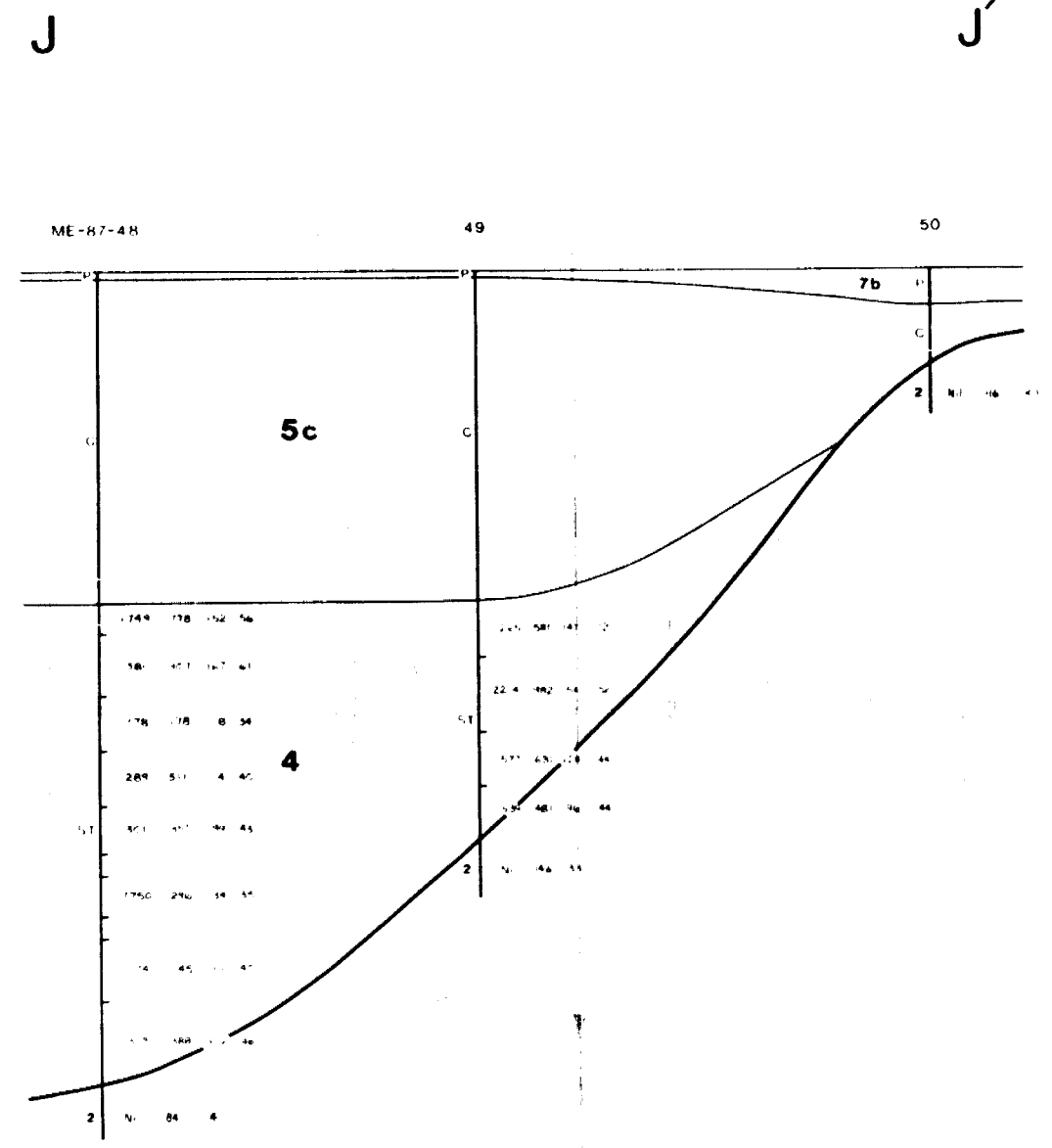
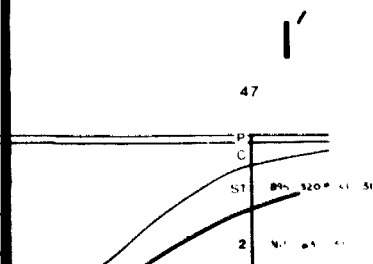
MARY ELLEN RESOURCES LTD.

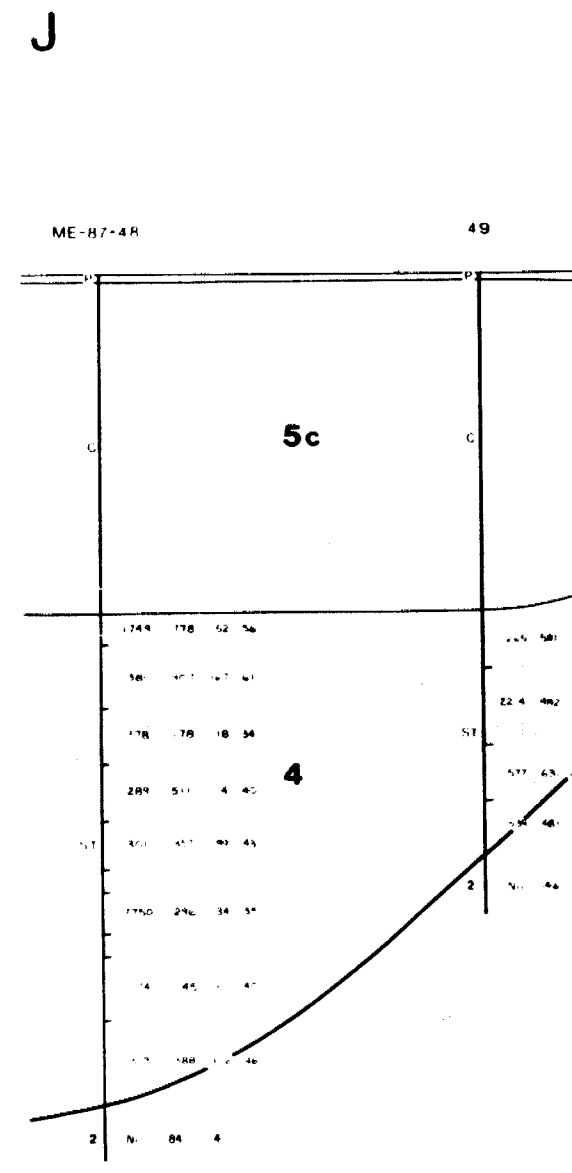
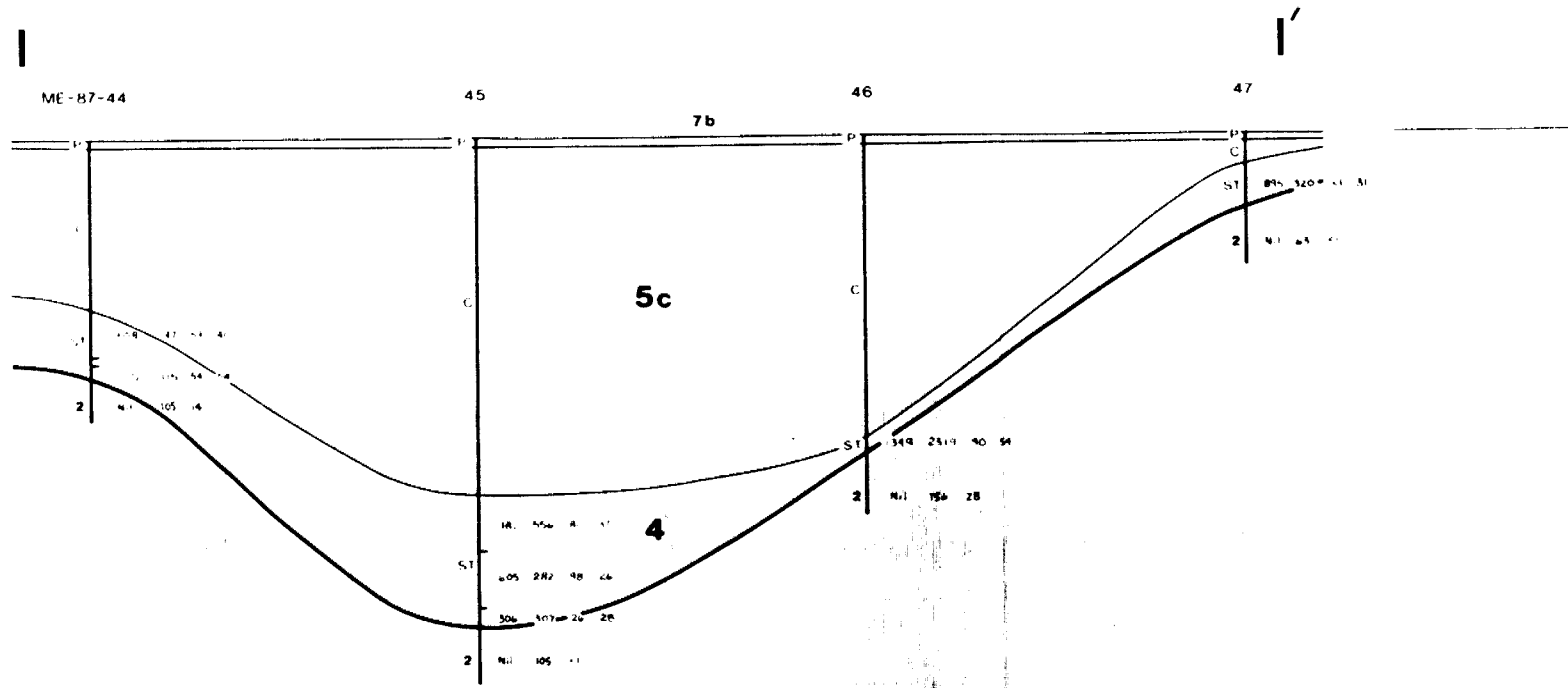
EBY PROPERTY

Eby Twp

Figure 2 - Stratigraphy

I-I' and J-J'





Matheson Till throughout much of the Abitibi region consists mainly of recycled bottom sediments of Lake Ojibway I and thus has either a very sandy or very clayey matrix and is clast-poor. The sand component is generally beige and the clay component is gray. On the Eby property where the Ojibway I sediments were thin or absent, the till is mainly bedrock-derived and is stony with a matrix of silty-sandy rock flour. The matrix of the till was generally described in the field as being gray or gray-beige but actually has a subtle green tone. Stratification is rarely apparent whereas sections that melted out in the deeper parts of Lake Ojibway II often display clay and sand interbeds and are underconsolidated.

Clast lithologies in the till are generally 60-70 percent Abitibi volcanics and metasediments, 20-30 percent granitoids, and 10-20 percent Cobalt sediments with the Abitibi component rising to greater than 90 percent in the lower 1 to 3 metres of some sections. Most of the granitoids as well as the Abitibi volcanics and Cobalt sediments are probably short-travelled as the Watabeag Batholith, a 30-40 km wide granitic complex, occurs only 15 km up-ice from the property. Thus the till should be a good medium for heavy mineral geochemical exploration.

5.2.2 Ojibway II Sediments (Abitibi Unit 5)

The following sediments were deposited on the Eby property while it was flooded by glacial Lakes Barlow and Ojibway II.

Subunit 5a: Glaciofluvial sand and gravel of the Highway Esker which was deposited partly as the bed of a river channel in the ice and partly as a subaqueous fan or delta at the mouth of this channel.

Subunit 5c: An upper ice-distal silt, clay member of the lake bed.

The Highway Esker glaciofluvial member (Subunit 5a) is restricted to the east end of the property and was intersected in seven drill holes. The esker is best represented in a 16 metre section in Hole 51 (Fig. 13). It consists mostly of sand rather than gravel indicating a rapid reduction in the flow energy of the meltwater channel during the transition from the erosional to the depositional stage. Delta development along the flanks of the esker during the retreat of the ice was minimal and the lower, ice-proximal silty sand layer (Subunit 5b) of the lake bed that is common elsewhere in the region is absent on the Eby property.

The esker sand is well bedded in grain sizes ranging from fine to coarse and often contains pebble laminations. It is a clean, washed beige colour and the pebbles are well rounded and polished. The delta sands are gray-beige to beige fine sand.

Ice-distal glaciolacustrine clay and silt (Subunit 5c) blanket the property except around bedrock outcrops where the Matheson Till rises to surface. The clay reaches a maximum thickness of 23 metres in a buried bedrock valley intersected by Hole 19 (Fig. 14) and has a significant levelling effect on the terrain surface. This is well illustrated on Section K-K' (Fig. 13).

The clay is soft and gray and locally grades downward into silt. Many sections were described as pure gray clay on the field logs but probably contain subtle beige silt varves.

5.2.3 Holocene Sediments (Abitibi Unit 7)

The glaciolacustrine clay on the Eby property is locally mantled by 0.5 to 2 metres of peat that was deposited during the 8,000 years that have elapsed since the draining of Lake Ojibway II.

6.

OVERBURDEN GEOCHEMISTRY

6.1 Regional Gold and Base Metal Background and Anomaly Threshold Levels

Heavy mineral gold anomaly threshold levels and properties of significant gold dispersion trains are detailed in Appendix G. In summary, visible gold particles of various sizes are randomly scattered through the till and the absence or presence of one or two of these particles in a standard 8 kilogram sample may result in an analytical background ranging from less than 10 to greater than 50,000 ppb (Table 5). Because of this great variability, we have established an anomaly threshold level of 10 grains of visible gold. Recognizing that some anomalies may be caused by gold occluded in sulphides or other minerals rather than by free gold grains, we also investigate any anomalies over a second, 1,000 ppb threshold. The 1,000 ppb value is based on the observation that heavy mineral concentrates from most gold dispersion trains have a gold content similar to that of the source mineralization; thus 1,000 ppb in the till is suggestive of anomalous bedrock and values over 3,000 ppb are suggestive of ore-grade mineralization. Significant anomalies, in addition to being caused by more than 10 gold grains or by occluded gold, also generally display vertical stratigraphic continuity within the host till horizon and may have an associated pathfinder metal, particularly arsenic or copper. Delicate or irregular gold grains are also significant as they normally indicate a proximal source. Any gold grains present should also be of similar size, indicating a common source.

The base metal background of a heavy mineral concentrate, and particularly of our high-density methylene iodide concentrates, is higher than that of a raw till sample, ranging up to several hundred ppm, because base metals tend to substitute to a significant extent for other metal ions in the structures of heavy silicate and sulphide minerals such as pyroxene and pyrite. The established anomaly threshold level for Cu and Zn, indicating the presence of ore-type minerals such as chalcopyrite and sphalerite in the sample, is 800 ppm. Because till concentrates from dispersion train samples tend to grade the same as the bedrock source mineralization, massive sulphide deposits which typically grade 50,000 ppm (5

percent) combined Cu-Zn often produce anomalies over 10,000 ppm in each metal. The anomaly threshold level for arsenic is about the same as for Cu and Zn but only those arsenic anomalies having a gold association are significant.

Significant base metal anomalies, like significant gold anomalies, normally display vertical continuity in the host till and have a pathfinder association. In the case of copper and zinc, the presence of grains of banded massive pyrite-chalcopyrite-sphalerite mineralization in the concentrate is a favourable indicator whereas the presence of only coarse crystalline vein-type chalcopyrite or sphalerite is unfavourable.

6.2 Eby Heavy Mineral Gold Background

The gold background of the till on the Eby property is higher than the Abitibi average. This undoubtedly reflects the proximity of the area to the regionally auriferous Kirkland Lake - Larder Lake Fault as we have found a similar gold background elsewhere along this fault and also along the Porcupine-Destor Fault and the Casa-Berdard Break.

The average number of gold grains reported for Eby samples containing background levels of gold (less than 10 grains) was 3.2 grains. Sand and gravel samples consistently contain less gold than till samples.

The calculated gold grain assays for background samples are generally between 10 and 800 ppb and show good correspondence with the Bondar-Clegg assays. This shows that:

1. We were often able to observe all of the gold grains that were recovered on the table.
2. In most samples, all of the recovered gold is free gold rather than occluded gold.

6.3

Eby Heavy Mineral Gold Anomalies

Assayers (Ontario) Ltd. analyzed the Eby overburden concentrates for gold using the fire assay method with atomic absorption finish. The weight of sample employed was the weight of the 3/4 concentrate minus a 1-gram subsample that was analyzed for Cu, Zn and As.

Gold assays exceeding the 1000 ppb anomaly threshold and ranging up to 25,000 ppb were reported for twenty-eight of the one hundred and eighty-one overburden samples (16 percent). Four additional samples (3 percent) containing visible gold would have given assays over 1000 ppb if the coarsest grain(s) had entered the 3/4 analytical split of the heavy mineral concentrate. Another eight samples (4 percent) yielded gold grain counts equal to or greater than the 10 gold grain anomaly threshold but assayed less than 1,000 ppb because all of the gold grains were fine. Thus, a total of forty samples representing 23 percent of the collected samples are anomalous.

In the Abitibi region, on average, 10 percent of samples that contain only background levels of gold yield anomalous results due to:

1. The chance occurrence of one or two coarse gold grains in the sample ("nugget effect"), or;
2. The chance occurrence of 10 or more fine gold grains in the sample ("cluster effect").

Thus 10 of every 23 anomalous samples on the Eby property can reasonably be assumed to be false "background anomalies", with the remaining anomalies reflecting either the higher-than-average gold background along the Kirkland Lake - Larder Lake Fault or the presence of dispersion trains related to significant bedrock mineralization, or both. That most of the anomalies are of the false background type is evident from the relatively uniform distribution of the anomalies throughout the drill area, as shown on Plan 2.

The anomalies on Plan 2 are plotted INPUT-fashion. Where two or more anomalous samples are present in a hole, the best anomaly is shown. Quadrants one through four (clockwise from upper right) represent greater than or equal to one thousand ppb Au, greater than or equal to ten grains of visible gold, stratigraphic continuity and a pathfinder metal association, respectively.

As numerous anomalous samples and holes are present, various screening processes are used to separate background noise from those anomalies which are, or may be, caused by dispersion from significant mineralized sources. The screening processes and anomalies discounted are listed in Table 7. In many cases anomalies are discounted for more than one reason.

The first screening method is to discount anomalies which have no stratigraphic continuity although we rarely eliminate anomalies solely on this basis. In this regard, an anomaly at the base of a till horizon is automatically assumed to have stratigraphic continuity as is an anomaly in a single sample till horizon. A lack of stratigraphic continuity is displayed by a single, isolated anomalous sample within or at the top of a multi-sample till horizon. A gold anomaly with no stratigraphic continuity is generally caused by a single nugget or by an erratic cluster of background gold grains. Such anomalies are so common that often they are vertically contiguous with one another or with a gold anomaly of another type. We refer to this as "chance continuity" and discount the anomaly as if it had no continuity.

A second phase of anomaly screening is the calculation of assays (Appendix C) using the formula/parameters discussed in Appendix G. In this case the calculated and measured (geochemical) assays are compared. Either good correlation or a low measured assay is indicative of sufficient visible gold being seen initially to account for the anomaly. We consider the correlation between calculated and measured assays to be "good" if the calculated assays are not more than twice as high as or fifty percent less than the measured assays. This allows for a doubling or halving of the normal thickness factor for flake gold particles used in the calculation. A low measured assay indicates that the largest grain of visible gold or a disproportionate number of the grains remained in the retained 1/4

Hole No.	Gold Anomalies		Grains V.G. (*Not Panned)	1st Phase Screening (Strat. Cont.)	2nd Phase Screening (Good Corr./ Low Assay)	3rd Phase Screening (Nugget Effect)	Remarks
	Sample No.	Au Assay (ppb) Meas. Calc.					
ME-87- 01	02	312 231	10	No	Yes	No	All gold grains abraded, 0.5% py., 200 grains aspy.
	02	06 72 163	14	Basal	Yes	No	All gold grains (except two) abraded, 2% py.
	03	06 230 100	10	Chance	Yes	No	All gold grains (except two) abraded, 1% py.
		07 1,035 503	6	Basal	Yes	Inferred	All gold grains (except one) abraded, 2% py.
	04	04 2,166 1,065	3	No	Yes	Observed	All gold grains abraded, 1% py. 97% of calc. gold contained in one 175x300 micron nugget.
		07 25,000 0	*0	Basal	High	Inferred	Check panned 1/4 conc.; no V.G., 1% py.
	05	03 1,068 621	*1	No	Yes	Observed	Gold grain abraded @ 225x225.
		06 275 165	10	Basal	Yes	No	Six abraded gold grains, 3 irregular, 1 delicate, 1% py.
	06	01 2,157 1,853	4	Chance	Yes	Observed	All gold grains abraded. 92% of calc. gold contained in one 250 x 400 micron nugget.
		02 856 421	11	Chance	Yes	No	All gold grains (except one) abraded.
	07	02 747 1,076	8	No	Yes	No	All gold grains abraded, 1% py.
		04 2,332 594	9	Basal	High	Inferred	All gold grains (except one) abraded, 1% py., 150 grains aspy. Check panned 1/4 conc.; 1I @ 25x25, 1A @ 75x100, 1% py.
	08	01 1,427 1,778	5	No	Yes	Observed	All gold grains abraded, 2% py. 80% of calc. gold contained in one 200x500 micron nugget.
	09	02 2,747 1,099	6	No	High	Observed	All gold grains (except one) abraded, 1% py. 62% of calc. gold contained in one 150x225 micron nugget.
	11	01 9,096 5,612	7	No	Yes	Observed	All gold grains abraded, 1% py. 95% of calc. gold contained in one 375x575 micron nugget.
		05 1,364 389	*1	Basal	High	Inferred	One irregular gold grain. Check panned 1/4 conc.; no V.G., 0.5% py., 5 grains galena.
	14	02 4,168 4,930	15	Basal	Yes	Observed	All gold grains abraded, 1% py. 91% of calc. gold contained in one 225x450 micron grain.
	15	01 694 386	10	Basal	Yes	No	All gold grains abraded, 1% py.
	17	01 1,159 730	10	Basal	Yes	No	All gold grains abraded, 1% py.
	20	01 2,718 1,661	5	Chance	Yes	Observed	All gold grains abraded, 2% py.
		02 1,025 420	9	Basal	High	Inferred	All gold grains abraded, 2% py. Check panned 1/4 conc.; 1A @ 75x150, 1% py.

Table 7 - Heavy Mineral Gold Anomaly Screening

Hole No.	Gold Anomalies		Grains V.G. (*Not Panned)	1st Phase Screening (Strat. Cont.)	2nd Phase Screening (Good Corr./ Low Assay)	3rd Phase Screening (Nugget Effect)	Remarks	
	Sample No.	Au Assay (ppb) Meas. Calc.						
ME-87- 21	01	428	210	11	No	Yes	No	All gold grains (except one) abraded, 1% py. All gold grains abraded, 2% py. Check panned 1/4 conc.; no V.G., 1% py.
	03	2,751	947	4	Basal	High	Inferred	
25	02	6,491	22	*1	Basal	High	Inferred	One abraded gold grain. Check panned 1/4 conc.; no V.G., 1% py., 1 grain galena.
30	01	6,507	3,887	6	No	Yes	Observed	All gold grains (except one) abraded, 1% py. 99% of calc. gold contained in one 250x600 micron nugget.
31	03	23	4,993	3	Basal	Yes	Observed	Two abraded gold grains and 1 irregular, 1% py. 93% of calc. gold contained in one 125x500 micron nugget.
33	01	1,002	0	*0	Basal	High	Inferred	Check panned 1/4 conc.; no V.G., 1% py.
36	01	1,360	882	9	No	Yes	No	All gold grains abraded, 1% py. All gold grains (except one) abraded, 1% py. 46% of calc. gold contained in one 225x325 micron nugget. All gold grains abraded, 2% py. All gold grains abraded, 2% py.
	05	858	2,187	8	No	Yes	Observed	
	07 08	1,028 306	1,425 230	8 15	Chance Basal	Yes Yes	No No	
38	02	2,086	1,120	8	No	Yes	Observed	All gold grains abraded, 2% py. 71% of calc. gold contained in one 150x325 micron nugget
44	01	1,908	760	4	Chance	High	Inferred	All gold grains abraded, 3% py. Check panned 1/4 conc.; no V.G., 3% py. One abraded gold grain. Original gold grain (100x275 micron abraded) found in 1/4 split.
	02	100	1,212	*1	Basal	Yes	Observed	
46	01	1,349	6,387	7	Basal	Yes	Observed	All gold grains (except one) abraded. 97% of calc. gold contained in one 325x400 micron nugget.
48	01	1,749	446	3	No	High	Inferred	All gold grains abraded, 20% py. All gold grains abraded, 5% py. 91% of calc. gold contained in one 275x750 micron nugget.
	06	7,750	6,431	7	No	Yes	Observed	
49	02	2,214	1,610	7	No	Yes	Observed	All gold grains (except one) abraded, 20% py. 68% of calc. gold contained in one 250x350 micron nugget.
51	06	2,090	1,172	8	No	Yes	Inferred	Four abraded, 1 irregular and 3 delicate gold grains, 2% py. ✓
57	01	5,938	6,574	17	Basal	Yes	Observed	Eleven abraded, 2 irregular and 4 delicate gold grains, 5% py. 74% of calc. gold contained in one 400x625 micron nugget. ✓

Table 7 - Heavy Mineral Gold Anomaly Screening (cont'd)

split of the concentrate. Thus either good correlation of measured and calculated assays or a low measured assay generally indicates background noise if the 10 gold grain threshold for dispersion trains is not met.

A third screening method is the direct elimination of nugget anomalies by check panning and analysis. Table 7, in addition to Low Assays and Good Correlation, includes another category - High Assays - which refers to those samples in which the number of gold grains sighted was not sufficient to explain the anomalies obtained. High Assays can be caused by any one of the following:

1. A missed nugget.
2. A sighted nugget for which the actual thickness is greater than the assumed thickness (0.1-0.2 x diameter) used in the assay calculation.
3. The difference in weight between the total concentrate on which the calculation is based and the 3/4 concentrate that is assayed (applies only to samples in which a nugget is present, as fine gold would be evenly distributed through the sample).
4. A large number of missed fine gold grains.
5. Invisible gold in pyrite or other heavy minerals.

Missed nuggets normally account for about 80 percent of High Assays, the thickness and weight factors for 10-20 percent, and fine gold and invisible gold for less than 10 percent. Only the fine gold and invisible gold anomalies are significant.

One method of evaluating anomalies in the High Assay category is to pan the retained 1/4 concentrates, and this was done for the Eby samples (Table 7). An absence or minimal amount of fine visible gold or less than ten percent sulfides in the 1/4 concentrate generally precludes the occurrence of fine gold or sulphide

gold in anomalous concentrations in the 3/4 analytical split, and such anomalies can be assumed to have been caused by a missed or unusually thick nugget. Samples which apparently contain multiple gold particles but do not meet the ten grain minimum (assuming visible gold in the 1/4 and 3/4 is directly proportional) are grouped with nugget anomalies provided sulphide levels are low. Where uncertainty exists the 1/4 concentrate can be analyzed by the non-destructive INA method with the hope of duplicating the 3/4 analysis.

Using the above screening process, thirty-three of the forty anomalies can be confidently discounted as background nugget occurrences. These anomalies fall into two groups -- Group 1 in which the number of gold grains observed during initial processing was sufficient to account for the reported assays but was either below or only marginally above the 10-grain anomaly threshold, and Group 2 (High Assay category) in which little or no gold was seen but the 1000 ppb anomaly threshold was exceeded.

Twenty-three of the thirty-three discounted anomalies are in Group 1. In addition to being caused by sub-anomalous concentrations of gold grains, fourteen of these anomalies lack stratigraphic continuity. Five of the other nine anomalies show chance continuity and the remaining four, by chance, occur in basal samples.

Ten of the discounted anomalies are in Group 2 and unexpectedly yielded assays over 1000 ppb. Three of these lack stratigraphic continuity and could be dismissed on this basis alone; one other anomaly shows chance continuity and the other six occur in basal samples. As a further check, the 1/4 concentrates of all ten samples were panned. No significant gold grain or sulphide mineral concentrations were found. It is probable that all of these anomalies were caused by unsighted nuggets and are of no significance. As a final check, the 1/4 concentrate splits of five samples showing basal continuity have been submitted for analysis to test for occluded gold. If occluded gold is present, the gold content of the 1/4 split should be similar to that of the 3/4 split.

6.3.1 Potentially Significant Gold Anomalies

The seven remaining anomalies occur in seven different drill holes (No. 02, 05, 14, 15, 17, 36 and 51) and probably are as insignificant as the 33 discounted anomalies. However, it is difficult to dismiss these seven anomalies at the 100 percent confidence level, principally because they all occur in basal samples and were all caused by more than 10 gold grains that, in each hole, tend to be of a common size. Moreover, six of the seven anomalies are proximal to the Kirkland Lake - Larder Lake Fault. The major negative factor in each case is that most of the gold grains are abraded. Furthermore, four of the anomalies are surrounded by non-anomalous holes which, considering the detailed nature of the hole pattern, severely limits the potential for a source of significant size. Finally, the gold grains in some cases are so fine that the assays are well under 1,000 ppb, thereby limiting both the size and grade of the source.

6.3.1.1 Hole 02 Anomaly

The anomaly in Hole 02 occurs in Sample 06 at the base of a 10 metre section of Matheson Till resting on bedrock at a depth of 10.5 metres. Processing of this sample produced a total of 14 gold grains; two irregular and 12 abraded with eight of these grains less than 50 microns in diameter and the remainder less than 125 microns. The concentrate assayed 72 ppb Au which is consistent with the number and size of gold grains present. Respective copper, zinc and arsenic analytical results were 125, 25 and 25 ppm. Metal levels in the underlying bedrock are not elevated.

The anomaly is directly over the Kirkland Lake - Larder Lake fault but the abraded nature of the gold grains suggests a distal source. Moreover the anomaly is very weak and is not repeated in nearby drill holes. In summary, the anomaly was probably caused by the cluster effect and the gold grains are part of the normal background for the area.

6.3.1.2 Hole 05 Anomaly

The anomaly in Hole 05 occurs in Sample 06 at the base of an 11 metre section of Matheson Till overlying bedrock at a depth of 14.5 metres. This sample yielded three delicate, one irregular and six abraded gold grains. The delicate and irregular gold grains were less than 50 microns whereas the abraded grains varied in size from 25 to 150 microns. Only one percent pyrite was noted in the concentrate.

The hole occurs approximately 250 metres down-ice from the interpreted position of the Kirkland Lake - Larder Lake Fault. The underlying bedrock contains, at best, background metal levels. The presence of visible delicate gold does suggest dispersion from a local source, most likely the fault; however, the low concentration and low analytical results are consistent with only a very localized or a low grade source.

6.3.1.3 Holes 14/15/17 Anomaly

The most continuous anomaly and therefore the one that most closely matches known dispersion trains occurs in Holes 14, 15 and 17. Hole 16, which was drilled between Holes 15 and 17, gave near-anomalous results. Inclusion of this hole gives a 300 metre long anomaly trending roughly parallel to the regional 165 degree azimuth of Late Wisconsinan ice flow and straddling the Kirkland Lake - Larder Lake Fault.

The Matheson Till in the anomalous area is only one to two samples thick (Fig. 9) and the anomaly is confined to the basal samples. The gold grain counts range from seven to twenty and all of the grains are abraded. With the exception of a 300 x 725 micron nugget in Hole 14, these grains are all relatively fine (25 to 200 microns). If the 4700 ppb contribution of the nugget is subtracted from the assay for the Hole 14 sample, all of the assays are in the relatively narrow but sub-anomalous range of 82 to 730 ppb. Thus the anomaly has the important dispersion train property of consistency between drill holes but is just as weak as the ones in Holes 02 and 05.

If the anomaly represents a true dispersion train, the abraded nature of the gold grains is problematic. Abraded grains normally indicate considerable transport either by ice alone or by fluvial interglacial transport and concentration (placers in Sangamon gravels) with subsequent reworking by ice. However, anomalous bedrock (slightly altered conglomerate with 175 ppb Au) is present in Hole 13, just 100 metres up-ice from the till anomaly, and Sangamon gravels are not present on the property. This suggests either that the gold was abraded prematurely or more likely that the anomalies in Holes 14, 15 and 17, like many of the other anomalies on the property, were caused by the cluster effect and strictly by chance occur in adjacent drill holes.

6.3.1.4 Hole 36 Anomaly

The anomaly in Hole 36 occurs in Sample 08 at the base of a 12 metre section of Matheson Till resting on bedrock at a depth of 18 metres. Processing of this sample produced 15 abraded gold grains with all grains less than 150 microns in diameter and eight of these less than 50 microns. The concentrate also contained two percent pyrite. An analytical value of 306 ppb Au is consistent with the calculated value of 230 ppb. Bracketing holes are not anomalous. This, combined with the abraded nature of the gold grains and the low assay, strongly suggests that the anomaly was carried by the cluster effect.

6.3.1.5 Hole 57 Anomaly

The Hole 57 anomaly is the only potentially significant anomaly over the komatiitic Larder Lake volcanics in the northeastern part of the drill area. Sample 01 of a thin Matheson Till section overlying bedrock at a depth of 18 metres yielded four delicate, two irregular and eleven abraded gold grains as well as 5 percent pyrite. Fifteen of the seventeen gold grains are finer than 100 microns; the other two are nuggets -- a very large, irregular one of 400 x 625 microns and a smaller abraded one of 300 x 375 microns.

The delicate and irregular grains are probably from a local source while the abraded grains are assumed to represent clustered background gold. The calculated and measured assays show good correlation (6,574 and 5,938 ppb, respectively), and the calculated contribution of the delicate and irregular grains is 4,867 ppb with 99.8 percent of this coming from the large nugget. The probable source is a known green carbonate zone 200 metres up-ice (Plan 2) where considerable trenching and shallow shaft development has been done previously.

6.4 Eby Heavy Mineral Arsenic and Base Metal Anomalies

Thirteen of the 181 overburden samples (7 percent) from 13 drill holes (21 percent) produced assays over the 800 ppm anomaly threshold level for copper or arsenic. Twelve of these 13 samples are anomalous in copper and the other (Sample 34-02) is anomalous in arsenic. Many other overburden samples scattered throughout the drill area display elevated but sub-anomalous levels of copper and arsenic (200-800 ppm). This suggests very weak but widespread mineralization which probably reflects normal variations in the metal contents of the different rock types present in the area. The majority of the zinc assays are less than 200 ppm -- commonly less than 100 ppm -- representing background levels only. No samples contain more than 800 ppm Zn.

A limited screening process similar to that previously used for gold anomalies has been employed to separate background noise from those anomalies which are, or may be, related to significant mineralized sources. The anomalies and screening processes are listed in Table 8. In some cases anomalies are discounted for more than one reason.

As with gold anomalies, one screening method is to eliminate anomalies which have no stratigraphic continuity. Six copper anomalies and the only arsenic anomaly do not show stratigraphic continuity (Table 8). With one exception these anomalies are weak. Sample 51-07 assayed 23,240 ppm Cu which has been traced to drill cuttings of a weakly mineralized boulder. Thus the anomaly is not

Hole No.	Sample No.	Strat. Cont.	Strat. Unit	Assay Values (ppm)			1/4 H.M.C.	Remarks
				Cu	Zn	As		
ME-87-01	02	No	M. Till	1,026	67	41	-	Not significant
09	06	No	M. Till	3,175	-	1	-	Not significant
34	02	No	M. Till	97	204	5,360	-	Not significant
39	01	Basal	M. Till	931	31	18	5% py., 0.2% cpy./bornite	Weak train
40	02	Basal	M. Till	1,208	48	7	10% py., 0.2% cpy.	Weak train
41	03	Basal	M. Till	1,136	58	165	7% py., 0.3% cpy./bornite	Weak train
44	02	Basal	M. Till	1,115	54	54	3% py., 0.2% cpy.	Weak train
46	01	Basal	M. Till	2,319	54	90	15% py., 0.5% cpy.	Weak train
48	02	No	M. Till	907	61	54	-	Not significant
49	02	No	M. Till	982	52	54	-	Not significant
51	07	No	M. Till	23,240	119	97	1% py., 5% cpy., 2% bornite	Not significant (mineralized boulder)
54	02	No	M. Till	1,089	29	178	30% py., 0.1% cpy.	Not significant
59	02	Basal	M. Till	1,116	88	85	5% py., 0.2% cpy./bornite	Weak train

Table 8 - Heavy Mineral Arsenic and Base Metal Anomaly Screening

considered significant. In two cases (Samples 01-02 and 49-02) the copper anomalies by chance coincide with gold anomalies that are equally insignificant.

Five copper anomalies (931-2,319 ppm) clustered in the basal samples of Holes 39, 40, 41, 44, and 46 just south of the Larder Lake komatiites near their intersection with the Kirkland Lake - Larder Lake Fault show real or apparent stratigraphic continuity as does a sixth copper anomaly in the basal sample of Hole 59 over the komatiites. Anomalies of this type can be further qualified by direct mineralogical observation. In the case of the Eby samples, the retained 1/4 concentrates were visually examined under a binocular microscope to ascertain the percentages of copper minerals present relative to the percentage of pyrite. In addition, small incorporated rock chips were observed for the presence of economically viable banded massive base metal sulphides versus less attractive vein-hosted disseminated sulphides. Sufficient percentages of copper minerals (0.2 to 0.5 percent chalcopyrite/bornite, occasionally associated with quartz) were observed to account for all of the anomalous assay results, and no sulphide bonding was observed.

Only background zinc and arsenic values are associated with these concentrates. The most obvious source for the copper is the underlying bedrock which in the five-hole anomaly area is generally described as strongly foliated and sheared basalts with 2-5% quartz carbonate veining. These basalts show slightly elevated copper levels of 83-156 ppm. The komatiite at Hole 59 assayed only 47 ppm Cu but is a soft talc-chlorite-carbonate schist and ODM has observed that similar rocks elsewhere can shed a large volume of copper when glaciated. None of the indicated copper mineralization is considered significant.

7. CONCLUSIONS AND RECOMMENDATIONS

7.1 Gold Potential of the Eby Property

The gold content of the till on the Eby property is high but this is a regional feature found along all major gold structures (Kirkland Lake - Larder Lake Fault, Porcupine - Destor Fault and Casa-Berardi Break) in the Abitibi Belt and is not in itself an indication of the gold potential of the property. Indeed the high background is a nuisance as it appears to be caused principally by visible gold that, due to erratic distribution (the cluster effect) and variations in grain size (the nugget effect), produces many false heavy mineral gold anomalies. Of forty anomalies obtained, only one anomaly in Hole 57 appears to be significant.

The Eby property was acquired by Mary Ellen Resources largely on the basis of the similarity of its geology to that found at the Kerr Addison Mine at Larder Lake. In the Kerr Addison model, gold can be expected to occur in altered (green carbonate) komatiitic flows and interflow sediments of the Larder Lake Group proximal to the Kirkland Lake - Larder Lake Fault. Some workers consider the gold to be entirely epigenetic while others such as Jensen and Hinse (1979) suggest that the gold in the interflow sediments accumulated sygenetically during Cycle II volcanism when the Kirkland Lake - Larder Lake Fault was a basin-marginal structure and that only the gold in the green carbonate zones is epigenetic.

On the Eby property, the Larder Lake komatiites are found only in the northeast and it is in this area that previous exploration has been concentrated. Only 12 of the reverse circulation holes were drilled here and these holes were positioned too far down-ice from the fault (1200 metres) and most of the known green carbonate zones to provide a good test of the gold potential. The only encouragement obtained is the 4,867 ppb delicate/irregular gold grain anomaly in Hole 57, and 99.8 percent of this anomaly is caused by one gold grain. The probable source of the gold is the most southerly of the known green carbonate zones which lies 200 metres up-ice.

Most of the reverse circulation drilling was done on the northwestern part of the property where the Kirkland Lake - Larder Lake Fault forms the contact between Kinojevis basalts and Timiskaming conglomerates. Green carbonate alteration is rare here and the Kerr Addison model does not apply. Any gold mineralization would probably be shear-controlled. Unfortunately most of the shearing occurs in the conglomerate which is generally considered to be younger than the main episode of gold mineralization (Downes, 1979). Some evidence of younger mineralization is provided by the 175 ppb gold anomaly in the conglomerate of Hole 13 but the till over and down-ice from the conglomerate did not yield any anomalies suggestive of concentrated mineralization.

7.2 Gold Targets

Follow-up exploration is recommended for the following three target areas:

- 1) The weak bedrock gold anomaly in Hole 13.
- 2) The green carbonate horizon 200 metres north of Hole 57.
- 3) The untested green carbonate zones further to the north.

7.2.1 Hole 13 Area

One or two core holes should be drilled to cross-section the geology under Hole 13 and determine whether the gold anomaly in the conglomerate is significant. Mary Ellen should select the hole sites using the geophysical data obtained from previous surveys.

7.2.2 Hole 57 Area

It is recommended that Mary Ellen review the assessment files to determine whether the gold found by early workers in the green carbonate zone 200 metres north of Hole 57 matches that found in the till at Hole 57 in terms of tenor and particle size. Some diamond drilling may be warranted west and southwest of the main green carbonate outcrop (Plan 2).

7.2.3 Northern Green Carbonate Zones

The paucity of positive results from the present drilling is due mainly to the unfavourable geology in the areas that were selected for drilling. The reverse circulation method itself cannot be faulted as it produced positive results in the one instance where holes were drilled near known mineralization. This suggests that the method could be applied successfully over the more northerly green carbonate zones that are closer to the Kirkland Lake - Larder Lake Fault. Much of this area is overburden covered and previous work has been concentrated around the few outcrop clusters that are present (Plan 2). Approximately 30 holes would give detailed coverage on a 100 x 200 metre grid pattern. Hole depth would probably average 10 metres and all-inclusive costs \$100/metre for a total project cost of \$30,000.

7.3

Property Acquisition

The Larder Lake komatiites strike in a northwesterly direction onto the adjoining property where they are covered by Kenogami Lake. Acquisition of this property would provide a broader target area in which to test the Kerr Addison model. The lake-covered area could probably be evaluated very effectively using reverse circulation drilling.



S. Averill, President

CERTIFICATE - STUART A. AVERILL

I, Stuart A. Averill, residing at 192 Powell Avenue, Ottawa, Ontario hereby certify as follows:

That I attended the University of Manitoba at Winnipeg, Manitoba and graduated with a B.Sc. (Hons.) in Geology in 1969.

That I have worked continuously in the field of mining exploration geology since 1971.

That I am President and a principal owner of Overburden Drilling Management Limited, 107-15 Capella Court, Nepean, Ontario, an independent geological consulting company that I founded in 1974.

That I qualify for and have recently applied for fellowship in the Geological Association of Canada.

That this technical report is based on data gathered on the subject property by employees of Overburden Drilling Management Limited and interpreted by me.

That I have no direct or indirect interest in Mary Ellen Resources Limited.



Stuart A. Averill, B.Sc. (Hons.)

Dated at Ottawa, Ontario this 9th day of June 1987

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APPENDIX A
REVERSE CIRCULATION DRILL HOLE LOGS

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

ELEV 331m

SITE NO-38

DATE 21/11/1987

HOLE NO ME-87-01 LOCATION LIVE 37W, 6.15 N
GEOLOGIST SHEP DRILLER YALSAK BIT NO. JCC388 BIT FOOTAGE 0-10m

SHIFT HOURS

MOVE TO HOLE _____

TO _____

DRILL 0930-1115

TOTAL HOURS

MECHANICAL DOWN TIME _____

CONTRACT HOURS

DRILLING PROBLEMS _____

OTHER _____

MOVE TO NEXT HOLE 1115-1130

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG					
0				0-0.1 ORGANICS					
0.1				0.1-0.4 <u>OJIBWAY II CLAY</u>					
1				beige soft clay					
2			C1	0.4-1.0 <u>MATHESON TILL</u>					
3				- grey to grey beige fine-med sand					
4			C2	- pebble 50% vol/seds					
5				20% granitoid					
6				- cobbly					
7			C3	5.0-5.2 boulder					
8				(mafic vol)					
9			C4	9.0-10.0 <u>BEDROCK</u>					
10			C5	(TIMISKAMING CONGLOMERATE)					
11				- dark green colour with lighter					
12				coloured granite and jasper clasts					
13				- distinct clasts with the absence					
14				of a matrix					
15				- massive					
16				- conglomerate (variable grain size)					
17				(Bit failed while drilling this unit)					
18				10.0 E.O.11					
19									
20									

NOTE Poor sample return

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE M0-37

DATE JAN 19 19 87

HOLE NO ME 87-02 LOCATION L37W 4+90N ELEV 329m

GEOLOGIST SHELP DRILLER HALSALL BIT NO. C865554 BIT FOOTAGE 9-12m

SHIFT HOURS
TO _____

MOVE TO HOLE 1115-1130

TOTAL HOURS

DRILL 1130-1310

MECHANICAL DOWN TIME _____

DRILLING PROBLEMS _____

CONTRACT HOURS

OTHER _____

MOVE TO NEXT HOLE 1310-1320

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG					
0				C-C8 ORGANICS					
0.8				C8-10.5 <u>MATHESON TILL</u>					
1									
2				grey to grey-buge fine-med sand					
3				- pebble 90% volcsids					
4				10% granitoid					
5				- cobbly					
6				30-50 5-15% grey gritty clay					
7									
8				80-82 30% grey gritty clay					
9				90-10.5 cobbly section					
10				- matrix poor					
11				- greater than 95% calcareous					
12				10.5-12.0 <u>BEDROCK</u> (Basalt)					
13				- dark green					
14				- fine grained					
15				- strongly calcareous					
16				- < 5% qb-carbonate veinlets					
17				- massive -> weakly foliated					
18				12.0 E.C.H					
19									
20									

NOTE Very poor return.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

SITE NO-36

DATE JAN 19 1987 HOLE NO MF87-03 LOCATION L37W 3190N ELEV 326m
 GEOLOGIST SHELP DRILLER HALSALL BIT NO CB45554 BIT FOOTAGE 12.0-27.5m
 SHIFT HOURS _____ MOVE TO HOLE 13.0-13.20
 _____ TO _____ DRILL 13.20-15.10
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 15.10-15.20

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-2.0		NO SAMPLE
2.0		2.0-14.0		<u>MATHESON TILL</u>
3.0	Δ		01	- grey beige - beige fine to med sand
4.0	Δ			- pebble 90% vol / seeds 10% granitoid
5.0	Δ		02	- cobbly
6.0	Δ			3.6-3.7 20% grey gritty clay
7.0	Δ		03	
8.0	Δ		04	4.6-12.0 cobbly section - 100% volcanics - matrix poor
9.0	Δ		05	
10.0	Δ			12.0-12.2 20% dark grey hard, gritty clay
11.0	Δ		06	
12.0	Δ			13.8-14.0 30% dark grey gritty clay
13.0	Δ		07	
14.0	Δ		08	<u>BEDROCK</u> (mafic volcanic)
15.0				- dark green - fine-grained - massive → weakly foliated - strongly calcareous - < 1% qtz-carbonate veinlets - fine-grained pyrite veinlets
16.0				
17.0				
18.0				
19.0				
20.0				15.5 E.O.H

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

SITE MO-35

DATE JAN 19 1987 HOLE NO ME-87-04 LOCATION 137 W 270 N ELEV 323M
 GEOLOGIST SHELP DRILLER HALSALL BIT NO. CB48554 BIT FOOTAGE 225
 SHIFT HOURS _____ MOVE TO HOLE 15:10 - 15:20
 _____ TO _____ DRILL 15:20 - 17:00
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG					
0				0-0.2 ORGANICS					
0.2				<u>MATHESON TILL</u>					
1	Δ	01		- grey-grey beige fine-med sand					
2	Δ			- pebbles 90% uolcalsids					
3	Δ	02		10% granitoid					
4	Δ			2.8-3.6 cobbly section					
5	Δ	03		- matrix poor					
6	Δ	04							
7	⊗			6.5-7.2 boulder					
8	Δ	05		(major vol)					
9	Δ	06							
10	Δ	07		10.5-12.0 <u>BEDROCK</u>					
11	▨	08A		(Basalt)					
12	▨	08B		10.5-10.9 strongly oxidized					
13				10-15% quartz-carbonate concretions					
14				10.9-11.7 med-dark green					
15				- foliated, appears sheared (slip planes)					
16				- fine-grained					
17				11.7-12.0 quartz-green					
18				- strongly sheared					
19				12.0 E.O.H.					
20									

NOTE poor return (0.2 - 3.0)

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

DATE JAN 20 19 87 HOLE NO ME87-05 LOCATION L37W, 1+90N ELEV 321m
 GEOLOGIST SHELP DRILLER MARSHALL BIT NO. CB68553 BIT FOOTAGE 0-15.5
 SHIFT HOURS _____ MOVE TO HOLE 9.00 - 9.10
 _____ TO _____ DRILL 9.10 - 11.40
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 11.40 - 11.50

NEW BIT CB68553

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANICS
1				0.2-3.5 GIBWAY SEDIMENTS
2				- grey, soft, pure clay
3				
4				3.5-14.5 <u>MATHESON TILL</u>
5				- grey beige-beige fine-med sand
6				- pebbles 60% vol/seds 40% granite
7				- occasional cobble
8				- occasional lithified silty clasts
9				by 5.6 pebbles 80% vol/seds 20% granitic
10				
11				7.8-8.0 boulder (mafic volc)
12				12.0-12.4 10% dark grey gritty clay
13				
14				14.5-16.0 <u>BEDROCK</u>
15				- mottled dark and olive green
16				- medium grained
17				- sheared
18				- distinct plagiocrystals
19				- scissuring
20				15.5 thin seam of reddish rock (hematite, jasper)
				(Gabbro or coarse-grained Mafic Volc)
				16.0 E.O. 11

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

SITE MO-62

DATE JAN 20 19 87

HOLE NO MERZ-06 LOCATION L37W, C190 N ELEV 321m
GEOLOGIST SHELP DRILLER HALSALL BIT NO. C68553 BIT FOOTAGE 13.5-32.5

SHIFT HOURS

MOVE TO HOLE 11.40 - 11.50

TO

DRILL 11.50 - 13.30

TOTAL HOURS

MECHANICAL DOWN TIME

CONTRACT HOURS

DRILLING PROBLEMS

OTHER

MOVE TO NEXT HOLE 13.30 - 13.40

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0.0		0.0 - 0.2		ORGANICS
0.2		0.2 - 6.6		OUTWASH SEDIMENTS - grey soft, pure clay
6.6		6.6 - 15.8		MATHESON TILL - grey fine-med sand - pebble 90% vol/seds 10% granitoid - cobbly
15.8		15.8 - 17.0		BEDROCK (Basalt) - dark green - fine-grained - massive → moderately foliated - < 0.1% Jasper blcks
17.0				E.O.H

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE MO-63

DATE JAN 20 19 87

HOLE NO ME-87-07 LOCATION L37W Base Line ELEV 321m
GEOLOGIST SDFP DRILLER HALSALL BIT NO CR68553 BIT FOOTAGE 32.5-500m

SHIFT HOURS
TO

MOVE TO HOLE 13.30-13.40

TOTAL HOURS

DRILL 13.40-13.30

CONTRACT HOURS

MECHANICAL DOWN TIME 13.40-13.50, 15.30-16.15 (musty tread problems)

DRILLING PROBLEMS

OTHER

MOVE TO NEXT HOLE 16.15-16.20

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.2		ORGANICS
1		0.2-6.5		OUTWASH SEDIMENTS - dark grey, soft, pure, clay
2				
3				
4				
5				
6				
7		6.5-15.8		MATHESON TILL - grey fine-med sand - pebble 80% vol/seds 20% granitoid - occasional cobble
8			c1	
9				
10			c2	
11			fcc2	9.0-9.4 boulder (mafic volc)
12			RETUR	
13			c2	11.0-13.8 cobbly section
14			c3	
15			c4	
16		15.8-17.4		BEDROCK - dark green - rubbly appearance, fragmented possibly sheared - strongly calcareous
17			c5	
18				
19				
20				

(Mafic volcanic)
17.4 E.O.H

OVERBURDEN DRILLING MANAGEMENT LIMITED
 REVERSE CIRCULATION DRILL HOLE LOG

SITE 110-64

DATE JAN 20 19 87 HOLE NO ME-87-08 LOCATION L37W 1100 S
 GEOLOGIST SHIEP DRILLER HALLSALL BIT NO. J000389 BIT FOOTAGE 0-19.0m
 SHIFT HOURS _____ MOVE TO HOLE 16.15-16.20
 _____ TO _____ DRILL 16.20-17.00
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

JAN 21/87 drill 07.30 - 08.30
 move 08.30 - 08.55

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.4		ORGANICS
1		0.4-9.5		OJIBWAY SEDIMENTS - dark grey, soft, pure clay
2				
3				
4				
5				
6				
7				
8				
9				
10		9.5-17.4		MATHESON TILL - grey fine-med sand - pebble 90% vol/seds 10% granitoid - occasional cobble
11			01	
12			02	
13			03	
14			04	
15			05	
16			06	
17		16.0-17.4		grey fine-med sand - ≈ 10% pebbles at end of section possibly Missisquoi Sediments
18				
19		17.4-19.0		BEDROCK (BASALT) - dark green - appears sheared (slip planes) - fine-grained - < 5% Jasper
20				

19.0 ± 0.1

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 21 1987 HOLE NO ME87-09 LOCATION SITE 110-27 41K0W 2+50N FLEU 321m
 GEOLOGIST SHARP DRILLER MALSALL BIT NO. J000389 BIT FOOTAGE 19-38.2m
 SHIFT HOURS _____ MOVE TO HOLE 08.30 - 08.55
 _____ TO _____ DRILL 08.55 - 11.00
 TOTAL HOURS _____ MECHANICAL DOWN TIME 09.10 - 09.20 water pump
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 11.00 - 11.10

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0 - 0.4		ORGANICS
1		0.4 - 17.6		<u>MATHESON TILL</u> - grey - grey beige fine-med sand - pebbles 80% volc /sds 20% granitoid (5-10% Jasper) - occasional cobble
2			01	
3				
4			02	
5				
6			03	
7				
8			04	
9				
10			05	
11				
12			06	
13				
14			07	
15				
16			08	
17				
18			09	
19				
20			10	
		17.6 - 19.2		<u>BEOROCK</u> (Basalt) - dark green - foliated, appears sheared - 5-10% sph-carbonate (pink) veinlets with trace fine-grained pyrite
		19.2		E.O.H

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE MO-26

DATE JAN 21 1987 HOLE NO ME47-10 LOCATION L91W0W 1150N ELEV 321m
 GEOLOGIST SHARP DRILLER HALLSALL BIT NO. J000389 BIT FOOTAGE 382-498m
 SHIFT HOURS _____ MOVE TO HOLE 11.00 - 11.10
 _____ TO _____ DRILL 11.10 - 12.45
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 12.45 - 12.55

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0 - 0.4 ORGANICS
1				0.4 - 2.0 <u>QJIBWAY SEDIMENTS</u> - grey, soft, pure clay
2				2.0 - 4.5 <u>MATHERSON TILL</u>
3			01	2.0 - 3.5 cobbly section - 80% calcisid - 20% granitoid - matrix poor - grey fine-med sand
4				3.5 - 4.5 grey beige - beige fine-med sand - pebbles 80% calcisids - 20% granitoid
5			02	
6				4.5 - 6.5 <u>BEAROCK</u> (Basalt) - dark green - foliated - fine-grained - strongly calcareous - 2-3% calcite veinlets - < 1% Jasper
7				6.5 E.O.H
8				
9				
10				
11				
12				
13				
14				
15				
16				
17				
18				
19				
20				

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE MO-24

DATE JAN 21 19 87

HOLE NO ME-87-12 LOCATION 41W 0+505 ELEV 321m

GEOLOGIST SHELPS DRILLER HALSALL BIT NO CR6555 BIT FOOTAGE 0-7m

SHIFT HOURS

MOVE TO HOLE 14.15 - 14.25

TO

DRILL 14.25 - 15.05

TOTAL HOURS

MECHANICAL DOWN TIME

CONTRACT HOURS

DRILLING PROBLEMS

OTHER

MOVE TO NEXT HOLE 15.05 - 15.45

New Bit

New Sub

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	>>>			C-0.4 ORGANICS
1				C.4-4.0 <u>CTIBWAY SEDIMENTS</u>
2				- grey, pure, soft clay
3				
4				4.0-5.2 <u>MATHISON TILL</u>
5			O1	- grey fine-med sand.
6			O2	- pebbles 70% volc/seds 30% granitoid
7				5.2-7.0 <u>BEDROCK</u> (Basalt)
8				- dark green
9				- fine grained
10				- appears sheared (slip planes)
11				- strongly calcareous
12				- 3-5% quartz carbonate (pink) veinlets
13				
14				7.0 E.O.H
15				
16				
17				
18				
19				
20				

OVERBURDEN DRILLING MANAGEMENT LIMITED
 REVERSE CIRCULATION DRILL HOLE LOG

SITE NO-21

DATE JAN 21 19 87 HOLE NO ME-87-13 LOCATION 45W, 2190N ELEV 321m
 GEOLOGIST SHIELP DRILLER HOLSOPE BIT NO C86555 BIT FOOTAGE 7-11m
 SHIFT HOURS MOVE TO HOLE 15.25 - 15.45
 TO DRILL 15.45 - 17.00
 TOTAL HOURS MECHANICAL DOWN TIME _____
 DRILLING PROBLEMS _____
 CONTRACT HOURS OTHER _____
 MOVE TO NEXT HOLE _____

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0-0.4				ORGANIC
0.4-2.0				<u>OUTWASH SEDIMENTS</u> - grey, pure, soft clay
2.0-2.4			01 02	<u>MATHESON TILL</u> - grey fine-med sand - pebbles 80% volc /sects 20% granitoid
2.4-4.0				<u>BEDROCK</u> (conglomerate) - dark green matrix surrounding light pink granitic clasts - mafic component (> 70%) is strongly foliated, fine grained - granitic clasts - rounded, fine-med grained and range in size from < 2mm to > 1cm.
3.4-3.6				strongly oxidized section of mafic material - appearance of small oxidized nodules (Fe, Mn)
4.0				E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

SITE MG-20

DATE JAN 22 1987 HOLE NO ME 87-14 LOCATION 45 W, 1190 N ELEV 321m
 GEOLOGIST SHEEP DRILLER HALSALL BIT NO. C36K555 BIT FOOTAGE 110-215m
 SHIFT HOURS _____ MOVE TO HOLE 7.50 - 8.10
 _____ TO _____ DRILL 08.10 - 9.00
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 9.00 - 9.15

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANIC
1				0.2-6.0 <u>QJIBWAY SEDIMENTS</u> grey, soft, pure clay
2				
3				
4				
5				
6				6.0-9.0 <u>MATHESON TILL</u>
7				- grey fine-med sand
8				- pebbles 90% volc/beds 10% granite trid
9				9.0-10.5 <u>BEDROCK</u>
10				(Conglomerate)
11				- mixed Jasper, basalt and granite chips
12				- basalt - dark-green, foliated
13				- granite clasts are med-coarse grained, pink
14				- clasts are var-sized <2mm -> >> 1cm
15				- mafic material comprises >60% of rock
16				- <0.1% fuchs. te
17				10.5 E.O.H
18				
19				
20				

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE MO-19

DATE JAN 22 1987

HOLE NO ME 47-15 LOCATION 45 W 0190N

GEOLOGIST SHELD DRILLER HALLSILL BIT NO CB16555 BIT FOOTAGE 26.5-31.0M

SHIFT HOURS

MOVE TO HOLE 9.00-9.15

TO

DRILL 9.15-10.15

TOTAL HOURS

MECHANICAL DOWN TIME

CONTRACT HOURS

DRILLING PROBLEMS

OTHER

MOVE TO NEXT HOLE 10.15-10.30

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.5		ORGANICS
1		0.5-7.0		QUIBWAY SEDIMENTS - grey, pure, soft clay
2				
3				
4				
5				
6				
7		7.0-8.0		MATHESON TILL - grey fine-med sand - pebble 50% cde/seds 20% granofed - occasional cobble
8				
9				
10				
11		8.0-9.5		BEDROCK (Conglomerate) - >70% dark green mafic material surround granitic clasts of various sizes (<2m-->1cm) - mafic material is fine-grained and foliated - granitic clasts - fine grained to coarse grained
12				
13				
14				
15				
16				
17				
18				
19				
20				
		9.5		E.O.H

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

SITE MO-19

DATE JAN 22 1987 HOLE NO ME 87-15 LOCATION 45 W 0190N
 GEOLOGIST SHELP DRILLER HARVEY BIT NO. CB61255 BIT FOOTAGE 265-310.0
 MOVE TO HOLE 9.00-9.15
 DRILL 9.15-10.15
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 MOVE TO NEXT HOLE 10.15-10.30

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	▲▲			0 - 0.5 ORGANICS
1				0.5 - 7.0 QJIBWAY SEDIMENTS
2				- grey, pure, soft clay
3				
4				
5				
6				
7				7.0 - 8.0 <u>MATHESON TILL</u>
8	▲▲	C1		- gray fine-med sand
9	▲▲	C2		- pebble 80% cde / seds 20% granitoid
10				- occasional cobble
11				8.0 - 9.5 <u>BEDROCK</u>
12				(conglomerate)
13				- > 70% dark green mafic material
14				surround granitic clasts of
15				various sizes (< 2m -> 1cm)
16				- mafic material is fine-grained
17				and foliated
18				- granitic clasts - fine grained to coarse grained
19				
20				9.5 E.O.H

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE NO-17

DATE JAN 22 1987 HOLE NO ME87-17 LOCATION 45 W, 1405 ELEV 321m
 GEOLOGIST SHARP DRILLER WALSALL BIT NO. CR6525 BIT FOOTAGE 90.5-98.0m
 SHIFT HOURS _____ MOVE TO HOLE 11.10-11.20
 _____ TO _____ DRILL 11.20-12.05
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 12.05-12.15

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-1.5 ORGANICS.
1				1.5-5.0 OTIBWAY SEDIMENTS
2				-grey, pure, soft clay
3				
4				
5				50-6.2 MATHESON TILL
6			01	50-54 cobbly section
7			02	>95% volc -matrix poor
8				54-6.2 - grey fine-med sand
9				-pebbles >90% volc/seeds <10% granitoid
10				6.2-7.5 BEDROCK
11				(Basalt)
12				-dark green
13				-fine-grained
14				-foliated, possible shearing
15				-strongly calcareous
16				- <1% qtz-carbonate veining
17				- <0.1% pyrite
18				-trace jasper
19				
20				7.5 E.O.H

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE mo-83

DATE JAN 22 19 87 HOLE NO ME-87-18 LOCATION 45 W, 2+10 S ELEV 321 m
 GEOLOGIST SHEP DRILLER HALSALL BIT NO. CR68555 BIT FOOTAGE 480-545m
 SHIFT HOURS _____ MOVE TO HOLE 12.10 - 12.15
 _____ TO _____ DRILL 12.15 - 13.20
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 13.20 - 14.30

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	>>			0-0.5 ORGANICS
1				0.5-5.8 <u>CLAYWAY SEDIMENTS</u> - grey, pure, soft clay
2				
3				
4				
5				
6	Δ		O1	5.8-6.6 <u>MATHESON TILL</u> - grey fine-med sand
7	Δ		O2A	- pebbles >95% vol/secs
8			C2B	6.6-8.6 <u>BEDROCK</u> 6.6-7.6 dark green with 20-30% of rock comprised of buff coloured qb-carbonate veinlets - veinlets contain fine grained pyrite and a trace of fuchsite - mafic rock is fine grained and appears sheared - 5% leucosene - <0.1% fine-grained pyrite (Basalt) 7.6-8.6 - mottled dark green and pink - medium to coarse grained - <1% carbonate veinlets - massive - 20-17% pyrite (Granodiorite)
9				
10				
11				
12				
13				
14				
15				
16				
17				
18				
19				
20				

8.6 E.O.H

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE NO-16

DATE JAN 22 19 87 HOLE NO ME 87-19 LOCATION 47W 01505 ELEV -321m
 GEOLOGIST SAFLP DRILLER HALSALL BIT NO. CBG1961 BIT FOOTAGE 0-28.5
 SHIFT HOURS _____ MOVE TO HOLE 13.20-14.30
 _____ TO _____ DRILL 14.30-15.40
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 15.40-16.00

NEW BIT

Pg 1 of 2

DEPTH IN METRES	GRAPHIC LOG INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	^ ^ ^		0-0.5 ORGANICS
1			0.5-2.50 QUIBWAY SEDIMENTS
2			- grey, soft, pure clay
3			
4			
5			
6			
7			
8			
9			
10			
11			
12			
13			
14			
15			
16			
17			
18			
19			
20			

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

DATE _____ 19 _____ HOLE NO. ME-87-19 LOCATION _____
 GEOLOGIST _____ DRILLER _____ BIT NO. _____ BIT FOOTAGE _____
 SHIFT HOURS _____ MOVE TO HOLE _____
 _____ TO _____ DRILL _____
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

Pg 2 of 2

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
21				
22				
23				
24				
25	Δ			250-27.4 <u>MATHESON TILL</u>
26	Δ	01		- grey fine-med sand
27	Δ	02		- pebbles 90% vol/seds
28		03		10% granitoid
29				27.4-29.0 <u>BEDROCK</u>
30				(mafic → ultramafic vol)
31				27.4-28.2 olive-green
32				- foliated, appears sheared
33				- fine-grained
34				- calcareous
35				28.2-29.0 dark green with 10-20% ytz-carbonate veinlets (pink)
36				- up to 50% veinlets in some sections
37				- < 0.1% sulphide
38				- trace fuchsite
39				29.0 E.O.H.
40				

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE no-13

DATE JAN 22 19 87

HOLE NO ME-87-20 LOCATION LSOW, 07605 ELEV 321m
GEOLOGIST SHELP DRILLER HOLSALL BIT NO. CB68461 BIT FOOTAGE 28.5-40.5m

SHIFT HOURS
TO _____

MOVE TO HOLE 15.40 - 16.00
DRILL 16.00 - 17.00

TOTAL HOURS _____

MECHANICAL DOWN TIME _____

CONTRACT HOURS _____

DRILLING PROBLEMS _____

OTHER _____

MOVE TO NEXT HOLE _____

JAN 23/87 maintenance 08.00 - 10.15
drill 10.15 - 11.20
move 11.20 - 11.30

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-1.5		ORGANICS
1		1.5-8.8		<u>OTIRWAY SEDIMENTS</u> - grey, pure, soft clay
2				
3				
4				
5				
6				
7				
8				
9		8.8-10.6		<u>MITHUESON TILL</u> - grey fine-med sand - pebbles > 95% vclt/seeds
10				
11		9.4-9.6		boulder (granite)
12		10.6-12.0		<u>BEDROCK</u> (Conglomerate) - consists of mafic volcanic and granitic clasts in a mafic matrix - mafic component (dark green) - > 60% foliated - fine-grained - granitic component (pink) - fine-med grained - < 1% sulphides - 17% jasper - vari sized clasts = 2mm to > 1cm
13				
14				
15				
16				
17				
18				
19				
20				

120 E.O.H.

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE M0-12

DATE JAN 23 19 87 HOLE NO ME-87-21 LOCATION L50W 11665 ELEV 321m
 GEOLOGIST SHHELP DRILLER HALSALL BIT NO. CB68961 BIT FOOTAGE 46.5-550m
 SHIFT HOURS _____ MOVE TO HOLE 11.20-11.30
 _____ TO _____ DRILL 11.30-12.50
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 12.50-13.00

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0-1.0				ORGANICS
1.0-9.0				<u>STIBWAY SEDIMENTS</u> -grey, soft, pure clay
9.0-13.0				<u>MATHESON TILL</u> -grey fine-med sand -pebbles 70% voltseds 30% granitoid -occasional pebble by 11.0 pebbles 90% voltseds 10% granitoid
13.0-14.5				<u>BEDROCK</u> (altered mafic - ultramafic volc) - med-dark yellow green - foliated, appears sheared - fine-grained - > 10% quartz-carbonate veinlets (pink) - < 1% Jasper - trace pyrite - < 0.1% sulphides
14.5				E.O.H

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

DATE JAN 23 19 87 HOLE NO ME-87-22 LOCATION SITE mo-11 450W 2160S ELEV 321m
 GEOLOGIST SHELPS DRILLER HALSALL BIT NO. CB6894L BIT FOOTAGE 550-67.5m
 SHIFT HOURS _____ MOVE TO HOLE 12.50-13.00
 _____ TO _____ DRILL 13.00-13.50
 TOTAL HOURS _____ MECHANICAL DOWN TIME 13.50-17.00 Clutch replacement
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

Jan 24 - Down time - clutch replacement

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-1.0		ORGANICS.
1	^ ^ ^	1.0-9.4		OTIBWAY SEDIMENTS - gray, soft, pure clay.
2				
3				
4				
5				
6				
7				
8				
9		9.4-11.2		MATHIESON TILL - grey fine-med sand - pebbles 70% voltseds 30% granitoid
10	Δ		01	
11	• Δ		02	
12			03	
13		11.2-12.6		BEDROCK (Basalt) - dark green -> black - fine-grained - siliceous - sheared - 5% of carb. (pink & white) veinlets - < 0.1% fine-grained pyrocl.
14				
15				
16				
17				
18				
19				
20				12.6 E.O.H.

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

SITE # 10

DATE JAN 25 19 87 HOLE NO ME-87-23 LOCATION L50 W 31605. ELEV 321m
 GEOLOGIST SHELD DRILLER HALSALL BIT NO CB6846 BIT FOOTAGE 67.5-79cm
 SHIFT HOURS _____ MOVE TO HOLE 12.00-12.15
 _____ TO _____ DRILL 12.15-14.55
 TOTAL HOURS _____ MECHANICAL DOWN TIME 15.00 - INJECTOR PROBLEMS -
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 14.55-15.00

NEW BIT CB68557 11.5-12.5 = 1m

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	>>			0-0.5 ORGANIC
1	>>			0.5-10.4 <u>OTJIBWAY SEDIMENTS</u>
2				grey, soft, pure clay
3				
4				
5				
6				
7				
8				
9				
10				10.4-11.0 <u>MATHESON TILL</u>
11	▲		01	- grey fine-med sand
12	▲		02	- pebbles 70% volc/seeds
13				30% granitoid
14				11.0-12.5 <u>BEDROCK</u>
15				- med green
16				- fine-grained
17				- siliceous
18				- appears sheared (slip faces.)
19				- 1% qtz - carbonate veinlets
20				12.5 E.O.H.

OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG

DATE JAN 26 1987 HOLE NO ME-87-24 LOCATION SITE # 89 659W, 7125 S ELEV 321M
 GEOLOGIST S. HELL DRILLER HALLSALL BIT NO CA66557 BIT FOOTAGE 12.5-20.0
 SHIFT HOURS _____ MOVE TO HOLE 7.30-8.00
 _____ TO _____ DRILL 9.00-10.40
 TOTAL HOURS _____ MECHANICAL DOWN TIME 9.00-9.00 (Progen line, generator problems)
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 10.40-10.50

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANICS
1				0.2-4.6 <u>ATIBWAY SEDIMENTS</u> - grey, pure, soft clay
2				
3				
4				
5	Δ Δ		01	4.6-5.8 <u>MATHESON TILL</u> - grey fine-med sand - pebbles 70% vol/seds 30% granitoid
6	Δ Δ		02	
7				
8				5.8-7.5 <u>BEDROCK</u> (Ultramafic volc) - dark grey to green black - line-grained - appears sheared - 1-2% pyrite in shears (slip planes) and as disseminations - minor saussuritization - moderately hard - weakly calcareous - < 1% gk-carbonate veinlets with hematite.
9				
10				
11				
12				
13				
14				
15				
16				7.5 E.O.H.
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 26 19 87 HOLE NO ME-87-25 LOCATION SITE #3, L59 W, L1005 FLAV 3210
 GEOLOGIST SAELP DRILLER HALSALL BIT NO B68557 BIT FOOTAGE 20.0-27.5
 SHIFT HOURS _____ MOVE TO HOLE 10.40 - 10.50
 _____ TO _____ DRILL 10.50 - 12.45
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 12.45 - 12.55

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.5		ORGANICS.
1		0.5-4.6		<u>OTIBAWAY SEDIMENTS</u> - grey beige to beige, soft, pure clay by 2.0 gray clay
2				
3				
4				
5		4.6-6.5	01	<u>MATHESON TILL</u> - grey fine-med sand
6			02	- pebble 60% vol/seds 40% granitoid
7			03	- occasional cobble.
8				
9		5.0-5.4		boulder (granite)
10		5.4-5.8		cobbly section
11		6.5-7.5		<u>BEDROCK</u> (Gabbro)
12				- mottled dark green and light green
13				- coarse grained
14				- massive
15				- magnetite - distinct magnetite crystals
16				- < 0.1% pyrite
17		7.5		E.O.H (bit finished)
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 26 19 87 HOLE NO ME-84-26 LOCATION SITE 49, L 54W, 5100S ELEV 321m
 GEOLOGIST SHELP DRILLER HALSALL BIT NO. CB68556 BIT FOOTAGE 0-9.5
 SHIFT HOURS _____ MOVE TO HOLE 12.45-12.55
 _____ TO _____ DRILL 12.55-14.15
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 14.15-14.25

NEW BIT CB68556

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.4		ORGANICS
1		0.4-6.5		<u>OTIBWAY SEDIMENTS</u> - grey, soft, pure clay
2				
3				
4				
5				
6				
7		6.5-8.0		<u>MATHESON TILL</u> - grey fine-med sand - pebble 60% vol/seds 40% granitoid
8				
9				
10		8.0-9.5		<u>BEDROCK</u> (Gabbro) - dark green - med-coarse grained - sheared (slip planes) <i>obscurred grain size and texture</i> - <<0.1 % pyrite
11				
12				
13				
14				
15				
16		9.5		E.O.H.
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 26 19 87 HOLE NO ME-87-27 LOCATION SITE #5, L54W, 4100S ELEV 321M
 GEOLOGIST SHUEP DRILLER MALSALL BIT NO. CB68556 BIT FOOTAGE 9.5-23.5
 SHIFT HOURS _____ MOVE TO HOLE 14.15-14.25
 _____ TO _____ DRILL 14.25-15.40
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 15.40-15.50

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	^ ^			0-0.2 ORGANICS
1				0.2-4.8 <u>OUTERWAY SEDIMENTS</u> - grey, soft, pure clay
2				
3				
4				
5	Δ		01	4.8-12.5 <u>MATHESON TILL</u> - grey fine med sand - pebble 70% vol/seds 30% granitoid - occasional cobble. 5.8-6.0 boulder (magic volc).
6	⊗		c2	
7	Δ		c3	
8	Δ		c4	
9	Δ		c5	
10	Δ		c6	
11	Δ		c7	
12	Δ			12.5-14.0 <u>BEDROCK</u> (magic-ultramafic volc) - mottled dark green and yellow - fine-grained - schistose, strongly fractured and/or sheared. - 5-10% qtz-carbonate veinlets - 20.1% pyrite associated with veinlets - trace fuchsite, jasper.
13				
14				
15				
16				
17				
18				
19				140 E.O.H.
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

SITE #6

DATE JAN 26 19 87 HOLE NO ME-87 28 LOCATION L54W 3+255 ELEV 321m
 GEOLOGIST SHARP DRILLER HALSALL BIT NO CBW554 BIT FOOTAGE 235-305
 SHIFT HOURS _____ MOVE TO HOLE 15.40-15.50
 _____ TO _____ DRILL 15.50-17.00
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 CONTRACT HOURS _____ DRILLING PROBLEMS _____
 _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

First two attempts, no fill, intersect bedrock at 1.5m.
 - moved to 3+255.

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANICS
1				0.2-3.5 <u>QJIBWAY SEDIMENTS</u> - beige, soft, pure clay - by 1.5 grey clay
2				
3				
4			01	3.5-5.6 <u>MATHESON TILL</u> - grey fine-med sand - pebbles 70% volc/seds 30% granitic
5			02	
6			03	
7				5.6-7.0 <u>BEDROCK</u> (Conglomerate) - dark olive green mafic component with what appear to be multistage qb-carbonate veinlets, red granite and sedimentary clasts - dark green rock - fine-grained - bleached - no small clasts occur in mafic matrix - mafic-granite contacts appear to represent granite clasts in a mafic matrix
8				
9				
10				
11				
12				
13				
14				
15				
16				
17				7.0 E.O.H.
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 27 1987 HOLE NO ME-87-29 LOCATION SITE #28, L91W, 3150N ELEV 322m
 GEOLOGIST SHELP DRILLER HALSOB BIT NO. C84554 BIT FOOTAGE 305-490
 SHIFT HOURS _____ MOVE TO HOLE 07.30 - 07.45
 _____ TO _____ DRILL 07.45 - 10.05
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 10.05 - 11.00

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANICS
1			Poor 01	0.2-11.5 MATHESON TILL
2			Return	- grey beige to beige fine-med sand
3			02	- pebbles 80% volc / sed 20% granitoid
4			03	- very cobbly
5			04	
6			05	
7			06	
8			07	
9			08	
10			09	
11			10	
12			11	11.5-13.5 BEDROCK (conglomerate)
13			12	- mosaic of dark green (mafic matrix), light pink (granite clasts) and red (jasper clasts)
14			13	- clast varisized < 2mm to > 1cm
15			14	- mafic component -> foliated, fine-grained
16			15	- granite component -> fine-grained
17			16	- < 0.1% fine-grained pyrite
18			17	
19			18	
20			19	13.5 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 27 19 87 HOLE NO ME-87-30 LOCATION SITE 91, L33W, 4100 N; ELEV 324 m
 GEOLOGIST SHELP DRILLER HALSALL BIT NO. CB66556 BIT FOOTAGE 91.0-99.6
 SHIFT HOURS _____ MOVE TO HOLE 10.05-11.00
 _____ TO _____ DRILL 11.00-12.35
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 12.35-12.45

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
			Poor	0-0.2 ORGANICS
1			01	0.2-8.0 <u>MATHISON TILL</u>
2		Return		- beige fine-med sand
3		Poor		- pebbles 80% vol/seds 20% granitoid
4			02	- very cobbly
5		Return		2.2-2.6 boulder (granitic)
6			03	by 4.0 matrix is gray fine-med. sand.
7			03	6.8-7.2 boulder (mafic vol).
8			04	7.6-8.0 cobbly section
9			05	100% vol. matrix poor
10				8.0-8.6 <u>BEDROCK.</u> (Basalt)
11				8.0-8.2 - med. olive green
12				- fine-grained
13				- foliated
14				- strongly oxidized
15				8.2-8.3 gk vein
16				8.3-8.6 same as 8.0-8.2
17				8.6 E.O.H (tricone bit cones left at bottom of hole).
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 27 1987 HOLE NO ME-87-32 LOCATION SITE #86, 133W, 2+00N, ELEV 322m
 GEOLOGIST SHELPS DRILLER HALSALL BIT NO C84613 BIT FOOTAGE 8.5-18.5m
 SHIFT HOURS _____ MOVE TO HOLE 14.40-14.50
 _____ TO _____ DRILL 14.50-15.45
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 15.45-15.55

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	> ^			0-0.2 ORGANICS
1				0.2-6.2 <u>QJIBWAY SEDIMENTS</u>
2				beige, soft pure clay
3				by 3.0 grey clay
4				
5				
6				6.2-8.0 <u>MATHESON TILL</u>
7	Δ	01		- grey fine-med sand
8	Δ	02		- pebbles 80% vol/seds 20% granitoid
9		03		
10				8.0-10.0 <u>BEDROCK</u>
11				- medium grey green
12				- fine-grained
13				- massive
14				- generally 5-10% qb-carbonate veinlets with up to 40% in some sections
15				- 20.1% pyrite
16				by 8.6 rock appeared strongly sheared, indurated
17				- very soft
18				
19				
20				10.0 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 27 19 87 HOLE NO ME-87-33 LOCATION SITE # 70, L31W, 4150N; ELEV. 326m
 GEOLOGIST SHELP DRILLER HOLSALL BIT NO. CB<3 BIT FOOTAGE 15.5-29.5m
 SHIFT HOURS _____ MOVE TO HOLE 15.45 - 15.55
 _____ TO _____ DRILL 15.55 - 16.45
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER 16.45 - 17.00 clean tanks.
 _____ MOVE TO NEXT HOLE _____

DEPTH IN METRES	GRAPHIC LOG INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	0-0.2		ORGANICS
1	0.2-8.6		<u>OTIWAY SEDIMENTS</u> -grey, soft, pure clay
2			
3			
4			
5			
6			
7			
8	8.6-9.0		<u>MATHEWSON TILL</u> -grey fine-med sand -pebbles 90% vol/seds -very cobbly, 10% granitoid
9		01	
10	9.0-11.0	02	<u>BED ROCK</u> (Basalt) -dark green -fine grained -massive → mod. foliated -10-15% qtz-carbonate veinlets -1% fine-grained pyrite and pyrite blebs
11			
12			
13			
14			
15	11.0		E.O.H.
16			
17			
18			
19			
20			

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 28 1987 HOLE NO ME-87-34 LOCATION SITE 43, L3/W, ST50N, ELEV 337m
 GEOLOGIST SHELP DRILLER HALLSALL BIT NO. CB16673 BIT FOOTAGE 295-400
 SHIFT HOURS 7.00-7.30
 TO DRILL 7.30-8.50
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 MOVE TO NEXT HOLE 8.50-9.05

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.2		ORGANICS
1		0.2-1.4		OTIBWAY SEDIMENTS - beige slightly gritty, soft clay
2		1.4-8.8		MATHISON TILL - grey beige - beige fine-med sand. - pebble 90% volc / seds 10% granitoid - very cobbly 7.4-8.0 boulder (massive volc)
3			01	
4			02	
5			03	
6			04	
7			05	
8			06	
9		8.8-10.5		BEDROCK (Interm / massive volc) - med. green grey - fine-grained - massive → weakly foliated - mod. hard - <0.1% very fine-grained pyrite - strongly calcareous - 1-2% of carbonate veinlets 9.3-9.5 appearance of light olive green (saussurite?) rock (oxidized and fractured) by 9.8 deep yellow colour. - strong shearing (slip planes)
10				
11				
12				
13				
14				
15				
16				
17				
18				
19				
20				10.5 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 28 19 87 HOLE NO ME-87-35 LOCATION SITE # 69, 131W, 3450N ; ELEV 325m
 GEOLOGIST SHELTON DRILLER HALSALL BIT NO. CB 68473 BIT FOOTAGE 90.0-53.5
 SHIFT HOURS _____ MOVE TO HOLE 8.50-9.05
 _____ TO _____ DRILL 9.05-10.40
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 10.40-10.50

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.2		ORGANICS
1		0.2-6.0		<u>OSIBWAY SEDIMENTS</u> - beige soft, pure clay by 3.0 grey clay.
2				
3				
4				
5				
6		6.0-11.8		<u>MATHEWSON TILL</u> - grey fine-med sand - pebbles 80% volc/seds 20% granitoid - occasional cobble
7			01	
8			02	
9				
10			03	
11			04	
12		11.8-13.5		<u>BEDROCK</u> (mafic-ultramafic volc) - dark green black - fine grained - massive, appears to be some shearing - mod. calcareous - 20.1% fine grained pyrite - generally 2-3% qtz - carbonate veinlets, with up to 20% - <1% jasper
13			05	
14				
15				
16				
17				
18				
19		13.5		<u>E.O.H.</u>
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 28 1987 HOLE NO ME-87-36 LOCATION SITE # 71, L31W, 2450N; ELEV 325m
 GEOLOGIST SHELP DRILLER HOLSOBY BIT NO. CB6673 BIT FOOTAGE 535-71.5m
 SHIFT HOURS _____ MOVE TO HOLE 10.40-10.50
 _____ TO _____ DRILL 10.50-13.30
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 13.30-13.40

NEW BIT CB6674 18.0 → 20.5
= 2.5m.

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	△△			0-0.2 ORGANICS
1				0.2-6.2 OTIBWAY SEDIMENTS - beige, soft, slightly gritty clay
2				
3				by 2.5 grey, soft, pure clay (slurry)
4				
5				
6				6.2-18.0 MATHIESON TILL
7	△	01		- grey → grey-beige fine-med. sand
8	△	02		- pebbles 50% volc/secs 20% granitoid
9	△	03		- occasional cobble
10	△	04		9.3-9.7 Boulder (granitic)
11	△			
12	△	05		
13	△			
14	△	06		
15	△	07		
16	△	08		17.0-18.0 Cobbley section 100% volc.
17	△			
18	△			18.0-20.5 BEDROCK (mafic volc)
19	△	09		- med-dark green - fine-grained - strongly foliated, possible shearing - 2-3% calc-carbonate veinlets - 20.1% Pyrite (associated with veinlets) - strongly calcareous
20	△			19.3-19.6 fracture filled with sand
				20.5 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 28 19 87 HOLE NO ME-87-37 LOCATION SITE #72, L31W, 1450N; ELEV 325
 GEOLOGIST SHELP DRILLER HALSALL BIT NO. CB68679 BIT FOOTAGE 2.5-6.5
 SHIFT HOURS _____ MOVE TO HOLE 13.30-13.40
 _____ TO _____ DRILL 13.40-15.00
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 15.00-15.10

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	▲ ▲			0-0.2 <u>ORGANICS</u>
1	△			0.2-0.6 <u>OUTWASH SEDIMENTS</u> - beige, soft, slightly silty clay
2	● ●	01		
3	△			0.6-2.7 <u>MATHESON TILL</u> - gray beige-beige fine-med sand - pebbles 70% volc/seeds - cobbly 30% granitoid
4	▨	02		
5				2.7-4.5 <u>BEDROCK</u> (Basalt) - dark green - fine-grained - massive to slightly foliated - <0.1% pyrite (cp?) - <1% ch-carbonate veinlets - med. calcareous - <0.1% jasper
6				
7				
8				
9				
10				
11				
12				4.5 <u>E.O.H.</u>
13				
14				
15				
16				
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 28 1987 HOLE NO ME-87-38 LOCATION SITE # 73, L3111, OLFSON; ELEV 825m.
 GEOLOGIST SHELP DRILLER HOLSALL BIT NO. CB68679 BIT FOOTAGE 4.5-21.5
 SHIFT HOURS _____ MOVE TO HOLE 15.00 - 15.10
 _____ TO _____ DRILL 15.10 - 16.45
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 16.45 - 17.00

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-1.0		ORGANICS
1		1.0-5.8		OUTWASH SEDIMENTS - grey, soft, slightly gritty clay by 3.0 grey and grey-beige varves.
2				
3				
4				
5				
6		5.8-13.5		MATHERSON TILL - grey-grey beige fine-med sand. - pebbles 80% volc./seeds 20% granitoid
7			01	
8			02	
9		7.8-8.4		cobbly section 100% volc
10		8.8-8.9		20% dark grey clay
11		10.4-11.5		cobbly section 100% volc
12		11.5-12.5		boulder (ultramafic)
13		12.6-13.0		boulder (interm/mafic volc)
14		13.0-13.5		cobbly section grey sand matrix (possible fracture in bedrock)
15		13.5-15.0		BEDROCK (Basalt) - dark green - fine-grained - strongly fractured. - 2-3% soft-carbonate veinlets (pink)
16				
17				
18				
19				
20				
		15.0		E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 29 19 87 HOLE NO ME-87-39 LOCATION SITE 45, L28W, ST50N; ELEV 325
 GEOLOGIST SHELP DRILLER HAYBALL BIT NO C88679 BIT FOOTAGE 215-280
 SHIFT HOURS _____ MOVE TO HOLE _____
 _____ TO _____ DRILL 07.30-0910
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER Travel 07.00-07.30
 _____ MOVE TO NEXT HOLE 0910-0930

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.5 ORGANICS
1				0.5-3.5 <u>OUTERWAY SEDIMENTS</u> - beige, soft, slightly silty clay by 2.0 grey, soft, pure clay
2				
3				
4			01	3.5-4.6 <u>MATHESON TILL</u> cobble fill - 70% vol/seds 30% granitoid - minor beige fine-med sand.
5			02A	
6			02B	
7				4.6-7.5 <u>BEDROCK</u> (Basalt) - dark green - fine-grained - 1-2% qb carbonate veinlets
8				5.6-7.5 beige to green - strongly sheared - 30-40% qb-carbonate veinlets
9				
10				
11				
12				
13				7.5 E.O.H.
14				
15				
16				
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 29 19 87 HOLE NO ME-87-40 LOCATION L27W, S165N, ELEV-325 m
 GEOLOGIST SHELP DRILLER HALSALL BIT NO. CBK79 BIT FOOTAGE 290-410
 SHIFT HOURS _____ MOVE TO HOLE 0910-0930
 _____ TO _____ DRILL 0930-10.50
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 10.50-11.00

DEPTH IN METRES	GRAPHIC LOG INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	>>>		0-0.5 ORGANICS
1			0.5-7.0 <u>OUTERWAY SEDIMENTS</u>
2			- grey, soft, pure clay
3			
4			
5			
6			
7	X X		7.0-10.0 <u>MATHESON TILL</u>
8	Δ	01	7.0-7.3 boulder (basalt)
9	Δ	02	7.3-8.8 - grey fine-med sand pebbles 80% vol/seeds 20% granitoid - occasional cobble.
10	Δ	03A	8.8-10.0 cobbly section - 100% vol. - matrix poor.
11		03B	
12		03A	
13			10.0-12.0 <u>BEDROCK</u>
14			10.0-10.8 (Basalt)
15			- dark green - fine grained - strongly foliated → strong shearing (slip planes) - oxidized - 10% qtz - <0.1% sulphides
16			10.8-11.2 (Granitic dyke)
17			- rusty colour - fine-med grained (mafic minerals) - siliceous (hard) - appears to be sheared - <0.1% very fine-grained pyrite
18			
19			
20			11.2-12.0 same as 10.0-10.8.
			12.0 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 29 19 87 HOLE NO ME-87-41 LOCATION L27W, 9+90N; FLEU-325m
 GEOLOGIST SHARP DRILLER HARSALE BIT NO. CB64674 BIT FOOTAGE 910-545
 SHIFT HOURS _____ MOVE TO HOLE 10.50-11.00
 _____ TO _____ DRILL 11.00-13.20
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 13.20-13.30

NEW BIT CB64675 10.3-13.5 = 3.2m

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.5		ORGANICS.
1		0.5-7.5		<u>OTJIBWAY SEDIMENTS</u> - beige soft, slightly gritty clay
2				
3				
4				
5				
6				
7		7.5-11.5		<u>MATHESON TILL</u>
8		7.5-7.7	01	boulder (mafic volc)
9		7.7-8.0	01	cobbly section > 95% volc
10			02	- minor grey fine-med sand
11		8.0-8.2	03	boulder - (mafic volc)
12		8.2-10.1		gray fine-med sand Rubbles 80% volc/salts 20% granitoid - occasional cobble
13		10.1-10.3		boulder (mafic volc)
14		10.3-11.5		Cobbly section (100% volc)
15		11.5-13.5		<u>BEDROCK</u> (Basalt) - dark green - fine-grained - strongly foliated and sheared - < 0.1% pyrite in shears - strongly calcareous - 1-2% soft-carbonate veins
16				
17				
18				
19				
20				
		13.5		E.O.H

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 29 1987 HOLE NO ME-87-43 LOCATION L27W, 1150N; ELEV 325m
 GEOLOGIST SHELD DRILLER HALSALL BIT NO. CB68675 BIT FOOTAGE 2.7-16.2m
 SHIFT HOURS _____ MOVE TO HOLE 14.15-14.25
 _____ TO _____ DRILL 14.25-15.30
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 15.30-16.00

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.5 ORGANICS
1				0.5-4.6 <u>OVERWAY SEDIMENTS</u> - huge soft, slightly gritty clay by 2.0 grey, soft, pure clay
2				
3				
4				
5			01	4.6-6.6 <u>MATHESON TILL</u> - grey fine-med sand
6			02	- rubble 80% vol/seds
7			03	20% granitoid
8				5.6-6.6 cobbly section > 90% vol. matrix poor
9				6.6-9.5 <u>BEDROCK</u> (Basalt)
10				- dark green
11				- fine-grained
12				- rubbly and ashy appearance (strongly sheared)
13				- moderately calcareous
14				- < 1% faspit
15				- 1% gtz-carbonate inclusions
16				
17				
18				
19				
20				8.5 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 29 19 87 HOLE NO ME-87-19 LOCATION SITE 67, 6794W, 6190N, ELEV 325M
 GEOLOGIST SHARP DRILLER HALL BIT NO. C86425 BIT FOOTAGE 6.2-23.7m
 SHIFT HOURS _____ MOVE TO HOLE 15.30-16.00
 _____ TO _____ DRILL 16.00-17.10
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	>>			0-0.2 <u>ORGANICS</u>
1				0.2-4.6 <u>OTIBWAY SEDIMENTS</u> - beige, soft, slightly gritty clay
2				
3				
4				
5	△		01	4.6-6.4 <u>MATHEWSON TILL</u> 4.6-5.0 cobbly section 90% vol/seds 10% granitoid
6	△		02	5.0-5.8 grey/beige - beige fine-med sand pebbles 70% vol/seds 30% granitoid
7	△		03	5.8-6.0 boulder (gabbro)
8				6.4-6.4 cobbly section
9				6.4-7.5 <u>BEDROCK</u> (Coarse-grained mafic volc. or Gabbro) - dark green - med grained - 30-40% gln-carbonated veinlets - very hard - magnetic - coarse grained crystals - <0.1% very fine-grained pyrite
10				
11				
12				
13				
14				
15				7.5 E.O.H.
16				
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 30 19 87 HOLE NO ME-87-45 LOCATION SITE #49, 424W, 5190N, ELEV 325M
 GEOLOGIST SHELPS DRILLER HALSALL BIT NO. CAW675 BIT FOOTAGE 237-320m
 SHIFT HOURS _____ MOVE TO HOLE _____
 _____ TO _____ DRILL 0930-10.45
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER 08.00-09.30 drill maintenance
 _____ MOVE TO NEXT HOLE 10.45-10.55

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.2		ORGANICS
1		0.2-9.5		AIRWAY SEDIMENTS - large soft, pure clay. by 3.0 grey, soft, pure clay
2				
3				
4				
5				
6				
7				
8				
9				
10		9.5-13.0		MATHERSON TILL - grey fine-med sand. - pebble 80% vol lseds 20% granitoid - occasional cobble.
11			01	
12			02	
13		11.5-13.0		cobbly section 100% dolc.
14			04	
15		13.0-14.5		BEDROCK (massive dolc) - dark green. - fine-grained - massive to moderately foliated - 2-3% qb-carbonate veinlets - < 1% disseminated pyrite
16				
17				
18				
19				
20				
		14.5		E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 30 1987 HOLE NO ME-87-96 LOCATION L24W, 4+40 N: ELEV 325m
 GEOLOGIST SHELPS DRILLER HALSALL BIT NO. 0868675 BIT FOOTAGE 38.2-48.2
 SHIFT HOURS MOVE TO HOLE 10.45-10.55
 TO DRILL 10.55-12.00
 TOTAL HOURS MECHANICAL DOWN TIME _____
 DRILLING PROBLEMS _____
 CONTRACT HOURS OTHER _____
 MOVE TO NEXT HOLE 12.00-12.10

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.2		ORGANICS
1		0.2-8.0		OTIBWAY SEDIMENTS - beige, soft, slightly gritty clay
2				
3				
4				30 grey, soft, pure clay
5				
6				
7				
8		8.0-8.4		MATHESON TILL - grey fine med sand
9				- pebbles 90% vol/seds 10% granitoid
10				
11		8.4-10.0		BEDROCK (Mafic volc) - dark green - fine grained - appears massive (strongly fractured) - 5-10% gtz-carbonate veinlets - strongly calcareous.
12				
13				
14				
15				
16				8.4-9.2 1-2% pyrite (arsenopyrite?)
17				9.2-10.0 1% gtz-carbonate veinlets <1% massive pyrite veinlets
18				
19				
20				10.0 E.O.H

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 30 1987 HOLE NO ME-87-47 LOCATION L24W, 3190 N, ELEV. 325 m
 GEOLOGIST SHELP DRILLER HALSALL BIT NO CBG675 BIT FOOTAGE 48.2-51.7
 SHIFT HOURS _____ MOVE TO HOLE 12.00-12.10
 _____ TO _____ DRILL 12.10-12.45
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 12.45-13.30

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANICS
1				0.2-0.8 <u>QUIBWAY SEDIMENTS</u> - huge, soft, slightly gritty clay
2				
3				0.8-2.0 <u>MATHESON TILL</u> - cobbly till. 100% volc - minor grey fine-med sand matrix
4				
5				
6				2.0-3.5 <u>BEDROCK</u> (mafic volc) - dark green - fine-grained - strongly fractured - 1-2% calc-carbonate veinlets - strongly calcareous by 32 - 50% calc-carbonate veinlets - 20.1% fine-grained pyrite
7				
8				
9				
10				35 E.O.H.
11				
12				
13				
14				
15				
16				
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 30 1987 HOLE NO ME-87-48 LOCATION SITE #74, LISW, L100N; ELEV 321M
 GEOLOGIST SHARP DRILLER HALSALL BIT NO CB6617 BIT FOOTAGE 512-752
 SHIFT HOURS _____ MOVE TO HOLE 12.45 - 13.30
 _____ TO _____ DRILL 13.30 - 16.10
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 16.10 - 16.20

Pg 1 of 2

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
	AA			0-0.2 ORGANICS
1				0.2-9.0 <u>OUTWASH SEDIMENTS</u>
2				- beige, soft, slightly gritty clay
3				
4				by 4.0 grey, soft, pure clay
5				
6				
7				
8				
9				9.0-22.0 <u>MATHESON TILL</u>
10			01	- grey-beige fine-med sand.
11			02	- pebbles 70% volc/seds 30% granitoid
12			03	- very cobbly
13			04	- up to 10% fragments containing fuchsite and fine grained pyrite.
14				9.8-15.8 Cobble section
15			05	14.6-14.8 60% fragments containing fuchsite
16				15.5-16.4 boulder (mafic volc)
17			06	16.4-17.5 cobble section
18				17.5-18.1 boulder (mafic volc)
19			07	18.1-19.8 gray fine-med sand, pebbles 70% volc/seds 30% granitoid
20				19.8-20.0 boulder (mafic volc)

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 30 1987 HOLE NO ME-87-48 LOCATION _____
 GEOLOGIST _____ DRILLER _____ BIT NO. _____ BIT FOOTAGE _____
 SHIFT HOURS _____ MOVE TO HOLE _____
 _____ TO _____ DRILL _____
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

Pg 2 of 2

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
21			08	20.0-22.0 cobbly section
22			09	22.0-23.5 <u>BEDROCK</u> - dark green - fine-grained - rubbly appearance (sheared?) - <0.1% very fine-grained pyrite (massive - volc.)
23				
24				
25				
26				23.5 E.O.H.
27				
28				
29				
30				
31				
32				
33				
34				
35				
36				
37				
38				
39				
40				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 30 1987 HOLE NO ME-87-49 LOCATION L18W, 5100N; ELEV 321 m.
 GEOLOGIST SHELZ DRILLER HALSALL BIT NO. CB68676 BIT FOOTAGE 0-17.0
 SHIFT HOURS _____ MOVE TO HOLE 16.10 - 16.20
 _____ TO _____ DRILL 16.20 - 17.10
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

NEW BIT CB68676
 Jan 31/87 Travel 7.00-7.30
 + load fuel
 Drill 7.30-8.30
 move 8.30-8.40

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANICS
1				0.2-9.0 <u>QUIBWAY SEDIMENTS</u>
2				- beige, soft, slightly gritty clay
3				
4				by 3.5 grey, soft, pure clay
5				
6				
7				
8				
9				9.0-15.4 <u>MATHESON TILL</u>
10	Δ		01	- grey fine-med sand.
11	Δ		02	- pebbles 70% vol/seeds 30% grain/rod
12	Δ			- cobbly
13	Δ		03	19.0-15.4 cobbly section
14	Δ		04	
15	Δ		05	15.4-17.0 <u>BEDROCK</u> (Basalt)
16				- dark green
17				- fine-med grained
18				- weakly- to moderately foliated
18				- magnetic
18				- 1% very fine-grained pyrite
18				- strongly calcareous
20				- 2-3% g/carbonate veinlets
				17.0 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 31 19 87 HOLE NO ME 87-50 LOCATION L18W, 31 80N, ELEV 321 m
 GEOLOGIST SHEP DRILLER HALSALL BIT NO. CB68676 BIT FOOTAGE 17.0-21.0m
 SHIFT HOURS _____ MOVE TO HOLE 08.30-08.40
 _____ TO _____ DRILL 08.40-10.10
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 10.10-11.15

*First attempt hit bedrock at 3m, no till.
(L18W, 400N).*

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-1.0 ORGANICS
1				1.0-2.5. <u>OTIWAY SEDIMENTS</u> - grey, soft, pure clay
2				
3				2.5-4.0 <u>BEDROCK</u> (Basalt) - dark green - fine-grained - med - strongly foliated - strongly calcareous - < 1% carbonates - 0.1% pyrite
4				4.0 E.O.H.
5				
6				
7				
8				
9				
10				
11				
12				
13				
14				
15				
16				
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE JAN 31 19 87 HOLE NO ME-87-51 LOCATION L0225W, 1700N, ELEV 322m
 GEOLOGIST SAHEL P DRILLER HALLSALL BIT NO. CBC5676 BIT FOOTAGE 210-46.5m
 SHIFT HOURS _____ MOVE TO HOLE 10.10 - 11.15
 _____ TO _____ DRILL 11.15 - 13.40
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

Pg 1 of 2

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANICS
0				0.2-22.4 OTTAWAY SEDIMENTS
0.2				0.2-2.5 huge, soft, slightly gritty clay
2.5				2.5-8.2 - beige fine-med sand - < 10% pebbles - strongly oxidized
8.2				8.2-8.6 very cobbly section - 100% volcanics - matrix poor
8.6				8.6-18.6 alternating layers of huge fine-med sand, med-coarse sand and pebbles (rounded) 80% volc/secs 20% granitoid (strongly oxidized)
19.0				* Mar 24/87 19.0-22.4 Matheson Till - examined geochems of sample 07, 08, 09
18.6				18.6-19.0 boulder (mafic volc)
19.0				19.0-22.4 cobbly section 80% volc/secs 20% granitoid
19.0				19.0-22.4 possibly Matheson Till

OVERBURDEN DRILLING MANAGEMENT LIMITED
 REVERSE CIRCULATION DRILL HOLE LOG

DATE JAN 31 19 87

HOLE NO ME-87-51 LOCATION _____

SHIFT HOURS
 _____ TO _____

GEOLOGIST _____ DRILLER _____ BIT NO. _____ BIT FOOTAGE _____

TOTAL HOURS _____

MOVE TO HOLE _____
 DRILL _____

CONTRACT HOURS _____

MECHANICAL DOWN TIME _____

DRILLING PROBLEMS _____

OTHER _____

MOVE TO NEXT HOLE _____

Pg 2 of 2

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
21		07		
22		08		
23		09		21.6-22.0 boulder (mafic volc.).
24		10		22.4-24.5 <u>BEDROCK</u> (Diorite) - med. grey - very silicious - massive - generally fine-grained - discrete med-coarse qtz and/or feldspar crystals
25				
26				
27				
28				
29				29.5 E.O.H.
30				
31				
32				
33				
34				
35				
36				
37				
38				
39				
40				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 2 19 87 HOLE NO ME-87-53 LOCATION LKW, 1116 N ; ELEV 322 m
 GEOLOGIST SHEP DRILLER HILSINK BIT NO B68677 BIT FOOTAGE 0-6.7 m
 SHIFT HOURS MOVE TO HOLE 10.50 - 11.00
 _____ TO _____ DRILL 11.00 - 12.00
 TOTAL HOURS MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS OTHER _____
 _____ MOVE TO NEXT HOLE 12.00 - 12.10

NEW BIT B68677

Second attempt to intersect till
 at 0+60N
 - intersected bedrock at 3m, no till

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.2		ORGANIC
1		0.2-5.5		OTIBWAY SEDIMENTS - huge, soft, pure clay (varved).
2				
3				
4				
5		5.5-6.6		BEDROCK (Basalt)
6			01	- dark green - fine-grained - slightly foliated - 57% pyr.^{te} - carbonate veinlets with associated euhedral pyr.^{te}
7				
8				
9				
10				
11				
12				
13				
14				
15				
16				
17				
18				
19				
20				
				6.6 E.O.H

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 2 19 87 HOLE NO ME-87-59 LOCATION L5W, 1150N, ELEV 322m
 GEOLOGIST SHEEP DRILLER HALLALL BIT NO. CR48677 BIT FOOTAGE 6.7-27.7
 SHIFT HOURS MOVE TO HOLE 12.00-12.10
 _____ TO _____ DRILL 12.10-13.20
 TOTAL HOURS MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS OTHER _____
 _____ MOVE TO NEXT HOLE 13.20-13.30

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	AA			0-0.2 ORGANICS
1				0.2-3.0 <u>OTLAWAY SEDIMENTS</u>
2				- grey, pure, soft clay.
3				4.0-8.0 grey silt
4				
5				5.0-12.2 <u>MATHESON TILL</u>
6			01	cobby 70% cobbles 30% granitoid
7			02	- grey fine-med sand
8			03	By 6.2 pebbles >95% ultramafic - grey fine-med sand - cobby
9				
10			04	By 11.0 pebbles 70% cobbles 30% granitoid
11			04	
12				12.2-15.0 <u>BEDROCK (Ultramafic)</u>
13			05	- dark green - fine-grained - very soft, greasy appearance and feel (false) - 1% fine-grained pyrite
14				
15				15.0 E.O.H.
16				
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 2 19 87 HOLE NO ME-87-53 LOCATION L4100 W, 2+25 N, ELEV 318 M
 GEOLOGIST SHELP DRILLER HALLSALL BIT NO CR68627 BIT FOOTAGE 217-292
 SHIFT HOURS _____ MOVE TO HOLE 13.20-13.30
 _____ TO _____ DRILL 13.30-13.45
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 13.45-13.55

DEPTH METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.2 ORGANICS
1				0.2-4.4 <u>QJIBWAY SEDIMENTS</u>
2				0.2- grey soft pure clay
3				2.5-4.4 grey silt
4				
5			01	4.4-5.7 <u>MATHERSON TILL</u>
6				- grey beige fine-med sand
7			02	- pebbles 70% volcseds 30% granitoid
8				- very cobbly
9				5.7-7.5 <u>BEDROCK</u>
10				(ultramafic)
11				- dark green grey
12				- fine-grained
13				- very soft (greasy appearance - calc)
14				- 0.1-0.5% fine-grained pyrite
15				7.5 E.O.H.
16				
17				
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 2 1987 HOLE NO ME-87-56 LOCATION L3100W, 2+75N; ELEV 318m
 GEOLOGIST SHELP DRILLER HABSALL BIT NO. CB66627 BIT FOOTAGE 292-312m
 SHIFT HOURS _____ MOVE TO HOLE 13.45-13.55
 _____ TO _____ DRILL 13.55-14.35
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 14.35-14.45

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	AA	0-0.2		ORGANIC.
1		0.2-10.6		<u>QUIRWAY SEDIMENTS</u> - beige, soft, pure clay. by 2.0 grey, soft, pure clay (slurry)
2				
3				
4				
5				
6				
7				
8				
9		8.6-10.6		grey very fine-fine sand 25% pebbles
10			01	
11		10.6-12.1		<u>BEDROCK</u> (Granodiorite) - dark reddish colour - med-grained - siliceous - massive - 0.1 % disseminated pyrite - mafic minerals comprised 25%
12			02	
13				
14				
15				
16				
17		12.1		E.O.H.
18				
19				
20				

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 2 19 87 HOLE NO ME-87-57 LOCATION L2400 W, 2125 N, ELEV 318m
 GEOLOGIST SHARP DRILLER WALSALL BIT NO. CR08677 BIT FOOTAGE 31.2-50.7
 SHIFT HOURS _____ MOVE TO HOLE 14.35-14.45
 _____ TO _____ DRILL 14.45-15.40
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 15.40-15.50

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.2		ORGANICS
1		0.2-17.0		<u>OUTWASH SEDIMENTS</u> - grey, soft, pure clay
2		3.0-6.8		grey-beige silt to very-fine sand
3		6.8-17.0		grey-beige to beige fine sand.
4		17.0-18.0	01	<u>MATHISON TILL</u> - grey fine-med sand - pebble 80% to vol% seds - very cobbly 20% granitoid
5		18.0-19.5	02	<u>BEDROCK</u> (Basalt) - dark green - fine-grained - foliated - 20-170 pyrite - 2-17% granitic veining
6		19.5		E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 2 19 87 HOLE NO ME-87-58 LOCATION L 1100 W, 2140 N, ELEV 318 m
 GEOLOGIST _____ DRILLER _____ BIT NO. CB6677 BIT FOOTAGE 50.7-67.1
 SHIFT HOURS MOVE TO HOLE 15.40-15.50
 _____ TO _____ DRILL 15.50-16.30
 TOTAL HOURS MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS OTHER _____
 _____ MOVE TO NEXT HOLE _____

DEPTH IN METRES	GRAPHIC LOG INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0			0-0.2 ORGANICS
1			0.2-13.4. <u>OUTWASH SEDIMENTS</u> - grey, soft, pure clay
2			
3			
4			3.5-13.4 beige fine-med. sand.
5			
6			
7			
8			
9			
10			
11			
12			
13			13.4-14.8. <u>MATHERSON TILL</u> - grey fine-med sand. - pebbles 70% volc / seeds 30% granitoid - very cobbly
14		01	
15		02	
16			14.8-16.4. <u>BEDROCK</u> (Ultramafic) - dark grey green. - fine grained - very soft (greasy appearance - calc) - foliated
17			
18			
19			
20			16.4 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 3 19 87 HOLE NO MP-87-59 LOCATION L 1400E 2175N - ELEV 318m
 GEOLOGIST SHZP DRILLER HALSALL BIT NO. CR68677 BIT FOOTAGE 67.1-98.6
 SHIFT HOURS MOVE TO HOLE 07.00 - 08.00
 TO _____ DRILL 08.00 - 09.20
 TOTAL HOURS MECHANICAL DOWN TIME _____
 DRILLING PROBLEMS _____
 CONTRACT HOURS OTHER _____
 MOVE TO NEXT HOLE 09.20 - 09.30

Pg 1 of 2

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	>>>>			0-1.5 ORGANICS.
1	>>>>			
2	>>>>			15-25.2 <u>OTURWAY SEDIMENTS</u>
3	>>>>			-grey, soft, pure clay
4	>>>>			(slurry)
5	>>>>			(varved appearance)
6	>>>>			
7	>>>>			
8	>>>>			
9	>>>>			
10	>>>>			
11	>>>>			
12	>>>>			
13	>>>>			
14	>>>>			
15	>>>>			
16	>>>>			
17	>>>>			
18	>>>>			
19	>>>>			
20	>>>>			

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE _____ 19 _____ HOLE NO ME-87-59 LOCATION _____
 GEOLOGIST _____ DRILLER _____ BIT NO. _____ BIT FOOTAGE _____
 SHIFT HOURS _____ MOVE TO HOLE _____
 _____ TO _____ DRILL _____
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE _____

Pg 2 of 2

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
21				
22				
23				
24				
25				
26				25.2 - 29.0 <u>MATHESON TILL</u>
27			01	25.2 - 25.8 boulder (interm. volc.)
28				25.8 - 29.0 grey huge fine-med sand
29			02	- boulders 90% volc rocks 10% granitoid
30				- very cobbly
31			03	29.0 - 31.5 <u>BEAROCK</u>
32				(Ultramafic)
33				- dark green grey
34				- fine grained
35				- foliated
16				- very soft (greasy appearance - volc)
17				- 10% vesicled material (soft, white, non-calcareous)
18				
19				
20				31.5 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 3 1987 HOLE NO ME-87-60 LOCATION L2100E, 3+50 N, ELEV 318m
 GEOLOGIST SHIELD DRILLER HALSALL BIT NO. CR10077 BIT FOOTAGE 98.6-118.1
 SHIFT HOURS _____ MOVE TO HOLE 0930-0930
 _____ TO _____ DRILL 0930-11.20
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 11.20-11.30

No overburden or bedrock sample.

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0	>>>	0-1.0		ORGANICS
1	>>>	1.0-19.5		STRAWBY SEDIMENTS
2				-grey, soft, pure clay
3				(varved)
4				
5				
6				
7				
8				
9				
10				
11				
12				
13				
14				
15		15.0-19.5		grey beige fine sand
16				
17				
18				
19		19.5		E.O.H.
20				3 attempts, rods plugged and cleared - hole abandoned.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 3 19 87 HOLE NO ME-87-61 LOCATION L3150E, 3150N; ELEV 318 m
 GEOLOGIST SHARP DRILLER HALEALL BIT NO. CB48676 BIT FOOTAGE 11.1-13.6
 SHIFT HOURS _____ MOVE TO HOLE 11.20-11.30
 _____ TO _____ DRILL 11.30-11.50
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 11.50-12.00

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0				0-0.5 ORGANICS
0.5				0.5-11.5 OVERWAY SEDIMENTS
0.5				0.5-1.5 beige, soft, slightly gritty clay
1.5				1.5-3.0 grey soft, pure clay
3.0				3.0-11.5 grey silt and very fine sand
11.5				11.5-13.5 <u>BEDROCK</u> (Ultramafic)
11.5				- dark green grey
11.5				- fine-grained
11.5				- foliated
11.5				- very soft (greasy appearance - talk!)
11.5				- 170 g/g veining (pink)
13.5				13.5 E.O.H.

**OVERBURDEN DRILLING MANAGEMENT LIMITED
REVERSE CIRCULATION DRILL HOLE LOG**

DATE Feb 3 19 87 HOLE NO ME-87-61 LOCATION L3150E, 3150N; ELEV 318m
 GEOLOGIST SHARP DRILLER HALSALL BIT NO CB65676 BIT FOOTAGE 1181-1316
 SHIFT HOURS _____ MOVE TO HOLE 11.20-11.30
 _____ TO _____ DRILL 11.30-11.50
 TOTAL HOURS _____ MECHANICAL DOWN TIME _____
 _____ DRILLING PROBLEMS _____
 CONTRACT HOURS _____ OTHER _____
 _____ MOVE TO NEXT HOLE 11.50-12.00

DEPTH IN METRES	GRAPHIC LOG	INTERVAL	SAMPLE NO.	DESCRIPTIVE LOG
0		0-0.5		ORGANICS
1		0.5-11.5		<u>OVERWAY SEDIMENTS</u>
2		0.5-1.5		beige, soft, slightly gritty clay
3		1.5-3.0		gray soft, pure clay
4		3.0-11.5		gray silt and very fine sand
5				
6				
7				
8				
9				
10			01	
11				
12		11.5-13.5		<u>BEDROCK</u> (Ultramafic)
13			02	- dark green grey - fine grained - foliated - very soft (greasy appearance - take!)
14				- 170 g/c vein (pink)
15				
16				
17		13.5		E.O.H.
18				
19				
20				

APPENDIX B
SAMPLE WEIGHTS - HEAVY MINERAL CIRCUIT

ARME1FEB.WR1

OVERBURDEN DRILLING MANAGEMENT LIMITED

TOTAL # OF SAMPLES IN THIS REPORT = 40

LABORATORY SAMPLE LOG

SAMPLE NO.	WEIGHT (KG.WET)			WEIGHT (GRAMS DRY)				AU	DESCRIPTION						CLASS							
	TABLE SPLIT	+10 CHIPS	TABLE FEED	TABLE CONC	M. I. CONC LIGHTS	CONC. TOTAL	NON MAG		NO. V.G.	CLAST SIZE	%	MATRIX S/U SD				ST CY COLOR						
												V/S	GR	LS	OT		SD	CY				
ME-87																						
04-01	9.4	2.2	7.2	129.5	89.6	39.9	23.5	16.4	1	265	P	90	10	NA	C	U	Y	Y	Y	B	B	TILL
-02	9.6	2.7	6.9	149.0	99.5	49.5	23.3	26.2	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	8.2	1.4	6.8	111.0	69.9	41.1	26.3	14.8	4	115	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	4.7	0.4	4.3	89.8	62.8	27.0	18.1	8.9	3	1065	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-05	7.4	1.0	6.4	130.9	72.2	58.7	35.8	22.9	1	216	P,C	95	5	NA	NA	U	Y	Y	Y	GY	GY	TILL
-06	9.3	2.2	7.1	126.2	79.1	47.1	22.1	25.0	1	46	C,P	95	5	NA	NA	U	Y	Y	Y	GY	GY	TILL
-07	5.4	0.6	4.8	81.4	58.1	23.3	13.3	10.0	0	NA	P,C	95	5	NA	NA	U	Y	Y	Y	GY	GY	TILL
03-01	9.1	1.4	7.7	127.8	85.7	42.1	26.0	16.1	1	147	P	70	30	NA	NA	U	Y	Y	Y	B	B	TILL
-02	9.4	1.4	8.0	119.9	76.4	43.5	24.5	19.0	0	NA	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	9.8	1.7	8.1	134.2	87.7	46.5	22.4	24.1	1	17	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	9.5	1.4	8.1	113.5	67.4	46.1	24.9	21.2	0	NA	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-05	10.0	2.2	7.8	186.8	130.4	56.4	27.9	28.5	3	592	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
02-01	6.8	0.6	6.2	82.1	57.7	24.4	18.0	6.4	0	NA	P	60	40	NA	NA	U	Y	Y	Y	B	B	TILL
-02	7.7	1.4	6.3	95.3	70.2	25.1	16.0	9.1	0	NA	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	8.9	1.7	7.2	155.0	114.9	40.1	26.8	13.3	1	38	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	9.4	1.7	7.7	107.6	71.2	36.4	24.7	11.7	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-05	8.8	1.4	7.4	112.0	76.5	35.5	23.0	12.5	5	87	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-06	9.1	0.7	8.4	134.9	77.2	57.7	34.4	23.3	14	163	P	90	10	NA	NA	U	Y	Y	Y	GG	GG	TILL
01-01	9.0	1.8	7.2	125.7	82.4	43.3	30.9	12.4	1	21	P	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
-02	8.4	1.2	7.2	147.6	98.1	49.5	31.9	17.6	10	231	P	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
19-01	9.0	3.1	5.9	132.7	96.6	36.1	27.4	8.7	1	77	P,C	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	6.9	1.2	5.7	141.6	109.7	31.9	23.9	8.0	1	8	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
17-01	8.6	1.0	7.6	147.0	95.1	51.9	37.5	14.4	20	730	P,C	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
16-01	8.0	1.2	6.8	175.9	129.3	46.6	35.5	11.1	7	82	P,C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
15-01	9.0	1.1	7.9	204.7	149.3	55.4	41.8	13.6	10	386	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
14-01	8.2	0.9	7.3	168.9	111.2	57.7	44.5	13.2	8	438	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	8.9	1.6	7.3	137.7	90.3	47.4	34.1	13.3	15	4930	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
13-01	5.5	0.5	5.0	106.4	79.7	26.7	19.3	7.4	4	183	P	90	10	NA	C	U	Y	Y	Y	GB	GB	TILL
12-01	8.8	1.9	6.9	111.4	77.6	33.8	24.2	9.6	1	62	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
11-01	8.8	1.3	7.5	252.8	220.4	32.4	24.4	8.0	7	5612	P,C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	6.5	0.2	6.3	246.3	208.2	38.1	29.0	9.1	4	144	P,C	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	8.2	1.7	6.5	339.9	304.5	35.4	26.8	8.6	5	220	P,C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	8.1	2.6	5.5	316.8	280.5	36.3	27.3	9.0	1	55	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-05	8.8	2.0	6.8	278.0	245.9	32.1	24.3	7.8	1	389	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
10-01	4.8	1.1	3.7	161.9	140.7	21.2	15.8	5.4	1	134	C	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
09-01	9.1	1.1	8.0	193.0	157.2	35.8	25.5	10.3	8	471	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	9.5	2.1	7.4	192.6	152.1	40.5	27.8	12.7	6	1099	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	8.4	0.9	7.5	169.4	135.4	34.0	23.8	10.2	5	379	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	8.5	1.5	7.0	124.2	86.0	38.2	26.0	12.2	5	421	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
09-05	8.8	1.5	7.3	129.2	87.4	41.8	29.4	12.4	1	13	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL

OVERBURDEN DRILLING MANAGEMENT LIMITED

LABORATORY SAMPLE LOG

12 FEB. WR.
TOTAL # OF SAMPLES IN THIS REPORT = 40

SAMPLE NO.	WEIGHT (KG. WET)			WEIGHT (GRAMS DRY)			AU			DESCRIPTION						CLASS						
	TABLE SPLIT	+10 CHIPS	TABLE FEED	TABLE CONC	M. I. CONC			NO. CALC	NO. V.G.	CLAST			MATRIX			ST	CY	COLOR				
					M.I.	CONC.	NON			SIZE	%	S/U	SD	ST	CY				COLOR			
					LIGHTS	TOTAL	MAG													V/S	GR	LS
ME-87																						
09-06	8.2	1.4	6.8	104.4	63.1	41.3	26.3	15.0	1	237	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-07	8.6	1.0	7.6	126.2	90.0	36.2	25.8	10.4	5	115	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-08	9.0	0.6	8.4	144.3	103.0	41.3	29.4	11.9	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-09	8.8	0.2	8.6	107.0	73.5	33.5	23.5	10.0	1	43	P,C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
01-03	9.1	1.4	7.7	113.4	73.7	39.7	28.0	11.7	1	3	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	6.5	0.2	6.3	100.9	69.4	31.5	22.3	9.2	6	201	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
03-06	9.0	1.5	7.5	128.5	82.4	46.1	30.9	15.2	10	100	P,C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-07	9.5	2.0	7.5	103.8	71.5	32.3	22.4	9.9	6	503	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
05-01	9.3	1.8	7.5	122.9	91.6	31.3	22.5	8.8	5	161	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	8.7	1.6	7.1	82.3	55.8	26.5	18.4	8.1	1	10	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	9.3	0.7	8.6	110.9	72.8	38.1	25.8	12.3	1	621	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	9.6	1.9	7.7	102.0	66.7	35.3	25.0	10.3	6	160	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-05	9.2	0.2	9.0	98.3	66.5	31.8	22.1	9.7	0	NA	P	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
-06	9.3	1.5	7.8	139.6	101.3	38.3	28.0	10.3	10	165	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
06-01	8.1	1.7	6.4	108.6	72.2	36.4	26.7	9.7	4	1853	P,C	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	9.1	2.0	7.1	138.3	92.3	46.0	33.6	12.4	11	421	P,C	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	9.4	1.8	7.6	155.7	106.0	49.7	34.6	15.1	6	67	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	9.0	2.0	7.0	166.9	127.3	39.6	28.2	11.4	4	209	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-05	5.6	1.0	4.6	104.1	78.2	25.9	19.6	6.3	3	97	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-06	8.5	2.0	6.5	168.1	134.7	33.4	25.0	8.4	1	26	P,C	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
07-01	8.7	2.2	6.5	169.8	126.1	43.7	30.1	13.6	4	556	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	9.0	1.4	7.6	148.2	96.0	52.2	34.4	17.8	8	1076	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	9.5	1.8	7.7	173.7	113.0	60.7	38.8	21.9	7	30	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	8.7	1.0	7.7	177.6	101.9	75.7	36.6	39.1	9	594	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
08-01	8.9	2.5	6.4	126.4	77.1	49.3	33.7	15.6	5	1778	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	9.3	2.3	7.0	121.3	74.6	46.7	32.2	14.5	4	336	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	9.3	0.8	8.5	174.0	116.6	57.4	36.4	21.0	5	111	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	8.7	0.4	8.3	140.0	72.6	67.4	50.0	17.4	4	179	P	85	15	NA	NA	U	Y	Y	Y	B	B	TILL
-05	6.1	0.5	5.6	94.6	52.9	41.7	28.3	13.4	1	3	P	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
18-01	8.1	1.2	6.9	109.5	58.9	50.6	33.8	16.8	8	183	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
20-01	8.9	2.6	6.3	151.9	102.5	49.4	33.7	15.7	5	1661	P	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	8.4	1.5	6.9	143.9	87.5	56.4	38.8	17.6	9	420	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
21-01	8.6	0.8	7.8	122.9	73.4	49.5	33.2	16.3	11	210	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	9.1	2.0	7.1	138.7	81.7	57.0	38.3	18.7	6	98	P	80	20	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	9.0	1.7	7.3	105.6	58.2	47.4	29.2	18.2	4	947	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
22-01	8.9	1.9	7.0	112.9	71.9	41.0	29.7	11.3	4	260	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	9.1	1.2	7.9	124.1	73.6	50.5	33.0	17.5	7	166	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
23-01	9.0	1.1	7.9	107.1	70.6	36.5	27.2	9.3	1	589	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
24-01	8.6	1.5	7.1	65.6	37.1	28.5	20.6	7.9	3	75	P	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
25-01	4.9	0.9	4.0	83.9	64.1	19.8	14.7	5.1	1	44	P	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL

OVERBURDEN DRILLING MANAGEMENT LIMITED

ME3FEB.WRT

TOTAL # OF SAMPLES IN THIS REPORT = 40

LABORATORY SAMPLE LOG

SAMPLE NO.	WEIGHT (KG.WET)			WEIGHT (GRAMS DRY)				AU		DESCRIPTION										CLASS		
	=====			=====				=====		=====										=====		
				M. I. CONC						CLAST					MATRIX							
	TABLE	+10	TABLE	TABLE	M.I.	CONC.	NON	NO.	CALC	SIZE	%	S/U	SD	ST	CY	COLOR	SD	CY				
SPLIT	CHIPS	FEED	CONC	LIGHTS	TOTAL	MAG	MAG	V.G.	PPB	V/S	GR	LS	DT									
ME-87																						
25-02	9.2	1.0	8.2	142.9	94.3	48.6	28.7	19.9	1	22	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
26-01	9.5	1.5	8.0	150.4	109.1	41.3	30.0	11.3	1	21	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
27-01	8.9	1.1	7.8	117.0	84.7	32.3	22.2	10.1	1	46	P	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	9.0	1.7	7.3	107.4	70.7	36.7	26.4	10.3	1	57	P,C	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	9.2	1.5	7.7	102.3	70.4	31.9	22.9	9.0	1	16	P	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	9.4	1.8	7.6	131.2	90.7	40.5	28.8	11.7	6	201	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-05	9.2	1.2	8.0	130.4	93.2	37.2	26.9	10.3	7	84	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-06	7.5	1.3	6.2	92.5	65.4	27.1	19.5	7.6	3	204	P	80	20	NA	NA	U	Y	Y	Y	GG	GG	TILL
28-01	9.0	1.4	7.6	89.8	55.1	34.7	25.0	9.7	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	9.0	1.5	7.5	71.4	36.9	34.5	24.2	10.3	1	26	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
29-01	7.6	1.1	6.5	76.5	49.7	26.8	19.9	6.9	0	NA	P	85	15	NA	NA	U	Y	Y	Y	B	B	TILL
-02	8.5	1.0	7.5	83.7	52.9	30.8	21.9	8.9	0	NA	P	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
-03	8.2	0.7	7.5	83.6	47.5	36.1	24.6	11.5	4	168	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-04	8.5	1.1	7.4	71.7	38.2	33.5	24.0	9.5	1	63	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-05	7.7	0.9	6.8	83.9	52.1	31.8	22.6	9.2	8	64	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-06	8.7	1.0	7.7	87.4	50.8	36.6	24.0	12.6	8	76	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-07	5.6	0.5	5.1	79.5	60.6	18.9	12.1	6.8	1	7	C	95	5	NA	NA	U	Y	Y	Y	GB	GB	TILL
30-01	8.0	1.2	6.8	145.1	105.0	40.1	24.9	15.2	6	3887	C	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
-02	7.3	0.8	6.5	108.7	80.4	28.3	18.3	10.0	8	256	C	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
-03	6.6	0.6	6.0	79.2	52.6	26.6	17.1	9.5	0	NA	C	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
-04	6.8	0.5	6.3	87.0	60.9	26.1	16.6	9.5	1	90	P	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
31-01	8.1	1.4	6.7	166.2	126.5	39.7	27.7	12.0	8	161	P,C	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
-02	8.6	1.0	7.6	114.9	78.8	36.1	24.4	11.7	4	200	P,C	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
-03	6.6	0.6	6.0	87.5	61.1	26.4	8.5	17.9	3	4993	P	85	15	NA	NA	U	Y	Y	Y	B	B	TILL
32-01	8.5	0.4	8.1	111.9	76.8	35.1	25.0	10.1	5	315	P	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	5.6	1.0	4.6	67.0	45.6	21.4	15.2	6.2	1	99	P,C	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
33-01	4.9	1.4	3.5	51.6	34.4	17.2	12.3	4.9	0	NA	P,C	90	10	NA	NA	U	Y	Y	Y	GG	GG	TILL
34-01	8.6	1.8	6.8	111.9	77.3	34.6	23.4	11.2	1	211	C	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
-02	8.1	1.4	6.7	114.3	86.2	28.1	19.9	8.2	0	NA	C	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
-03	8.3	1.0	7.3	125.5	94.7	30.8	21.3	9.5	6	207	P,C	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
-04	8.3	1.4	6.9	127.9	93.9	34.0	24.5	9.5	0	NA	P	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
-05	7.6	0.7	6.9	110.7	81.2	29.5	21.1	8.4	0	NA	P	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
35-01	8.1	1.7	6.4	81.7	50.6	31.1	22.5	8.6	0	NA	C	85	15	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	8.5	1.6	6.9	133.6	100.5	33.1	23.6	9.5	4	765	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	8.6	1.6	7.0	166.7	131.3	35.4	25.6	9.8	8	396	C	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
-04	9.0	2.2	6.8	110.6	82.7	27.9	19.5	8.4	0	NA	P,C	85	15	NA	NA	U	Y	Y	Y	B	B	TILL
36-01	8.0	1.3	6.7	116.6	84.2	32.4	22.6	9.8	9	882	P,C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	8.2	1.6	6.6	89.8	52.2	37.6	27.0	10.6	0	NA	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-03	6.8	1.2	5.6	82.2	54.6	27.6	19.1	8.5	1	79	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
36-04	9.0	2.0	7.0	121.5	86.0	35.5	25.0	10.5	7	173	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL

OVERBURDEN DRILLING MANAGEMENT LIMITED

14 FEB. WRI

TOTAL # OF SAMPLES IN THIS REPORT = 59

LABORATORY SAMPLE LOG

SAMPLE NO.	WEIGHT (KG. WET)			WEIGHT (GRAMS DRY)				AU		DESCRIPTION						CLASS						
	TABLE SPLIT	+10 CHIPS	TABLE FEED	TABLE CONC	M. I. LIGHTS	M. I. CONC. TOTAL	NON MAG	NO. MAG	CALC V.G.	PPB	SIZE	%	S/U SD			ST	CY	COLOR				
				M. I. CONC						CLAST			MATRIX									
										V/S GR			LS OT			SD CY						
ME-87																						
36-05	8.7	2.2	6.5	153.0	111.0	42.0	28.4	13.6	8	2187	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-06	8.7	1.8	6.9	165.1	121.5	43.6	29.8	13.8	5	125	P	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
-07	8.6	2.0	6.6	165.5	116.4	49.1	32.4	16.7	8	1425	P,C	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
-08	9.1	0.6	8.5	191.2	131.4	59.8	35.1	24.7	15	230	C	90	10	NA	NA	U	Y	Y	Y	66	66	TILL
37-01	7.5	1.1	6.4	136.7	103.6	33.1	22.3	10.8	7	261	P,C	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
38-01	9.3	2.4	6.9	142.1	106.7	35.4	23.8	11.6	6	114	P,C	90	10	NA	NA	U	Y	Y	Y	66	66	TILL
-02	9.0	1.2	7.8	162.0	122.6	39.4	23.6	15.8	8	1120	P	90	10	NA	NA	U	Y	Y	Y	66	66	TILL
-03	8.8	1.5	7.3	137.5	105.6	31.9	21.3	10.6	0	NA	P	90	10	NA	NA	U	Y	Y	Y	66	66	TILL
-04	7.4	1.4	6.0	128.4	100.2	28.2	15.9	12.3	0	NA	C	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-05	5.3	1.4	3.9	85.6	73.0	12.6	7.2	5.4	0	NA	P,C	99	1	NA	NA	U	Y	Y	Y	66	66	TILL
39-01	7.6	2.1	5.5	111.4	83.6	27.8	18.6	9.2	7	132	P	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
40-01	8.5	2.2	6.3	120.0	93.5	26.5	18.8	7.7	6	106	P,C	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-02	8.3	2.4	5.9	89.8	66.1	23.7	15.8	7.9	4	212	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
41-01	7.7	2.2	5.5	103.9	77.1	26.8	17.8	9.0	9	185	P	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
-02	9.2	3.2	6.0	93.2	66.7	26.5	19.2	7.3	1	19	C	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-03	3.5	0.6	2.9	69.7	57.3	12.4	8.6	3.8	1	3	P	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
42-01	6.7	1.4	5.3	136.8	113.2	23.6	16.4	7.2	6	300	C	85	15	NA	NA	U	Y	Y	Y	B	B	TILL
43-01	8.9	1.3	7.6	102.5	70.0	32.5	22.7	9.8	9	127	C	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
-02	9.0	1.8	7.2	126.5	91.8	34.7	20.8	13.9	1	18	C	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
44-01	8.7	2.8	5.9	110.3	76.3	34.0	15.1	18.9	4	760	P	75	25	NA	NA	U	Y	Y	Y	66	66	TILL
-02	3.7	1.2	2.5	47.2	36.0	11.2	7.8	3.4	1	1212	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
45-01	8.5	1.4	7.1	112.3	86.1	26.2	19.2	7.0	4	50	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-02	8.6	2.3	6.3	105.5	80.2	25.3	14.9	10.4	7	843	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-03	8.3	0.8	7.5	156.2	90.0	66.2	45.5	20.7	6	194	C	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
46-01	5.2	1.4	3.8	87.8	74.1	13.7	10.2	3.5	7	6387	C	90	10	NA	NA	U	Y	Y	Y	66	66	TILL
47-01	7.1	2.2	4.9	132.3	106.5	25.8	15.8	10.0	2	993	C	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
48-01	9.7	2.3	7.4	117.9	76.1	41.8	31.5	10.3	3	446	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-02	9.3	1.9	7.4	146.5	95.2	51.3	36.8	14.5	4	141	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-03	8.7	1.3	7.4	149.1	102.5	46.6	33.9	12.7	4	86	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-04	8.7	1.8	6.9	187.2	144.0	43.2	32.0	11.2	1	32	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-05	8.2	1.2	7.0	141.0	101.8	39.2	26.7	12.5	5	382	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-06	8.6	2.1	6.5	140.0	101.7	38.3	27.3	11.0	7	6431	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
-07	6.2	1.1	5.1	141.1	110.3	30.8	21.6	9.2	0	NA	P	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
-08	8.7	1.6	7.1	143.5	108.8	34.7	20.9	13.8	0	NA	P	80	20	NA	NA	U	Y	Y	Y	66	66	TILL
49-01	5.2	3.2	2.0	66.4	52.2	14.2	10.5	3.7	0	NA	P	70	30	NA	NA	U	Y	Y	Y	66	66	TILL
-02	8.4	2.5	5.9	178.5	133.5	45.0	33.4	11.6	7	1610	P	70	30	NA	NA	U	Y	Y	Y	66	66	TILL
-03	8.4	2.2	6.2	177.0	148.1	28.9	21.4	7.5	1	70	P	70	30	NA	NA	U	Y	Y	Y	66	66	TILL
-04	8.5	1.7	6.8	158.1	119.8	38.3	29.2	9.1	1	1	P	70	30	NA	NA	U	Y	Y	Y	66	66	TILL
51-01	8.3	0.0	8.3	153.7	121.7	32.0	23.9	8.1	0	NA	TR	NA	NA	NA	NA	U	Y	Y	Y	B	B	TILL
-02	8.4	1.0	7.4	153.0	122.4	30.6	22.1	8.5	0	NA	C	80	20	NA	NA	U	Y	Y	Y	B	B	TILL
-03	7.8	1.2	6.6	126.3	99.9	26.4	17.8	8.6	0	NA	C	85	15	NA	NA	U	Y	Y	Y	66	66	TILL
-04	5.6	1.2	4.4	100.9	84.1	16.8	12.1	4.7	3	281	C	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
-05	8.4	0.3	8.1	105.8	76.0	29.8	21.8	8.0	0	NA	P,C	90	10	NA	NA	U	Y	Y	Y	B	B	TILL
-06	7.1	0.0	7.1	66.8	37.5	29.3	21.0	8.3	8	1172	TR	NA	NA	NA	NA	U	Y	Y	Y	B	B	TILL

ME4FEB.WR1

OVERBURDEN DRILLING MANAGEMENT LIMITED

TOTAL # OF SAMPLES IN THIS REPORT = 59

LABORATORY SAMPLE LOG

SAMPLE NO.	WEIGHT (KG.WET)			WEIGHT (GRAMS DRY)				AU	DESCRIPTION	CLASS												
	TABLE SPLIT	+10 CHIPS	TABLE FEED	M. I. CONC							CLAST			MATRIX								
				TABLE CONC	M.I. LIGHTS	CONC. TOTAL	NON MAG				NO. MAG	NO. V.G.	CALC PPB	SIZE	%	S/U	SD	ST	CY	COLOR		
ME-87																						
-07	8.6	1.7	6.9	157.7	96.3	61.4	38.2	23.2	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
-08	5.4	1.3	4.1	88.2	63.0	25.2	18.6	6.6	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
-09	6.8	1.5	5.3	199.1	140.0	59.1	21.5	37.6	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
52-01	8.3	1.4	6.9	66.2	21.1	45.1	22.5	22.6	1	67	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
54-01	9.3	1.3	8.0	203.2	64.1	139.1	63.4	75.7	8	18	P	95	5	NA	NA	U	Y	Y	Y	GN	GB	TILL
-02	8.9	1.3	7.6	318.2	105.7	212.5	90.7	121.8	2	32	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
-03	9.4	2.0	7.4	303.5	102.0	201.5	101.2	100.3	1	0	P	95	5	NA	NA	U	Y	Y	Y	GN	GN	TILL
-04	7.9	1.0	6.9	394.0	96.6	297.4	103.7	193.7	2	10	P	85	15	NA	NA	U	Y	Y	Y	BBN	BBN	TILL
55-01	8.1	1.9	6.2	134.3	89.8	44.5	25.1	19.4	0	NA	P	85	15	NA	NA	U	Y	Y	Y	BBN	BBN	TILL
56-01	4.7	0.5	4.2	71.8	41.4	30.4	22.9	7.5	0	NA	C	99	1	NA	NA	U	Y	Y	Y	GB	GB	TILL
57-01	8.1	1.5	6.6	113.3	57.0	56.3	33.0	23.3	17	6574	P	95	5	NA	NA	U	Y	Y	Y	GG	GG	TILL
58-01	7.8	0.7	7.1	100.7	55.7	45.0	32.5	12.5	8	227	P.C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
59-01	9.1	1.9	7.2	127.1	94.0	33.1	24.8	8.3	0	NA	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	8.5	2.6	5.9	104.4	70.9	33.5	20.6	12.9	0	NA	C	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
61-01	7.8	0.0	7.8	82.4	35.6	46.8	33.3	13.5	0	NA	TR	NA	NA	NA	NA	U	Y	Y	Y	B	B	TILL

APPENDIX C
GOLD GRAIN COUNTS AND CALULATED VISIBLE GOLD ASSAYS

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

armofeb.wrl

TOTAL # OF PANNINGS 19

NUMBER OF GRAINS

SAMPLE #	PANNED	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
				T	P	T	P	T	P	T	P								
ME-87																			
04-01	N	150 X 175	31 C	1											1				
															1	23.5	265		
-02	N	NO VISIBLE GOLD																	
-03	Y	25 X 100	13 C	1											1			EST. 2% PYRITE	
		75 X 75	15 C	1											1				
		75 X 100	18 C	1	1										2				
															4	26.3	115		
-04	Y	25 X 75	10 C			1									1			EST. 1% PYRITE	
		50 X 75	13 C	1											1				
		175 X 300	44 C	1											1				
															3	18.1	1065		
-05	N	125 X 225	34 C	1											1				
															1	35.8	216		
-06	N	75 X 100	18 C	1											1				
															1	22.1	46		
-07	N	NO VISIBLE GOLD																	
03-01	N	125 X 150	27 C	1											1				
															1	26.0	147		
-02	N	NO VISIBLE GOLD																	
-03	N	50 X 75	13 C	1											1				
															1	22.4	17		
-04	N	NO VISIBLE GOLD																	
-05	Y	25 X 75	10 C			1									1			EST. 2% PYRITE	
		100 X 200	29 C	1											1				
		175 X 225	38 C	1											1				
															3	27.9	592		
02-01	N	NO VISIBLE GOLD																	

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

arn01feb.wr1

TOTAL # OF PANNINGS 19

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR		DELICATE		TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
					T	P	T	P	T	P						
ME-87																
-02	N		NO VISIBLE GOLD													
-03	N		75 X 100	18 C	1							1				
												1	26.8	38		
-04	N		NO VISIBLE GOLD													
-05	Y		25 X 25	5 C		1						1				EST. 2% PYRITE
			25 X 50	8 C		1						1				
			25 X 75	10 C				1				1				
			50 X 50	10 C	1							1				
			75 X 125	20 C	1							1				
												5	23.0	87		
-06	Y		25 X 25	5 C		4		2				6				EST. 2% PYRITE
			25 X 50	8 C		2						2				
			50 X 75	13 C	1							1				
			50 X 100	15 C		3						3				
			75 X 125	20 C	1	1						2				
												14	34.4	163		
01-01	N		75 X 75	15 C	1							1				
												1	30.9	21		
-02	Y		25 X 25	5 C		2						2				EST. 0.5% PYRITE
			25 X 75	10 C		2						2				200 GRAINS ARSENOPYRITE (FINE)
			50 X 50	10 C		1						1				
			50 X 75	13 C		1						1				
			75 X 75	15 C		1						1				
			75 X 125	20 C	1							1				
			100 X 125	22 C	2							2				
												10	31.9	231		
19-01	N		100 X 125	22 C	1							1				
												1	27.4	77		
-02	N		50 X 50	10 C	1							1				
												1	23.9	8		

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

arnofeb.wrl

TOTAL # OF PANNINGS 19

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR		DELICATE		TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
					T	P	T	P	T	P						
ME-87																
17-01	Y	25 X	50	8 C					4				4			EST. 1% PYRITE
		25 X	75	10 C	3	1							4			
		50 X	50	10 C	1								1			
		50 X	75	13 C	1	2							3			
		50 X	100	15 C		1							1			
		50 X	150	20 C	1								1			
		75 X	100	18 C	1	1							2			
		75 X	125	20 C	1								1			
		100 X	125	22 C	1								1			
		125 X	225	34 C	1								1			
		150 X	225	36 C	1								1			
													20	37.5	730	
16-01	Y	25 X	25	5 C								1	1			EST. 1% PYRITE
		25 X	50	8 C	1	1							2			
		50 X	50	10 C		1							1			
		50 X	75	13 C	1								1			
		50 X	100	15 C		1							1			
		75 X	125	20 C	1								1			
													7	35.5	82	
15-01	Y	25 X	75	10 C	1								1			EST. 1% PYRITE
		50 X	50	10 C	1								1			
		50 X	75	13 C	2								2			
		50 X	100	15 C		1							1			
		75 X	75	15 C	2								2			
		75 X	125	20 C	1								1			
		100 X	125	22 C		1							1			
		150 X	225	36 C	1								1			
													10	41.8	386	
14-01	Y	25 X	25	5 C					1				1			EST. 1% PYRITE
		25 X	75	10 C		1							1			
		75 X	175	25 C	1								1			
		100 X	100	20 C	1								1			
		100 X	125	22 C	1								1			
		125 X	125	25 C	1								1			
		125 X	175	29 C	1								1			
		150 X	150	29 C	1								1			
													8	44.5	438	
-02	Y	25 X	50	8 C	1	1							2			EST. 1% PYRITE
		25 X	75	10 C	1	1							2			

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

arm01feb.wr1

TOTAL # OF PANNINGS 19

NUMBER OF GRAINS

SAMPLE #	PANNED	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
				T	P	T	P	T	P	T	P								
ME-87		50 X 50	10 C	3											3				
		50 X 125	18 C	3											3				
		75 X 75	15 C	1											1				
		75 X 100	18 C	2		1									3				
		300 X 725	81 C	1											1				
															15	34.1	4930		
13-01	Y	25 X 25	5 C			1									1			EST. 1% PYRITE	
		50 X 75	13 C	1											1				
		75 X 100	18 C	1											1				
		100 X 125	22 C	1											1				
															4	19.3	183		
12-01	N	75 X 125	20 C	1											1				
															1	24.2	62		
11-01	Y	25 X 50	8 C			2									2			EST. 1% PYRITE	
		50 X 50	10 C			1									1				
		50 X 100	15 C			1									1				
		75 X 125	20 C	1											1				
		100 X 175	27 C	1											1				
		375 X 575	77 C	1											1				
															7	24.4	5612		
-02	Y	25 X 25	5 C			1									1			EST. 1% PYRITE	
		50 X 125	18 C			1									1			250 GRAINS ARSENOFYRITE (FINE)	
		75 X 100	18 C	1											1				
		100 X 125	22 C	1											1				
															4	29.0	144		
-03	Y	50 X 50	10 C			3									3			EST. 1% PYRITE	
		100 X 100	20 C	1											1			200 GRAINS ARSENOFYRITE (FINE)	
		125 X 150	27 C	1											1				
															5	26.8	220		
-04	N	75 X 125	20 C	1											1				
															1	27.3	55		
-05	N	125 X 250	36 C			1									1				

GOLD CLASSIFICATION

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VISIBLE GOLD FROM SHAKING TABLE AND PANNING

armofeb.wr1

TOTAL # OF PANNINGS 19

NUMBER OF GRAINS

SAMPLE #	PANNED Y/N	DIAMETER	THICKNESS	ABRADED		IRREGULAR		DELICATE		TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
				T	P	T	P	T	P				
ME-87											1 24.3	389	
10-01	N	100 X	125	22 C		1					1		
											1 15.8	134	
09-01	Y	25 X	25	5 C		1					1		EST. 1% PYRITE
		25 X	50	8 C		2					2		150 GRAINS ARSENOPIRYTE (FINE)
		50 X	50	10 C		1					1		
		75 X	100	18 C		1					1		
		75 X	125	20 C		1					1		
		100 X	150	25 C		1					1		
		150 X	175	31 C		1					1		
											8 25.5	471	
-02	Y	25 X	25	5 C		2					2		EST. 1% PYRITE
		50 X	50	10 C					1		1		
		150 X	225	36 C	2						2		
		150 X	250	38 C	1						1		
											6 27.8	1099	
-03	Y	25 X	50	8 C		1					1		EST. 1% PYRITE
		25 X	75	10 C		1					1		
		75 X	100	18 C		1					1		
		75 X	125	20 C		1					1		
		150 X	175	31 C		1					1		
											5 23.8	379	
-04	Y	25 X	25	5 C		1					1		EST. 1% PYRITE
		25 X	75	10 C		1					1		200 GRAINS ARSENOPIRYTE (FINE)
		75 X	125	20 C		1					1		
		100 X	100	20 C		1					1		
		100 X	250	34 C		1					1		
											5 26.0	421	
09-05	N	50 X	75	13 C		1					1		
											1 29.4	13	

LD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

ME2FEB.WR1

TOTAL # OF PANNINGS 29

NUMBER OF GRAINS

SAMPLE # PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL	NON MAG GMS	CALC V.G. ASSAY FPB	REMARKS
				T	P	T	P	T	P	T	P								
ME-87																			
09-06	N	150 X 175	31 C	1										1					
														1	26.3	237			
-07	Y	25 X 50	8 C			1								1				EST. 1% PYRITE	
		50 X 50	10 C			2								2					
		50 X 125	18 C	1										1					
		75 X 125	20 C	1										1					
														5	25.8	115			
-08	N	NO VISIBLE GOLD																	
-09	N	75 X 100	18 C	1										1					
														1	23.5	43			
01-03	N	25 X 50	8 C	1										1					
														1	28.0	3			
-04	Y	25 X 25	5 C										1	1				EST. 1% PYRITE	
		25 X 50	8 C					1					1	1				400 GRAINS ARSENOPYRITE (FINE)	
		50 X 75	13 C			1							1	1					
		75 X 100	18 C			1							1	1					
		75 X 125	20 C	2									2	2					
													6	22.3	201				
03-06	Y	25 X 25	5 C			2							1	3				EST. 1% PYRITE	
		25 X 50	8 C			3							3	3					
		25 X 100	13 C						1				1	1					
		50 X 75	13 C	1									1	1					
		50 X 125	18 C			1							1	1					
		75 X 100	18 C	1									1	1					
													10	30.9	100				
-07	Y	25 X 50	8 C	1									1	1				EST. 2% PYRITE	
		50 X 50	10 C									1	1	1					
		50 X 100	15 C			1							1	1					
		75 X 100	18 C	1									1	1					
		100 X 175	27 C	1									1	1					
		125 X 175	29 C	1									1	1					
													6	22.4	503				

OLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

MEZ FEB. MR1

TOTAL # OF PANNINGS 29

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR		DELICATE		TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
					T	P	T	P	T	P	T	P				
ME-87																
05-01	Y		25 X 25	5 C								1	1			EST. 1% PYRITE
			25 X 75	10 C					1				1			
			50 X 100	15 C	1	1							2			
			100 X 125	22 C	1								1			
													5	22.5	161	
-02	N		50 X 50	10 C	1								1			
													1	18.4	10	
-03	N		225 X 225	42 C	1								1			
													1	25.8	621	
-04	Y		50 X 75	13 C	2	1	1						4			EST. 1% PYRITE
			75 X 100	18 C	1								1			
			75 X 125	20 C		1							1			
													6	25.0	160	
-05	N		NO VISIBLE GOLD													
-06	Y		25 X 25	5 C		1			3				4			EST. 1% PYRITE
			25 X 50	8 C		1				1			2			
			50 X 75	13 C	1	1							2			
			75 X 125	27 C		1							1			
			75 X 150	22 C	1								1			
													10	28.0	165	
06-01	Y		50 X 100	15 C	1								1			EST. 2% PYRITE
			50 X 125	18 C	1								1			
			75 X 150	22 C	1								1			
			250 X 400	58 C	1								1			
													4	26.7	1853	
-02	Y		25 X 50	8 C		1							1			EST. 2% PYRITE
			50 X 50	10 C	1								1			
			50 X 75	13 C	2								2			
			50 X 100	15 C	1								1			
			75 X 75	15 C	1								1			
			75 X 125	20 C							1		1			
			75 X 200	27 C	1								1			
			100 X 100	20 C	1								1			
			100 X 125	22 C	1								1			

OLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

MEZFEB.WR1

TOTAL # OF PANNINGS 29

NUMBER OF GRAINS

SAMPLE #	PANNED Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
				T	P	T	P	T	P	T	P								
ME-87		100 X 150	25 C	1											1				
															11	33.6	421		
-03	Y	25 X 25	5 C			1									1			EST. 2% PYRITE	
		25 X 50	8 C	1											1				
		25 X 75	10 C					1							1				
		50 X 75	13 C	1											1				
		50 X 125	18 C					1							1				
		75 X 75	15 C	1											1				
															6	34.6	67		
-04	Y	75 X 75	15 C	1											1			EST. 1% PYRITE	
		75 X 100	18 C	1											1				
		75 X 150	22 C	1				1							2				
															4	28.2	209		
-05	Y	25 X 25	5 C			1									1			EST. 1% PYRITE	
		50 X 75	13 C	1											1				
		75 X 125	20 C	1											1				
															3	19.6	97		
-06	N	75 X 75	15 C	1											1				
															1	25.0	26		
07-01	Y	50 X 100	15 C	1											1			EST. 2% PYRITE	
		50 X 150	20 C					1							1				
		75 X 100	18 C	1											1				
		175 X 250	40 C	1											1				
															4	30.1	556		
-02	Y	25 X 75	10 C			1									1			EST. 1% PYRITE	
		50 X 100	15 C			1									1				
		75 X 125	20 C			1									1				
		75 X 175	25 C	1											1				
		100 X 200	29 C			1									1				
		125 X 150	27 C	1											1				
		150 X 225	36 C	1											1				
		175 X 250	40 C	1											1				
															8	34.4	1076		

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

RME2FEB.WR1

NUMBER OF GRAINS

TOTAL # OF PANNINGS 29

SAMPLE #	PANNED Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL GMS	NON MAG	CALC V.G. ASSAY PPB	REMARKS
				T	P	T	P	T	P	T	P								
ME-87 -03	Y	25 X 25	5 C			3								3		EST. 1% PYRITE			
		25 X 50	8 C			1								1					
		25 X 75	10 C			1								1					
		50 X 50	10 C	1										1					
		50 X 100	15 C	1										1					
													7	38.8	30				
-04	Y	25 X 25	5 C			3								3		EST. 1% PYRITE 150 GRAINS ARSENOPIRYTE (FINE)			
		25 X 50	8 C	1										1					
		25 X 100	13 C			1								1					
		50 X 75	13 C					1						1					
		75 X 75	15 C			1								1					
		75 X 125	20 C	1										1					
		225 X 250	44 C	1										1					
													9	36.6	594				
08-01	Y	25 X 75	10 C	1	1									2		EST. 2% PYRITE			
		50 X 75	13 C		1									1					
		125 X 125	25 C	1										1					
		200 X 500	61 C	1										1					
													5	33.7	1778				
-02	Y	25 X 75	10 C			1								1		EST. 1% PYRITE			
		50 X 150	20 C			1								1					
		100 X 150	25 C	1										1					
		150 X 175	31 C	1										1					
													4	32.2	336				
-03	Y	25 X 25	5 C			2								2		EST. 1% PYRITE			
		25 X 50	8 C			1								1					
		75 X 100	18 C	1										1					
		100 X 150	25 C	1										1					
													5	36.4	111				
-04	Y	25 X 75	10 C			1								1		EST. 1% PYRITE			
		50 X 75	13 C	1										1					
		50 X 100	15 C	1										1					
		150 X 200	34 C	1										1					
													4	50.0	179				
-05	N	25 X 50	8 C	1										1					

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

HRME2FEB.WR1

TOTAL # OF PANNINGS 29

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR		DELICATE		TOTAL	NON	CALC V.G.	REMARKS		
					T	P	T	P	T	P	T	P	MAG	ASSAY				
															1	28.3	3	
18-01	Y		25 X 50	8 C	1								1		EST. 2% PYRITE			
			25 X 75	10 C									1					
			50 X 75	13 C										1				
			50 X 100	15 C	1									1				
			75 X 75	15 C	2									2				
			100 X 100	20 C	1									1				
			100 X 125	22 C	1									1				
															8	33.8	183	
20-01	Y		50 X 125	18 C	1	1							2		EST. 2% PYRITE			
			75 X 100	18 C	1								1					
			100 X 125	22 C			1							1				
			225 X 450	59 C	1									1				
															5	33.7	1661	
-02	Y		25 X 25	5 C			2						2		EST. 2% PYRITE			
			50 X 100	15 C	1								1					
			50 X 125	18 C	1									1				
			75 X 100	18 C	3									3				
			100 X 125	22 C	1									1				
			125 X 250	36 C	1									1				
															9	38.8	420	
21-01	Y		25 X 25	5 C			1						1		EST. 1% PYRITE			
			25 X 50	8 C			1						1					
			25 X 75	10 C			1	1					2					
			50 X 50	10 C			1							1				
			50 X 75	13 C	1									1				
			50 X 100	15 C	1	1								2				
			75 X 100	18 C	1									1				
			75 X 125	20 C	1									1				
			75 X 150	22 C	1									1				
															11	33.2	210	
-02	Y		25 X 75	10 C	1		1						2		EST. 1% PYRITE			
			50 X 50	10 C	1	1							2					
			75 X 125	20 C			1							1				
			100 X 100	20 C	1									1				
															6	38.3	98	

OLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

RME2FEB.WR1

TOTAL # OF PANNINGS 29

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS	
					T	P	T	P	T	P	T	P	T	P							
ME-87																					
-03	Y		25 X 75	10 C	1											1			EST. 2% PYRITE		
			75 X 100	18 C	1											1					
			150 X 200	34 C	1											1					
			200 X 275	44 C	1											1					
																4	29.2	947			
22-01	Y		50 X 100	15 C	2											2			EST. 1% PYRITE		
			75 X 125	20 C	1											1					
			150 X 150	29 C	1											1					
																4	29.7	260			
-02	Y		25 X 50	8 C												1			EST. 1% PYRITE		
			25 X 75	10 C												1					
			50 X 50	10 C	1											1					
			50 X 75	13 C	1											1					
			75 X 100	18 C	1											1					
			75 X 125	20 C	1											1					
			100 X 125	22 C												1					
																7	33.0	166			
23-01	N		150 X 300	42 C	1											1					
																1	27.2	589			
24-01	Y		25 X 25	5 C												2			EST. 1% PYRITE		
			100 X 100	20 C	1											1					
																3	20.6	75			
25-01	N		75 X 75	15 C	1											1					
																1	14.7	44			

OLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

ME3FEB.WR1

TOTAL # OF PANNINGS 17

NUMBER OF GRAINS

SAMPLE #	PANNED Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
				T	P	T	P	T	P	T	P								
ME-87																			
25-02	N	50 X 100	15 C	1										1					
														1	28.7	22			
26-01	N	50 X 100	15 C	1										1					
														1	30.0	21			
27-01	N	75 X 100	18 C	1										1					
														1	22.2	46			
-02	N	75 X 125	20 C	1										1					
														1	26.4	57			
-03	N	50 X 75	13 C	1										1					
														1	22.9	16			
-04	Y	50 X 50	10 C	1	1									2				EST. 1% PYRITE	
		50 X 75	13 C	1										1					
		75 X 75	15 C	1										1					
		75 X 125	20 C	1										1					
		100 X 150	25 C	1										1					
														6	28.8	201			
-05	Y	25 X 50	8 C	1	3									4				EST. 1% PYRITE	
		50 X 100	15 C			1								1					
		75 X 75	15 C		2									2					
														7	26.9	84			
-06	Y	25 X 50	8 C		1									1				EST. 1% PYRITE	
		75 X 100	18 C	1										1					
		125 X 125	25 C	1										1					
														3	19.5	204			
28-01	N	NO VISIBLE GOLD																	
-02	N	50 X 100	15 C	1										1					
														1	24.2	26			
29-01	N	NO VISIBLE GOLD																	

OLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

RME3FEB.WR1

TOTAL # OF PANNINGS

17

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR		DELICATE		TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
					T	P	T	P	T	P	T	P				
ME-87																
-02	N		NO VISIBLE GOLD													
-03	Y		25 X 25	5 C									1			EST. 1% PYRITE
			50 X 50	10 C									1			
			50 X 125	18 C									1			
			100 X 150	25 C									1			
													4	24.6	168	
-04	N		75 X 125	20 C									1			
													1	24.0	63	
-05	Y		25 X 25	5 C									3			EST. 1% PYRITE
			25 X 50	8 C									2			
			50 X 50	10 C									1			
			50 X 75	13 C									1			
			75 X 75	15 C									1			
													8	22.6	64	
-06	Y		25 X 25	5 C									4			EST. 1% PYRITE
			25 X 50	8 C									1			
			25 X 100	13 C									1			
			75 X 75	15 C									1			
			50 X 100	15 C									1			
													8	24.0	76	
-07	N		25 X 50	8 C									1			
													1	12.1	7	
30-01	Y		25 X 25	5 C									2			EST. 1% PYRITE
			25 X 50	8 C									1			
			50 X 50	10 C									1			
			250 X 600	71 C									1			
													6	24.9	3887	
-02	Y		25 X 25	5 C									1			EST. 1% PYRITE
			25 X 50	8 C									1			
			50 X 50	10 C									1			
			50 X 75	13 C									1			
			125 X 150	27 C									1			

OLD CLASSIFICATION

SIBBLE GOLD FROM SHAKING TABLE AND PANNING

ME3FEB.WR1

NUMBER OF GRAINS

TOTAL # OF PANNINGS 17

SAMPLE # PANNED

ABRADED		IRREGULAR		DELICATE		TOTAL	NON	CALC V.G.	REMARKS
T	P	T	P	T	P	MAG	ASSAY		
GMS		PPB							

Y/N	DIAMETER	THICKNESS	T	P	T	P	T	P	GMS	PPB	REMARKS
-----	----------	-----------	---	---	---	---	---	---	-----	-----	---------

ME-87

8 18.3 256

-03 N NO VISIBLE GOLD

-04 N 100 X 100 20 C 1

1

1 16.6 90

31-01 Y 25 X 25 5 C 2
 50 X 75 13 C 1 1
 75 X 75 15 C 1
 75 X 100 18 C 1 2

2 EST. 1% PYRITE
 2
 1
 3

8 27.7 161

-02 Y 25 X 25 5 C 2
 75 X 100 18 C 1
 125 X 150 27 C 1

2 EST. 1% PYRITE
 1
 1

4 24.4 200

-03 Y 25 X 25 5 C 1
 100 X 100 20 C 1
 125 X 500 56 C 1

1 EST. 1% PYRITE
 1
 1

3 8.5 4993

32-01 Y 25 X 25 5 C 2
 25 X 50 8 C 1
 100 X 100 20 C 1
 150 X 175 31 C 1

2 EST. 1% PYRITE
 1
 1
 1

5 25.0 315

-02 N 75 X 125 20 C 1

1

1 15.2 99

33-01 N NO VISIBLE GOLD

34-01 N 100 X 200 29 C 1

1

1 23.4 211

-02 N NO VISIBLE GOLD

-03 Y 25 X 25 5 C 1

1

EST. 1% PYRITE

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

18ME3FEB.WR1

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	NUMBER OF GRAINS				NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS			
					ABRADED		IRREGULAR					DELICATE		TOTAL
					T	P	T	P				T	P	
TOTAL # OF PANNINGS					17									
ME-87			50 X 75	13 C	1	1			2					
			75 X 75	15 C	1				1					
			100 X 100	20 C	1		1		2					
									6	21.3	207			
-04	N		NO VISIBLE GOLD											
-05	N		NO VISIBLE GOLD											
35-01	N		NO VISIBLE GOLD											
-02	Y		25 X 25	5 C				1	1		EST. 1% PYRITE			
			75 X 100	18 C	2				2					
			200 X 250	42 C	1				1					
									4	23.6	765			
-03	Y		25 X 25	5 C		1			1		EST. 1% PYRITE			
			25 X 50	8 C	1		1		2					
			50 X 50	10 C	1				1					
			50 X 125	18 C	2				2					
			50 X 300	34 C	1				1					
			250 X 300	50 C	1				1					
									8	25.6	396			
-04	N		NO VISIBLE GOLD											
36-01	Y		25 X 25	5 C		1			1		EST. 1% PYRITE			
			25 X 75	10 C		1			1					
			50 X 75	13 C	1				1					
			75 X 75	15 C	1				1					
			100 X 125	22 C	2				2					
			100 X 150	25 C		1			1					
			125 X 150	27 C	1				1					
			150 X 200	34 C	1				1					
									9	22.6	882			
-02	N		NO VISIBLE GOLD											
-03	N		100 X 100	20 C	1				1					
									1	19.1	79			
36-04	Y		25 X 25	5 C		3			3		EST. 1% PYRITE			

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

ARME3FEB.WR1

NUMBER OF GRAINS

TOTAL # OF PANNINGS 17

SAMPLE #	PANNED Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR		DELICATE		TOTAL MAG GMS	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
				T	P	T	P	T	P						
ME-87		25 X 50	8 C						1			1			
		50 X 100	15 C	2								2			
		75 X 175	25 C		1							1			
												7	25.0	173	

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

ME4FEB.WR1

TOTAL # OF PANNINGS 35

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR		DELICATE		TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
					T	P	T	P	T	P	T	P				
ME-87 36-05	Y		25 X 25	5 C								2	2	28.4	2187	EST. 1% PYRITE
			25 X 50	8 C						1	1					
			50 X 50	10 C					1		1					
			50 X 75	13 C	1						1					
			75 X 75	15 C	1						1					
			175 X 400	52 C	1						1					
			225 X 325	50 C	1						1					
													8	28.4	2187	
-06	Y		25 X 25	5 C								1	1	29.8	125	EST. 1% PYRITE
			25 X 50	8 C						2	2					
			50 X 100	15 C	1						1					
			125 X 125	25 C	1						1					
													5	29.8	125	
-07	Y		25 X 50	8 C								1	1	32.4	1425	EST. 2% PYRITE
			25 X 75	10 C							1					
			75 X 100	18 C	1						1					
			100 X 175	27 C					1		1					
			125 X 150	27 C	2						2					
			150 X 150	29 C	1						1					
			250 X 300	50 C	1						1					
													8	32.4	1425	
-08	Y		25 X 25	5 C								4	4	35.1	230	EST. 2% PYRITE
			25 X 50	8 C							2					
			50 X 50	10 C							2					
			50 X 75	13 C							2					
			50 X 100	15 C	1						1					
			75 X 75	15 C	1						1					
			75 X 100	18 C							1					
			75 X 125	20 C							1					
			100 X 150	25 C	1						1					
													15	35.1	230	
37-01	Y		25 X 25	5 C								3	3	22.3	261	EST. 1% PYRITE
			50 X 75	13 C	1						1					
			75 X 100	18 C	1						1					
			100 X 125	22 C	2						2					
													7	22.3	261	

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

ME4FEB.WR1

TOTAL # OF PANNINGS 35

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR		DELICATE		TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS
					T	P	T	P	T	P						
ME-87 38-01	Y		25 X 25	5 C								2		EST. 2% PYRITE		
			50 X 75	13 C	1						1					
			50 X 100	15 C		1					1					
			75 X 75	15 C		1					1					
			75 X 100	18 C	1						1					
												6	23.8	114		
-02	Y		25 X 25	5 C								3		EST. 2% PYRITE		
			25 X 75	10 C							1					
			50 X 125	18 C	1						1					
			75 X 125	20 C		1					1					
			125 X 175	29 C		1					1					
												8	23.6	1120		
-03	N		NO VISIBLE GOLD													
-04	N		NO VISIBLE GOLD													
-05	N		NO VISIBLE GOLD													
39-01	Y		25 X 25	5 C								2		EST. 2% PYRITE		
			50 X 50	10 C			1				2					
			50 X 75	13 C			1				1					
			75 X 75	15 C	1						1					
			75 X 100	18 C	1						1					
												7	18.6	132		
40-01	Y		25 X 25	5 C								1		EST. 2% PYRITED		
			25 X 75	10 C	1						1					
			50 X 50	10 C	1	1					2					
			50 X 75	13 C	1						1					
			75 X 100	18 C	1						1					
												6	18.8	106		
-02	Y		50 X 50	10 C								1		EST. 3% PYRITE		
			75 X 75	15 C							1					
			75 X 100	18 C	1						1					
			75 X 125	20 C	1						1					
												4	15.8	212		
41-01	Y		25 X 25	5 C							3		EST. 3% PYRITE			

OLD CLASSIFICATION

RESIDUAL GOLD FROM SHAKING TABLE AND PANNING

FORM 4 FEB. 67

TOTAL # OF PANNINGS

35

NUMBER OF GRAINS

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL	NON MAG	GMS	CALC V.G. ASSAY	PPB	REMARKS
					T	P	T	P	T	P	T	P	T	P								
ME-87			25 X 50	8 C			2								2							
			25 X 100	13 C					1						1							
			50 X 50	10 C			1								1							
			50 X 75	13 C			1								1							
			75 X 150	22 C					1						1							
															9	17.8	165					
-02	N		50 X 75	13 C	1										1							
															1	19.2	19					
-03	Y		25 X 25	5 C			1								1						EST. 15% PYRITE	
															1	8.6	2.827489					
42-01	Y		25 X 50	8 C			2								2						EST. 1% PYRITE	
			50 X 50	10 C	1										1							
			50 X 75	13 C	1	1									2							
			125 X 150	27 C	1										1							
															6	16.4	300					
43-01	Y		25 X 25	5 C			2								2						EST. 1% PYRITE	
			25 X 50	8 C			1		1						2							
			50 X 50	10 C	2										2							
			50 X 100	15 C	1										1							
			75 X 75	15 C	1										1							
			75 X 100	18 C			1								1							
															9	22.7	127					
-02	N		50 X 75	13 C	1										1							
															1	20.8	18					
44-01	Y		25 X 100	13 C			1								1						EST. 3% PYRITE	
			50 X 100	15 C	1										1							
			50 X 125	18 C			1								1							
			150 X 225	36 C	1										1							
															4	15.1	760					
-02	N		100 X 275	36 C	1										1							
															1	7.8	1212					

GOLD CLASSIFICATION

SIBLE GOLD FROM SHAKING TABLE AND PANNING

RME4FEB.WR1

NUMBER OF GRAINS

TOTAL # OF PANNINGS

35

ABRADED IRREGULAR DELICATE TOTAL NON

SAMPLE # PANNED

Y/N

DIAMETER

THICKNESS

T

P

T

P

T

P

TOTAL

NON

CALC V.G.

MAG

ASSAY

GMS

PPB

REMARKS

ME-87
45-01

Y

25 X 25
50 X 50
50 X 75

5 C
10 C
13 C

1
1
2

1
1
2

EST. 3% PYRITE

4 19.2 50

-02

Y

25 X 25
50 X 50
50 X 75
100 X 125
125 X 250

5 C
10 C
13 C
22 C
36 C

2
1
1
1
1

1

2
1
2
1
1

EST. 2% PYRITE

7 14.9 843

-03

Y

50 X 75
50 X 100
50 X 175
75 X 125
100 X 175

13 C
15 C
22 C
20 C
27 C

2
1
1
1
1

2
1
1
1
1

EST. 7% PYRITE

6 45.5 194

46-01

Y

25 X 25
25 X 50
50 X 75
75 X 100
100 X 100
325 X 400

5 C
8 C
13 C
18 C
20 C
63 C

1
1
1
1
1
1

1

1
2
1
1
1
1

EST. 10% PYRITE

7 10.2 6387

47-01

Y

125 X 200
175 X 200

31 C
36 C

1
1

1
1

EST. 1% PYRITE

2 15.8 993

48-01

Y

25 X 50
125 X 200
125 X 225

8 C
31 C
34 C

1
1
1

1
1
1

EST. 20% PYRITE

3 31.5 446

-02

Y

25 X 50
100 X 125
100 X 150

8 C
22 C
25 C

2
1
1

2
1
1

EST. 20% PYRITE

4 36.8 141

WORLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

ME4FEB.WR1

TOTAL # OF PANNINGS 35

NUMBER OF GRAINS

SAMPLE #	PANNED Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL GMS	NON MAG GMS	CALC V.G. ASSAY FPB	REMARKS
				T	P	T	P	T	P	T	P								
ME-87																			
-03	Y	25 X 25	5 C											1				EST. 10% PYRITE	
		50 X 75	13 C											1					
		50 X 125	18 C											1					
		100 X 100	20 C											1					
														4	33.9		86		
-04	N	75 X 100	18 C											1					
														1	32.0		32		
-05	Y	50 X 75	13 C											1				EST. 10% PYRITE	
		50 X 100	15 C											1					
		100 X 125	22 C											2					
		100 X 200	29 C											1					
														5	26.7		382		
-06	Y	25 X 25	5 C											1				EST. 5% PYRITE	
		25 X 100	13 C											1					
		50 X 75	13 C											1					
		75 X 100	18 C											1					
		100 X 125	22 C											1					
		200 X 200	38 C											1					
		275 X 750	81 C											1					
														7	27.3		6431		
-07	N	NO VISIBLE GOLD																	
-08	N	NO VISIBLE GOLD																	
49-01	N	NO VISIBLE GOLD																	
-02	Y	25 X 50	8 C											1				EST. 20% PYRITE	
		25 X 75	10 C											1					
		50 X 50	10 C											2					
		75 X 75	15 C											1					
		150 X 300	42 C											1					
		250 X 350	54 C											1					
														7	33.4		1610		
-03	N	75 X 125	20 C											1					
														1	21.4		70		

WORLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

SAMPLE #	PANNED	Y/N	DIAMETER	THICKNESS	NUMBER OF GRAINS				TOTAL	NON MAG GMS	CALC V.G. ASSAY PPB	REMARKS		
					ABRADED		IRREGULAR						DELICATE	
					T	P	T	P					T	P
ME4FEB.WR1					TOTAL # OF PANNINGS				35					
ME-B7														
-04	Y		25 X 25	5 C		1			1			EST. 20% PYRITE		
									1	29.2	1			
51-01	N		NO VISIBLE GOLD											
-02	N		NO VISIBLE GOLD											
-03	N		NO VISIBLE GOLD											
-04	Y		50 X 100	15 C					1	1		EST. 1% PYRITE		
			75 X 75	15 C			1			1				
			100 X 125	22 C	1					1				
									3	12.1	281			
-05	N		NO VISIBLE GOLD											
-06	Y		25 X 25	5 C					1	1		EST. 2% PYRITE		
			25 X 50	8 C		2			1	3				
			50 X 100	15 C					1	1				
			75 X 100	18 C			1			1				
			100 X 150	25 C	1					1				
			125 X 200	100 M	1					1				
									8	21.0	1172			
-07	N		NO VISIBLE GOLD											
-08	N		NO VISIBLE GOLD											
-09	N		NO VISIBLE GOLD											
52-01	N		100 X 100	20 C	1					1				
									1	22.5	67			
54-01	Y		25 X 25	5 C				1	1	2		EST. 30% PYRITE		
			25 X 50	8 C		1		1		2		25% SPECULAR HEMATITE		
			25 X 75	10 C				1		1				
			50 X 50	10 C		2				2				
			50 X 75	13 C				1		1				
									8	63.4	18			
-02	Y		25 X 25	5 C					1	1		EST. 25% PYRITE		

GOLD CLASSIFICATION

VISIBLE GOLD FROM SHAKING TABLE AND PANNING

PERME4FEB.WR1

TOTAL # OF PANNINGS 35

NUMBER OF GRAINS

SAMPLE #	PANNED Y/N	DIAMETER	THICKNESS	ABRADED				IRREGULAR				DELICATE				TOTAL MAG GMS	NON MAG GMS	CALC V.G. ASSAY FPB	REMARKS
				T	P	T	P	T	P	T	P	T	P						
ME-87		75 X 175	25 C											1	1			55% SPECULAR HEMATITE	
														2	90.7	32			
-03	Y	25 X 25	5 C				1								1			EST. 25% PYRITE 60% SPECULAR HEMATITE	
														1	101.2	0			
-04	Y	25 X 25 50 X 125	5 C 18 C				1 1								1 1			EST. 20% PYRITE EST. 25% SPECULAR HEMATITE	
														2	103.7	10			
55-01	N	NO VISIBLE GOLD																	
56-01	N	NO VISIBLE GOLD																	
57-01	Y	25 X 25 25 X 50 25 X 75 50 X 50 50 X 75 75 X 75 75 X 125 300 X 375 400 X 625	5 C 8 C 10 C 10 C 13 C 15 C 20 C 59 C 81 C				1 1 1 1 2 2 2 1 1							3 1	4 2 1 2 2 2 1 1			EST. 5% PYRITE	
														17	33.0	6574			
58-01	Y	25 X 25 25 X 50 75 X 75 75 X 200 100 X 125	5 C 8 C 15 C 27 C 22 C				2 2 1 1			1				1	3 1 2 1 1			EST. 15% PYRITE	
														8	32.5	227			
59-01	N	NO VISIBLE GOLD																	
-02	N	NO VISIBLE GOLD																	
61-01	N	NO VISIBLE GOLD																	

APPENDIX D

BINOCULAR LOGS - BEDROCK CHIP SAMPLES

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 01-05				All pebbles and cobble cuttings including 10% Cobalt sediments					TILL
02-07	medium green	Massive	0.1 to 0.15	Interlocking	35% plagioclase with 40% chloritized pyroxene and 10% chlorite. Trace quartz	10% calcite veinlets Trace dissem. calcite	NIL	5% scattered leucoxene	MAFIC VOLC.
03-08	medium green	Massive	0.1 to 0.2	Interlocking	35% plagioclase with 40% chloritized pyroxene and 10% chlorite. Trace quartz	2% calcite veinlets 10% dissem. calcite	Trace pyrite patches in vein only	3% scattered leucoxene	MAFIC VOLC.
04-08A	Medium green 10% chiopochrome (weathered)	Massive locally sheared	0.1 to 0.15	Interlocking	35% plagioclase with 50% chloritized pyroxene. 5% chlorite. 2% quartz 1% sericite.	3% calcite veinlets 2% dissem. calcite.	Trace pyrite in veins only	2% scattered leucoxene	MAFIC VOLC.
04-08B	Pale buff (bleached)	Massive locally sheared	0.1 to 0.15	Interlocking locally texture masked by bleaching	Mafic minerals completely bleached 35% plagioclase 10% sericite 5% quartz Trace trace kurchite	Carbonatized 15% dissem. calcite 35% dissem. mg. carb. (slow reacting to H ₂ O)	NIL		MAFIC VOLC

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 05-07	Medium to dark green	Massive 10% vein; epidote/ quartz veining.	0.3 to 0.5	Interlocking	30% plagioclase with 40% pyroxene and 15% chlorite 10% epidote veining and 2% in adjacent wall rock. Trace quartz.	1% fracture calcite	Trace dissem. calcite.	2% leucocena Trace hematite and red hematite staining. Mostly in thin fractures and adjacent to epidote vein.	MAFIC VOLC.
06-07	Dark green	Massive 5% quartz/epidote veinlets.	0.1 to 0.2	Interlocking	40% plagioclase with 50% partly chloritized (locally) pyroxene and 5% chlorite.	Trace fracture calcite	NIL	Trace dissem. hematite and magnetite	MAFIC VOLC.
07-05	Dark gray-green	Massive	0.3 to 0.7	Interlocking Ophitic	40% plagioclase with 45% chlorite from pyroxene. 3% epidote 5% chlorite Trace quartz	2% fracture calcite	NIL	5% dissem. magnetite Red hematite staining on plagioclase laths (patchily)	DIABASE
08-06	Medium green	Massive 10% fractures	20.05	Interlocking to aphanitic	Plagioclase and chloritized pyroxene proportion unknown 10% epidotized fracture surfaces.	NIL	NIL		MAFIC VOLC.
09-10	Medium green	Massive 5% vein; qtz/carb. 5% fractures	20.05	Interlocking to aphanitic.	Plagioclase and chloritized pyroxene proportions unknown 5% quartz/carbonate veins (stained pale pink by hematite)	5% calcite in hairline fractures	Trace pyrite in vein	2% hematite in vein and fractures	MAFIC VOLC.

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 10-02	Medium green	Massive	0.1 to 0.2	Interlocking	40% plagioclase with 50% pyroxene (partly chloritized) and 5% chlorite	5% fracture calcite	Trace dissem. pyrite		MAFIC VOLC.
11-06	Medium green	Massive 25% sample shows cooling fractures	<0.1 to glass	Interlocking; locally Hypocrystalline	50% soft pale green chloritized px, 40% plag, 1-2% Qtz, 3% fracture chlorite	5% fracture calcite 3% dissem. calcite.	Trace dissem. pyrite		MAFIC VOLC.
12-02	Dark green	Massive	0.05 to 0.15	Interlocking	35% plagioclase with 55% pyroxene (chloritized) and 5% chlorite. 1% Qtz 1% saussurite.	3% vein calcite Trace dissem calcite locally up to 5%	Trace dissem pyrite		MAFIC VOLC.
13-02	Variocoloured - matrix pale gy-grn. to ochre (weath). Clasts gn(bk) to pink (porphy)	Matrix well foliated; sand grains stretched 2:1	Matrix 0.2-1.0 Clasts 5.0 to cobble size	Gravelly - 40%. Sand matrix of 0.2-1.0 mm, 60%. clasts of 5mm to cobble size (several 1cm cuttings of id. of clasts)	Matrix: 90% pale gn. bleached int-mud. volc.; 5% Qtz, 4% jasper; 1% ol. & gn. ch. clasts: 80% volc. as above; 20% Qtz & id. gn.	Average 3% Fe/Mg carb (occurs only in certain volc. clasts and grains)	Matrix only contains 25% dissem. pyrite		CONGLOMERATE
14-03	Variocoloured. - matrix pale gy-grn. to bleached buff clasts pale gy-grn. (volc.) to pink (green por.)	Matrix sheared & crushed Volc. clasts stretched 5:1 Granitic clasts unformed to brecciated	Matrix 0.2-1.0 but crushed Clasts 5mm to cobbles	Gravelly - 50% pebbles and cobbles, 50% med-coarse sand matrix	Matrix: 60% pale gn. volc., 5% Qtz, 1% jasper, 4% mats. py. also 5-10% ch. chl. & 10% bleached chl. or sec. clasts: 50% pale gn. volc., 50% pink granitic & porphyry	Average 10% Fe/Mg carb; conc. in matrix and selected clasts	Matrix contains 0.1% dissem. py. and py.		CONGLOMERATE

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 15-02	Purple (hematite -stained)	Well foliated, crushed. 20% fill pebble content.	Matrix: 0.5-1.0 Pebbles 0.3-0.5	Crumbly. 90% coarse sand matrix 10% granules + pebbles (any coarser clasts obscured by fill pebble content).	Matrix sand is 80% gn. int-mat. vol. 10-20%. gran. + por. <5%. qtz, 2% Jasper -hematite, also 15% gn. chl. fill pebbles to establish comp.	5% dissem Fe/Mg carb.	Nil (hematized)	2-3% finely dissem spec. hem. (hydro- thermal) imparts purple colour to rock	CONGLOM- ERATE
16-02	Med. gy. -gn to buff yellow	Severely deformed. 30% buff sericite seams (shears) 40% qtz-carb. veins + assoc. silicified wallrock Only 30% host	Host 0.2-0.5	Host is mainly med. sand grains	Host consists of 70% pale gn. volc. lithics, 30-35% qtz sand and 15% gy-gn. matrix chlorite	10% dissem Fe/Mg carb except sericitic sheets are carbonate free	<0.1% irregularly dissem. py.	Sericite seams locally contains trace fuchsite	GRAY- WACKÉ
17-02	Dk. gn.	Moderate foliation superimposed on strong primary fluctal structure. Hyaloelastic - 2% remnant fragments. 5% amygdaloid		Rounded fragments in glass sharded matrix with strong fluctal alignment of shards.	Fragments aphanitic and matrix glassy and hard - can't discern minerals. Texture, structure + colour indicate basaltic composition	5% vein calcite	Tr. dissem cubic py.		MAFIC VOLC. (hyalo- clastic)
18-02A	Green pink to buff	20% fill pebbles, 50% (classified on basis of percentage marked by #02 B below but finer is intrusive and bleached from till boulder)	bleached	f.g. amygdaloidal - texture and structure (classified by bleaching), 30% gran- ulitic (chilled), indicating basalt is wallrock, bleached	Low-top basalt + mineral radiolite as quartz, indicating granodiorite , not contamination	20% dissem. Fe/Mg carb. in basalt, 5% fracture Fe/Mg carb. in granodiorite	Tr. dissem. py. in both basalt + granodiorite	Granodiorite contains 3-5% leucosome after sphere	MAFIC VOLC. + GRANO- DIORITE
-02B	Pale pink	Massive to weakly foliated 2% qtz - carb. veins	Phenos 1.0-3.0 G. mass 0.05-0.2	Porphyritic with equigr. interlocking groundmass	50% white to pink fild. phenos Groundmass: - 60% gray plug - 20% chl. - 30% quartzite	3% vein + fracture Fe/Mg carb.	Tr. dissem. py.		GRANO- DIORITE

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 19-03	Pale pink to buff	Very similar feldspar phenos.	to #18-02B crushed	but all of matrix to buff sericitic	and many of mylonitic.	5% fracture + dissem. Fe/Mg carb	Rare trace pyrite		GRANO-DIORITE
20-03		Sample contains a variety of pebbles \approx 50% are of Timiskaming sediments many phases to be		bedrock.	and cuttings; but of too				TILL
21-04	Buff (bleached) matrix; pink granodiorite clasts	Matrix sheared and crushed; \approx 40% of sample. Clasts fractured and veined with Qtz-carb.	Matrix 0.5-1.0 Clasts	Cravilly with coarse sand matrix and clasts probably to coarse pebble or cobble size (relationships obscured by deformation)	Matrix: Bleached volc. and porphyry sand (indistinguishable) 5-10% Qtz. Sand, tr. jasper, Clasts: Some bleached volc; $>$ 50% granodiorite per. ac #16-1B	10% dissem. + fracture Fe/Mg carb. throughout - 1% chl, 20% sericite / tr. fuchsite	Tr. dissem. cubic py.		CONGLOMERATE
22-03	Dark green	Massive 10% calcite veins 1% Fe-carbonate	0.05 to 0.2	Interlocking	35% plagioclase with 35% chloritized pyroxene and 15% chlorite 2% quartz	< 1% dissem. calcite	Nil	1% scattered kuroxene 1% saussurite along fractures	MAFIC VOLC.
23-02	Medium to Dark green	Massive 2% calcite veins	0.2 to 0.5	Interlocking	40% plagioclase with 40% chloritized pyroxene and 10% chlorite 5% quartz	< 1% dissem. calcite	< 1% dissem. pyrite	2% scattered white kuroxene < 1% dissem. hematite 1% epidote	MAFIC VOLC.

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 24-02	Dark green to black	Massive with 1% calcite vein, rare amygdules, 1% fractures	0.05 to 0.1	Interlocking	40% plagioclase with 45% chloritized pyroxene 1% quartz	< 1% dissem. calcite	< 1% dissem. pyrite	< 1% dissem. hematite with 2-5% dissem. magnetite	MAFIC VOLC.
25-03	Dark green	Massive with fractures	0.3 to 1.2	Interlocking to Diabasic	40% plagioclase with 40% fresh pyroxene and 10% chlorite 1-2% quartz 5% epidote & chlorite vein and fracture	Nil	< 1% dissem. pyrite	1-2% dissem. magnetite 1% leucosene	GABBRO (DIABASE)
26-02	Dark green	Massive	0.2 to 0.8	Interlocking to Diabasic	40% plagioclase with 40% fresh pyroxene and 15% chlorite, 2% rutile-ilmenite 1% epidote 1% quartz	Nil	< 1% dissem. pyrite	1% magnetite < 1% leucosene	GABBRO (DIABASE)
27-07	Variable dk. gn. to buff	Highly dismembered - 20% sericite shears especially in matrix and 15-20% chlorite filled fractures esp. in clasts	Where apparent, sand size is 0.2 - 1.0 mm. Clasts probably to cobble size	Mostly obscured by deformation. Gravely, probably clast-supported, coarse sand matrix	Both sand + clasts are mainly bleached, silicified (?-hard) volc. lithics 25% Qtz in matrix 10% por. gran. clasts 15-20% each secondary hydrothermal chl. and sericite	5% fracture calcite	Tr. pyrite also tr. fuchsite		CONGLOMERATE
28-03	Pale gn. to orange-pink	Moderately fol. and fractured 10% matrix, 90% clasts	Matrix 0.2-1.5 Clasts > 1cm	Clast-supported cobbly gravel with coarse sand matrix	Sand lithics 99% - f.g. volc., 1% jasper; also 5% Qtz grains and 5% chlorite Clasts all f.g. volc.; often stained manganese	2-3% fracture Fe/Mg carb.	Tr. pyrite		CONGLOMERATE

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 29-08	Medium gy.-gn. to brick red	Massive to weakly foliated	Matrix 0.2-0.5 Clasts 2.0-10.0	Matrix-supported gravel -- 70% med. sand matrix, 30% clasts of granule to fine pebble size	Pebbles + 85% of sand are pale gray-green volc. lithics -- lithics are often stained brick red + vesicled granite, 1% gtz sand, 15% chlorite	5% fracture calcite	Nil		CONGLOM- -ERATE
30-05	Bleached pale gray to buff	Completely macro to micro-brecciated and sheared. 10% gtz.-carb. veins	<0.05 where preserved	Equigranular interlocking, whirl preserved (Typical f.g. basalt)	70% sample completely replaced by cherty silica + carbonate Mafic minerals partly to entirely bleached from remaining chips	30% dissem. vein + especially fracture calcite	1% pyrite as breccia infill clusters		MAFIC VOLC.
31-04	Medium green	Massive	<0.1 to 0.1	Interlocking	55% saussuritized plagioclase 40% chloritized pyroxene	5% dissem. calcite	0.1% dissem. pyrite	0.5% dissem. magnetite	MAFIC VOLC.
32-03	Medium green-gray to buff where bleached	Massive 10% gtz.-carb. veins	0.05 to 0.1	Equigranular interlocking	50-60% light green pyroxene 10-15% dk. green chlorite 30% light gray- green plagioclase	10% dissem. vein and fracture calcite	0.5% dissem. pyrite		MAFIC VOLC.
33-02	Medium to dark green	Massive few chloritic shears	0.1 to 0.2	Interlocking	50% partly saussuritized plagioclase 40-45% chloritized pyroxene 5% quartz	3% dissem calcite	Trace dissem. pyrite	Trace dissem. leucorene	MAFIC VOLC.

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 34-06	Medium grey-green to buff where bleached	Massive with occasional amygdules, few chloritic shears 10% carbonate vein and fracture infilling	0.05 to 0.1	Interlocking	40% light green px 10% dk gn. chlrite 50% plagioclase - partly saussuritized	2% dissem. calcite 10% vein and fracture gr. calcite	0.5% dissem. pyrite associated with fractures and amygdules		MAFIC VOLC.
35-05	Dark green.	Massive having 3% calcite/epidote veinlets.	0.1 to 0.2	Interlocking.	40-45% partly saussuritized plagioclase 45-50% chloritized pyroxene.	1% dissem. Calcite	0.1% hairline fractures pyrite infilling and trace dissem. pyrite.	5% very finely dissem. magnetite	MAFIC VOLC.
36-09	Medium green	Sheared 1% quartz (carbonate veins).	<0.1 to 0.1	Interlocking	50-55% partly saussuritized plagioclase 40% chloritized pyroxene	5% fracture and 2% dissem. Calcite	Trace dissem. pyrite		MAFIC VOLC.
37-02	Medium to dark green	Massive	0.1 to 0.2	Interlocking	45% plagioclase 35% pyroxene 10% chloritized pyroxene 5% quartz	3% fracture and 2% dissem. Calcite	Trace dissem. pyrite	0.5% very finely dissem. magnetite	MAFIC VOLC.
38-06	Medium green	Massive	<0.1 to 0.1	Interlocking	25-30% plagioclase partly saussuritized 20% scattered saussurite 20-25% pyroxene partly chloritized 20% chlorite 6% quartz	3% fracture and trace dissem. calcite.	NIL		MAFIC VOLC.

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 39-02A	Medium green	Massive having a few chloritic fractures and shears 5% qtz-carbonate vein	0.05 to 0.1	Equigranular Interlocking	40% light gn px, partly chloritized 10% dark gn chlorite 50% partly saussuritized plagioclase	5% quartz-calcite vein and fracture interting (white spots) 2% dissem. calcite	Nil		MAFIC VOLC.
39-02B	Medium green to yellow-green	Extensive shears upto 15% of sample with 5 to 10% qtz-carbonate veins frequently displaying brecciation	0.05 to 0.1	Interlocking equigranular texture where preserved	Notice minerals partly to entirely bleached pyroxene chlorite, plagioclase chlorite/sericite r. 15% in shears	5 to 50% average 20% Fe Mg carbonate primarily in veins	< 1% dissem. pyrite with trace chalcopyrite associated with carbonate veins	20-30% granular hematite associated with qtz-carbonate veins	MAFIC VOLC.
40-03A	medium to dark green	Massive having a few chloritic shears	0.1 to 0.15	Interlocking	30-35% partly saussuritized plag. 35-40% partly chloritized pyroxene 15-20% chlorite 2% quartz.	4% dissem. and 1% fracture calcite	Trace dissem. pyrite	5% leucocoxene	MAFIC VOLC.
40-03B	Pink speckled black	Weakly foliated with 5% chloritic shears	Ground-mass: 0.1-0.2 Phenos 0.5-1.0	Porphyratic 20% indistinct phenos 80% equigran. interlocking g.mass.	20% pink to gray feld. phenos Groundmass: 60-70% pink to gray feld 20% qtz 15% chloritized bt.	15% dissem. Fe/Mg carb. 3% Fe/Mg carb (± quartz) veins assoc. w. chloritic shears	Tr. py. in host 1% py in 4-8 carb veins		GRAND-DIORITE
41-04	Medium to dark green	Massive 2% qtz carb veinlets sheared: few chloritic shears	0.1	Interlocking	25-30% saussuritized plagioclase. 5% saussurite 30% partly chloritized pyroxene 30% chlorite 2%	5% dissem. and 1% fracture calcite	Trace dissem. pyrite		MAFIC VOLC.

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 42-02	Dark green	Massive 1% quartz/calcite veinlets.	0.1 to 0.2	Interlocking	35-40% plagioclase 25-30% partly chloritized pyroxene 20% chlorite 2% quartz	5-10% (variable) dissem. calcite 1% fracture calcite	NIL	4% leucoxene scattered	MAFIC VOLC.
43-03	Dark green	Massive	0.1 to 0.2	Interlocking	45% saussuritized plagioclase 50% completely chloritized pyroxene 3-5% quartz	1% calcite infilling hairline fractures and trace dissem.	NIL	Trace dissem. hematite and magnetite	MAFIC VOLC.
44-03	Dark green	Massive 5% epidote/calcite veins. locally sheared.	0.1 to 0.15	Interlocking	30% completely saussuritized plagioclase 55% completely chloritized pyroxene 10% epidote	Trace dissem. calcite	Trace dissem. pyrite	0.1% dissem. magnetite	MAFIC VOLC.
45-04	Dark green	Massive with 10% chlorite/calcite shear	0.05 to 0.1	Interlocking	30-35% plagioclase saussuritized. 10% saussurite 25-30% partly chlori- tized pyroxene 25-30% chlorite	Trace dissem. calcite	Trace dissem. pyrite	Trace dissem. magnetite	MAFIC VOLC.
46-02	Medium to dark green	Massive	0.05 to 0.1	Interlocking	15% saussuritized plagioclase 15% scattered saussurite 25-30% chloritized pyroxene 25-30% chlorite	5% dissem. and 2% fracture calcite	0.3% dissem. pyrite	0.1% finely dissem. magnetite	MAFIC VOLC.

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 47-02	Dark green to buff colour where bleached	Massive with micro and macro brecciation, 10% quartz-carbonate stringers 2% amygdules infilled with quartz-carbonate	0.05 to 0.1	Equigranular interlocking	40% fresh pale gn. px., 10% dk. gn. chlorite, 40% light grey-green plagioclase	10% quartz-calcite veins 10% dissem. calcite	Nil	0.05% specular hematite	MAFIC VOLC. (Basalt)
48-09	Dark green	Massive	0.05 to 0.3	Equigranular interlocking	50% fresh pale gn. px., 15-20% dk. gn. chlorite, 30% plag.	1% dissem. calcite	Nil	5-10% fine to coarse dissem. octahedral mag., locally leucosae and ochre oxidation stain	MAFIC VOLC. (Basalt)
49-05	Dark green	Massive	0.1 to 0.2	Equigranular interlocking to sub-diabasic	50-60% fresh pale gn. px., 15% dk. gn. chlorite, 30% plag.	2% disseminated calcite 5% calcite stringers	0.5% dissem. cubic pyrite	5% fine to coarse dissem. octahedral magnetite	MAFIC VOLC. (Basalt)
50-01	Dark green	Massive 1% amygdules, infilled with calcite and quartz	0.05 to 0.2	Equigranular interlocking	30% fresh pale gn. px., 15-20% dark gn. chlorite, 50% light grey plag.	5-10% dissem. calcite	0.05% dissem. cubic pyrite		MAFIC VOLC. (Basalt)
51-10	Medium gy.-pink	Well foliated with 10-20% shear-crush zone	Ground mass: 0.1-0.2 Phenos: 2.0-3.0	Strongly porphyritic 70% phenos, 30% groundmass having equigran. interlocking text	70% pale gray pink plag., phenos. Groundmass: 50% plag., 30% chl, 10% Qtz.	Phenocrysts only contain 20% dissem. Fe/Mg carb	1% coarse py. cube clusters	Groundmass contains 10% spec. hem., 1% coarse oct., 1% coarse int., 1% coarse dissem. cubic py.	SYENITE

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 52-02	Dark green	Massive	0.3	Equigranular, interlocking to sub-diabasic	60% fresh pale gn. px., 15% dk. gn. chlorite, 30% plag. (partly hem. stained red), 1% Qtz	3% dissem. dolomite	0.2% coarse dissem. cubic py.	10% coarse interstitial iron oxides - mainly not, locally hem., leucovene	MAFIC VOLC. (Basalt)
53-01	Dark green	Massive except for weak chloritic shears + 10% associated carb. stringers	Variable 0.025 to 0.1	Equigranular, interlocking	50-60% fresh pale gn. px., 15% dk. gn. chl., 30% plag.	2% dissem. calcite 10% calcite stringers, locally w. red hematite-stained wallrock	0.1% dissem. cubic py. mostly occurring in red wall-rock of calcite str.	3% fine to coarse dissem. octahedral magnetite. Calcite str. locally have 10% dissem. spec. hem.	MAFIC VOLC. (Basalt)
54-05	Mottled dk. gn. and white	Well foliated to sub-schistose	Masked by fabric - probably 0.2-0.3	Masked by fabric	70% pale gn. to white falc., 15% dk. gn. chl.	10% very slow-reacting Fe/Mg carb.	0.05% coarse dissem. cubic py.	7% finely divided to octahedral magnetite	ULTRAMAFIC VOLC. (komatiite)
55-02	Mottled dk. gn., gray-white + buff (weath. carb.)	Well foliated to schistose	Masked by fabric - probably 0.2-0.3	Masked by fabric	50-60% pale gn. to white falc., 10% dk. gn. chlorite	25% very slow reacting Fe/Mg carb.	Trace fine to coarse cubic pyrite	10% finely divided magnetite partly alt. to spec. hematite	ULTRAMAFIC VOLC. (komatiite)
56-03	Brick red	Mod. foliated. 1% komatiite (talc-chl. schist) xenoliths of 1-5 mm.	Ground mass: 0.1-0.15 Phenos	Porphyritic with equigran. interlocking groundmass	2% chloritized bt., phlogopite phenos. 98% Groundmass: 80% red-stained feld., 10% chloritized bt., 5% Qtz	8% dissem. calcite	Tr. dissem. cubic py.	5% specular hem. in groundmass (causes red feld. stain)	SYENITE

SAMPLE NUMBER	COLOUR	STRUCTURE	GRAIN SIZE (mm)	TEXTURE	MINERALOGY				NAME
					Silicates	Carbonates	Sulphides	Other	
ME-87 -57-02	Dk. gn.	Massive with 3% chloritic shears and 1% Qtz-carb. veins.	Phenos 1-5 G. mass 0.1	Spinifex	30% pale gn. to rarely dk. gn. px. phenos. Groundmass: -60% pale gn. px., 20% plag.	5% dissem. calcite 1% veinlet calcite	10% py. in calcite veinlets only	3% dissem. octahedral magnetite	ULTRA- MAFIC VOLC. (pyroxenitic komatiite)
-58-02	Dk. gn.	Well fol. to schistose. 10% Qtz-carb. veins	0.15	Equigran. interlocking obscured by alt. and schistosity	30-40% dk. gn. chl., 30% pale gn. talc, 10% harder plag.	20% dissem. Fe/Mg carb. 5-10% vein calcite	Tr. coarse dissem. cubic py.	0.1% dissem. magnetite Qtz-carb. veins contain 3% spec. hem	ULTRA- MAFIC VOLC. (komatiite)
-59-03	Dk. gn.	Well foliated to schistose	Masked by fabric, probably 0.1-0.3	Masked by fabric	60-70% pale gn. to white talc, 30% dk. gn. chlorite	5-10% dissem. Fe/Mg carb	Tr. dissem. cubic py.	0.1% dissem. magnetite	ULTRA- MAFIC VOLC. (komatiite)
-60									
-61-02	Dk. gn. -black	Well foliated to schistose	0.2 -largely masked by schistosity	Equigranular, interlocking (largely masked by schistosity)	50% dk. gn. chlorite, 20-30% pale gn. to white talc, 10-20% harder plag.	10% dissem. calcite	Tr. dissem. cubic py.	Tr. dissem. magnetite	ULTRAMAFIC VOLC. (komatiite)

APPENDIX E
SWASTIKA BEDROCK ANALYSES



SWASTIKA LABORATORIES LIMITED

P.O. BOX 10, SWASTIKA, ONTARIO P0K 1T0

TELEPHONE: (705) 642-3244

ANALYTICAL CHEMISTS • ASSAYERS • CONSULTANTS

Certificate of Analysis

Certificate No. 65495

Date: February 2, 1987

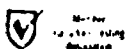
Received Jan. 23, 1987 12 Samples of Bedrock Chips

Submitted by Mary Ellen Resources, Kirkland Lake, Ontario.

SAMPLE NO.	GOLD PPB	SILVER PPM	COPPER PPM
ME-87			
01-05	Nil	Nil	69
02-07	Nil	Nil	31
03-08	Nil	Nil	109
04-08A	Nil/Nil	Nil	139
04-08B	Nil	Nil	29
05-07	Nil	Nil	96
06-07	Nil	Nil	83
07-05	Nil	Nil	139
08-06	Nil	Nil	85
09-10	Nil/Nil	Nil	183
10-02	Nil	Nil	127
11-06	Nil	Nil	95

Per

G. Lebel
G. Lebel - Manager



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SWASTIKA LABORATORIES LIMITED

P.O. BOX 10, SWASTIKA, ONTARIO POK 1T0
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Certificate of Analysis

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Certificate No. 65554 Date: February 10, 1987

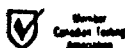
Received Jan. 29, 1987 36 Samples of Bedrock Chips

Submitted by Mary Ellen Resources Ltd., Kirkland Lake, Ontario.

SAMPLE NO.	GOLD PPB	COPPER PPM	SAMPLE NO.	GOLD PPB	COPPER PPM
ME-87			ME-87		
12-02	N11	96	31-04	N11	109
13-02	120/230	14	32-03	N11	60
14-03	N11	77	33-02	N11	88
15-02	N11	5	34-06	N11	85
16-02	N11	49	35-05	N11	90
17-02	N11	17	36-09	N11	107
18-02A	N11	10	37-02	N11	94
18-02B	N11	2	38-06	N11	6
19-03	N11	58	39-02A	N11	9
20-03	N11	39	39-02B	N11	83
21-04	N11	14	40-03A	N11	86
22-03	N11	1	40-03B	N11	107
23-02	N11	32	41-04	N11	98
24-02	N11	111	42-02	N11/N11	103
25-03	N11/N11	142	43-03	N11	102
26-02	N11	119	44-03	N11	105
27-07	N11	85			
28-03	N11	73			
29-08	N11	52			
30-05	N11	96			

JTE: Arsenic results to follow.

Per 
G. Lebel - Manager



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SWASTIKA LABORATORIES LIMITED

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Certificate of Analysis

Certificate No. 65604

Date: Feb. 11, 1987

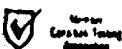
Received Feb. 2 & 3, 1987 16 Samples of bedrock chips

Submitted by Mary Ellen Resources Limited, Kirkland Lake, Ontario

SAMPLE NO.	GOLD PPB	COPPER PPM
<u>ME-87</u>		
45-04	Nil	105
46-02	Nil	156
47-02	Nil	63
48-09	Nil	84
49-05	Nil	146
50-01	Nil	116
51-10	Nil	70
52-02	10/Nil	160
53-01	Nil	77
54-05	Nil	8
55-02	Nil	14
56-02	Nil	38
57-02	Nil	79
58-02	10	56
59-03	Nil	47
61-02	Nil	73

NOTE: Arsenic results to follow.

Per G. Lebel
G. Lebel, Manager



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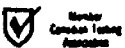
Certificate No. 65495 - A Date: February 17, 1987

Received Jan. 23, 1987 12 Samples of Bedrock Chip

Submitted by Mary Ellen Resources Ltd., Kirkland Lake, Ontario.

SAMPLE NO.	ARSENIC PPM
ME-87	
01-05	<1
02-07	<1
03-08	<1
04-08A	8
04-08B	64
05-07	4
06-07	<1
07-05	<1
08-06	<1
09-10	<1
10-02	<1
11-06	<1

Per G. Lebel
G. Lebel - Manager



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TELEPHONE: (705) 642-3244
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Certificate of Analysis

Certificate No. 65554 - A Date: February 17, 1987

Received Jan. 29, 1987 36 Samples of Bedrock Chip

Submitted by Mary Ellen Resources Ltd., Kirkland Lake, Ontario.

SAMPLE NO.	ARSENIC PPM	SAMPLE NO.	ARSENIC PPM
ME-87		ME-87	
12-02	< 1	31-04	5
13-02	44	32-03	< 1
14-03	51	33-02	< 1
15-02	68	34-06	< 1
16-02	32	35-05	< 1
17-02	16	36-09	< 1
18-02A	33	37-02	< 1
18-02B	22	38-06	< 1
19-03	1	39-02A	< 1
20-03	40	39-02B	25
21-04	61	40-03A	< 1
22-03	< 1	40-03B	64
23-02	4	41-04	66
24-02	39	42-02	< 1
25-03	7	43-03	< 1
26-02	< 1	44-03	14
27-07	20		
28-03	54		
29-08	52		
30-05	61		

Per G. Lebel
G. Lebel - Manager



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Certificate No. 65604 - A

Date: February 20, 1987

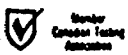
Received Feb. 2 & 3, 1987 16 Samples of Bedrock

Submitted by Mary Ellen Resources Ltd., Kirkland Lake, Ontario.

SAMPLE NO.	ARSENIC PPM
ME-87	
45-04	<1
46-02	28
47-02	<1
48-09	4
49-05	33
50-01	<1
51-10	51
52-02	31
53-01	<1
54-05	54
55-02	73
56-02	27
57-02	56
58-02	15
59-03	49
61-02	<1

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Per G. Lebel
G. Lebel - Manager





ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z9 TELEPHONE (416) 239-3527

Certificate of Analysis

Certificate No. ODM-13/02/6015 Date February 27, 1987

Received Feb. 23/87 59 Samples of Concentrates

Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm
ME-87-44-02	100	1115	54	54
45-01	181	556	80	37
02	605	282	98	26
45-03	306	307	126	28
46-01	1349	2319	90	54
47-01	895	320	<1	31
48-01	1749	778	152	56
02	380	907	167	61
03	378	178	118	34
04	289	511	14	40
05	301	357	99	43
06	7750	296	134	35
07	74	145	<1	40
48-08	313	388	116	46
49-01	265	581	147	112
02	2214	982	54	52
03	577	631	128	44
49-04	539	481	96	44
51-01	95	15	44	19
ME-87-51-02	160	13	68	19

ANALYSED AND CERTIFIED


J. van Engelen Mgr.



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M4Z 2Z2 TEL: (416) 236-1507

Certificate of Analysis

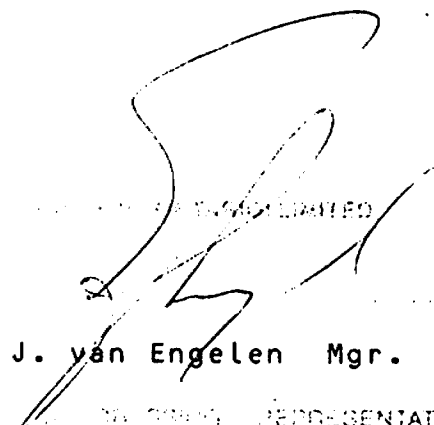
Certificate No. ODM-13/03/6015 Date: February 27, 1987

Received Feb. 23/87 59 Samples of Concentrates

Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb

For Argentex Resources exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm	Co ppm
ME-87-51-03	198	72	34	25	
04	368	96	44	23	
05	113	15	59	14	5
06	2090	53	68	19	148
07	237	23240	97	119	660
08	29	762	303	26	280
51-09	42	564	<1	28	185
52-01	287	194	195	38	275
54-01	252	714	<1	39	1375
02	85	1089	178	29	1228
03	30	701	105	27	974
54-04	14	375	141	37	2040
55-01	125	146	156	25	307
56-01	267	245	101	45	140
57-01	5938	391	118	48	137
58-01	438	280	65	59	374
59-01	58	239	128	42	362
59-02	16	1116	85	88	76
ME-87-61-01	134	40	105	28	426


J. van Engelen Mgr.

APPENDIX F
ASSAYERS HEAVY MINERAL ANALYSES



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33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 · TELEPHONE (416) 239-3527

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Certificate of Analysis

Certificate No. ODM-10/01/5929

Date: February 17, 1987

Received Feb. 4/87 40 Samples of MI - Concentrates

Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm	Co ppm
ME 01-01	554	26	45	14	19
01-02	312	1026	41	67	21
02-01	328	20	16	12	15
02	143	153	51	52	64
03	298	123	63	54	78
04	44	109	57	280	60
05	105	206	26	52	53
02-06	72	125	25	25	64
03-01	306	79	36	25	84
02	183	136	54	38	122
03	420	95	64	28	76
04	56	128	43	24	70
03-05	<5	205	68	30	118
04-01	270	42	15	13	23
02	52	106	28	17	49
03	184	107	53	41	72
04	2166	152	40	25	43
05	449	514	100	90	112
06	154	554	91	258	158
04-07	25000	116	66	25	69

ASSAYERS (ONTARIO) LIMITED

Per J. van Engelen

J. van Engelen Mgr.



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 · TELEPHONE (416) 239-3527

Certificate of Analysis

RECEIVED FEB 27 1987

Certificate No. ODM-11/01/5951

Date: February 20, 1987

Received Feb. 11/87 40 Samples of Heavy Mineral Concentrates

Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm
01-03	129	138	16
01-04	565	614	<1
03-06	230	168	22
03-07	1035	168	29
05-01	374	88	36
05-02	31	106	31
05-03	1068	138	16
05-04	40	206	26
05-05	20	72	21
05-06	275	135	28
06-01	2157	128	23
06-02	856	145	35
06-03	222	98	44
06-04	331	85	25
06-05	41	120	54
06-06	223	76	<1
07-01	914	244	41
07-02	747	96	38
07-03	152	115	<1
07-04	2332	138	8

ASSAYERS (ONTARIO) LIMITED

Per _____

J. van Engelen Mgr.



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 · TELEPHONE (416) 239-3527

Certificate of Analysis

Certificate No. ODM-10/02/5929 Date: February 17, 1987

Received Feb. 4/87 40 Samples of MI-Concentrates

Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm	Co ppm
09-01	456	16	44	9	13
02	2747	59	53	20	45
03	352	92	31	54	72
04	206	96	39	23	51
09-05	198	300	60	54	68
10-01	448	45	30	12	16
11-01	9096	127	64	25	63
02	172	109	37	21	52
03	148	139	38	24	79
04	857	221	72	54	185
11-05	1364	265	41	41	79
12-01	224	183	53	28	98
13-01	92	27	43	13	18
14-01	587	37	49	14	48
14-02	4168	53	51	23	91
15-01	694	91	50	14	56
16-01	275	68	71	25	86
17-01	1159	55	39	25	52
19-01	103	108	46	25	76
19-02	209	64	58	17	47

ASSAYERS (ONTARIO) LIMITED

Per

J. van Engelen Mgr.



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 · TELEPHONE (416) 239-3527

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FEB 23 1987

Certificate of Analysis

Certificate No. ODM-11/02/5051

Date: February 20, 1987

Received Feb. 11/87 40 Samples of Heavy Mineral Concentrates

Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm
08-01	1427	115	<1
08-02	741	206	6
08-03	397	45	<1
08-04	366	20	28
08-05	242	22	<1
09-06	281	3175	<1
09-07	196	387	<1
09-08	195	302	27
09-09	174	365	12
18-01	323	218	27
20-01	2718	125	30
20-02	1025	93	16
21-01	428	78	31
21-02	250	76	17
21-03	2751	254	31
22-01	280	148	<1
22-02	839	150	19
23-01	790	138	31
24-01	270	164	17
25-01	435	125	<1

ASSAYERS (ONTARIO) LIMITED

Per _____

J. van Engelen Mgr.



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 · TELEPHONE (416) 239-3527

Certificate of Analysis

RECEIVED MAR 09 1987

Certificate No. ODM-12/01/5990

Date: February 27, 1987

Received Feb. 18/87 40 Samples of Concentrates

Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm
ME-87-25-02	6491	104	21	65
26-01	208	109	15	87
27-01	159	159	8	52
02	370	100	2	52
03	280	79	28	101
04	360	69	34	65
05	178	47	43	56
27-06	100	195	25	48
28-01	219	95	<1	50
28-02	110	99	22	304
29-01	487	35	49	44
02	927	104	101	53
03	318	77	<1	44
04	136	93	19	41
05	290	122	6	150
06	116	69	31	36
29-07	461	263	109	63
30-01	6507	57	23	32
02	986	29	30	32
ME-87-30-03	139	122	<1	44

ASSAYERS (ONTARIO) LIMITED

Per J. van Engelen

J. van Engelen Mgr.



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 · TELEPHONE (416) 239-3527

Certificate of Analysis

Certificate No. ODM-12/02/5990 Date: February 27, 1987

Received Feb. 18/87 40 Samples of Concentrates

Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. E. comb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm
ME-87-30-04	317	69	39	55
31-01	506	75	116	53
02	369	46	122	44
31-03	23	57	95	59
32-01	819	55	126	53
32-02	594	196	205	71
33-01	1002	89	339	66
34-01	322	98	142	118
02	117	97	5360	204
03	770	111	220	113
04	236	38	155	76
34-05	376	38	262	100
35-01	154	33	199	81
02	281	31	337	80
03	181	32	75	66
35-04	219	28	18	37
36-01	1360	21	42	60
02	367	27	<1	58
03	542	30	78	76
ME-87-36-04	274	28	77	74

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J. van Engelen Mgr.



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 · TELEPHONE (416) 239-3527

Certificate of Analysis

Certificate No. ODM-12/02/5990

Date: February 27, 1987

Received Feb. 18/87 40 Samples of Concentrates

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For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm
ME-87-30-04	317	69	39	55
31-01	506	75	116	53
02	369	46	122	44
31-03	23	57	95	59
32-01	819	55	126	53
32-02	594	196	205	71
33-01	1002	89	339	66
34-01	322	98	142	118
02	117	97	5360	204
03	770	111	220	113
04	236	38	155	76
34-05	376	38	262	100
35-01	154	33	199	81
02	281	31	337	80
03	181	32	75	66
35-04	219	28	18	37
36-01	1360	21	42	60
02	367	27	<1	58
03	542	30	78	76
ME-87-36-04	274	28	77	74

ASSAYERS (ONTARIO) LIMITED

Per _____

J. van Engelen Mgr.



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 - TELEPHONE (416) 239-3527

RECEIVED MAR 16 1987

Certificate of Analysis

Certificate No. ODM-13/01/6015 Date: February 27, 1987
Received Feb. 23/87 59 Samples of Concentrates
Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm
ME-87-36-05	858	90	59	28
06	76	56	75	40
07	1028	73	37	59
36-08	306	216	<1	36
37-01	258	41	38	21
38-01	224	197	64	60
02	2086	211	<1	44
03	33	153	78	29
04	<5	246	14	34
38-05	<5	88	38	20
39-01	257	931	18	31
40-01	158	470	119	52
40-02	868	1208	7	48
41-01	340	595	68	54
02	623	660	86	49
41-03	126	1136	165	58
42-01	320	126	59	37
43-01	326	297	181	71
43-02	39	772	164	74
ME-87-44-01	1908	97	59	40


J. van Engelen Mgr.

APPENDIX G

HEAVY MINERAL GOLD ANOMALY THEORY

**(Authors, figures and tables referenced in this
appendix are located in the main body of the report)**

1.

Regional Gold Background

Most gold occurrences in the Abitibi belt are of the free gold type. Even in Casa-Berardi or Hemlo-type deposits having a high pyrite/arsenopyrite content, most of the gold is free although very fine grained (50 microns). Thus, all tills over the Abitibi belt contain scattered free gold particles. Due to the nugget effect -- the chance occurrence of a coarse gold particle in a given sample -- the gold backgrounds of small till samples collected at the same site will vary by several orders of magnitude.

The nugget effect can be overcome if a sample of sufficient size is collected and all of the gold is concentrated into a small heavy mineral fraction that is then analyzed in its entirety (Clifton et. al., 1967). We have found that at least 50 kg of till would be needed to overcome the nugget effect. However, it is impractical to collect, process or analyze samples of this size. We have standardized to 7-9 kg samples because reverse circulation drills deliver this quantity of material during one metre of advance.

Rather than trying to eliminate the nugget effect, we have developed procedures for recognizing and discounting anomalies that are caused by it. Specifically we measure the dimensions of all gold grains sighted on the table or recovered by panning and use these dimensions to calculate the expected contribution of each gold grain to the concentrate assay. In this way, the cause of each high assay is identified and nugget anomalies are screened out.

Most gold particles occur as thin flakes and it is difficult to position these flakes on edge to measure their thickness. However, we have found that each flake can be treated as a disc in which the thickness is a function of the diameter. For flakes of less than 1000 microns diameter, this relationship is expressed by the following equation:

$$t = 0.2d - 0.01 \frac{(d-100)}{100} d$$

Thus, by simply measuring the diameters of the gold flakes that separate from the samples during tabling, it is possible to calculate the relative volume of gold in a given flake and from this relative volume to calculate the geochemical assay that the flake would produce in a sample of specific size. Clifton (1967) showed that a 100-micron flake will produce a value of approximately 100 ppb in a 15-gram sample. Conveniently, the analyzed 3/4 concentrates of reverse circulation samples also weigh about 15 grams. The range of assays produced in a concentrate of this size by a single gold flake of varying size is shown in Table 5.

It is apparent from the figures in Table 5 that till concentrates that contain no free gold will assay less than 10 ppb provided occluded gold is also absent. Concentrates containing a single gold particle will assay from 10 ppb to more than 55,000 ppb depending on the size of the gold particle. Thus the normal background for till concentrates ranges from less than 10 ppb to more than 55,000 ppb.

We have found that fewer than 30 percent of till concentrates from the Abitibi region yield gold assays lower than 10 ppb. Most samples give assays of 20 to 500 ppb, because they contain one to five gold particles in the 25 to 150 micron range. Erratic clustering of these fine grains occasionally results in an assay over 1000 ppb. Another five to fifteen percent of samples contain a coarser gold grain that produces an assay over 1000 ppb. Occluded gold is rarely present.

2. Gold and Base Metal Anomaly Threshold Levels

Gray (1983) observed that heavy mineral gold assays in a number of dispersion trains tested by Asarco were 3000 ppb or higher. We have arrived at the same 3000 ppb threshold figure in a different manner. As early as 1976, we recognized that the grade of our concentrates within 1 km of source on base metal and uranium dispersion trains was similar to the grade of the source provided the source was of normal width (5 to 10 metres) and was oriented perpendicular to the direction of glacial ice advance. We have since proved that the same relationship applies to gold dispersion trains. Thus, assuming that gold mineralization must grade a

minimum of 3 g/tonne (3000 ppb) to be significant, the anomaly threshold level in our concentrates is 3000 ppb.

It is not uncommon for gold deposits in the Abitibi belt to have a subcropping strike length of only 100 metres. Most of these deposits strike sub-parallel to bedrock stratigraphy and sub-perpendicular to glaciation. Using the 3000 ppb anomaly threshold level, a cross-ice reverse circulation drill hole separation of 100 metres would be needed to detect the deposits. However, most of the deposits occur in anomalous horizons that are much larger than the deposits themselves. If a low anomaly threshold is used and careful gold grain counts are made, the anomalous horizons can be detected with confidence using a 300-400 metre hole separation. This greatly reduces exploration costs. We therefore consider any gold values over 1000 ppb to be potentially anomalous.

3. Stratigraphic Properties of a Dispersion Train

Glacial processes are systematic and heavy mineral dispersion trains in tills have specific configurations (Averill, 1978). For example, dispersed material tends to be sheeted progressively upward in the ice with increasing distance from source, causing the trains to rise in the till and thicken down-ice. Lateral spreading, in contrast, is minimal and most trains are tapered ribbons rather than fans.

ODM has traced nine gold dispersion trains (Table 4) and several base metal and uranium trains to source on both new discoveries and known deposits. These trains have had the following properties:

1. At a specific distance from source, the mineralization was confined to a specific level within a specific till unit.
2. The train was at least two samples (2-3 m) thick unless:
 - (a) The host till was very thin;
 - or (b) The train was intersected within 100 m of source.

3. The width of the train was not more than twice the cross-ice length of the source mineralization.
4. The maximum length of the train for deposits oriented perpendicular to glaciation was 1 km (gold) to 5 km (base metals/uranium).

4. Properties of a Visible Gold Dispersion Train

Five to fifteen percent of background till samples over the Abitibi belt produce heavy mineral gold anomalies higher than our 1000 ppb threshold due to the nugget effect. For the reverse circulation/heavy mineral method to be effective, significant free gold dispersion trains, which are relatively rare, must be differentiated with confidence from the numerous nugget anomalies. This is done on the basis of the gold grain counts rather than the assays. We have found that the gold particles in significant dispersion trains have the following properties:

1. At least 10 gold particles are present per 7 kg of till matrix.
2. The gold particles are of a common size, reflecting the size of crystallization at source.
3. The gold particles are of a common shape, reflecting a common distance of transport from source.
4. Since most gold dispersion trains are traceable for less than one km (Table 4) and gold particles become abraded after one km of ice transport (Fig. 6), the shape of the gold particles is usually irregular or delicate.

Background nugget anomalies, unlike dispersion trains, do not normally repeat vertically or horizontally in the section, although with 10 to 15 percent of samples containing anomalies of this type, chance repetition does occur. Another property common to some gold dispersion trains is the presence of pathfinder minerals

because many gold deposits are polymetallic. Even deposits that are considered to be strictly free gold occurrences often have alteration halos containing sufficient pyrite, arsenopyrite, galena, chalcopyrite or molybdenite for a pathfinder association to be evident in the dispersion train. Nugget anomalies have no pathfinder association.

5. **Properties of an Occluded Gold Dispersion Train**

We have encountered only one occluded gold dispersion train among nine gold trains tested. In one other train, the gold was very fine and more was recovered as composite gold/sulphide grains than as free grains.

In occluded gold trains it is not possible to use gold particle shape to predict distance to source. The distance must be gauged from the vertical positions of the anomaly in the host till and of the till in the stratigraphic succession. In several other respects, however, occluded gold dispersion trains are easier to trace than free gold dispersion trains, especially if the gold is occluded in sulphide minerals. The following specific advantages are cited:

1. A pathfinder mineral association is generally present.
2. The pathfinder minerals often occur in sufficient concentrations that they can be seen in pebbles as well as in the heavy mineral fraction, and the host rock can therefore be determined.
3. The source mineralization is often conductive and can be located by geophysical methods.
4. Gold/pathfinder metal ratios in the concentrates are relatively constant, and any interference from background nuggets is readily recognized.
5. The dispersion trains are longer and more uniform than free gold trains.

Some of these advantages apply only to unoxidized till samples from drill holes. Occluded gold is chemically reconstituted into the clay fraction if the host sulphides are destroyed by oxidation. Thus, in surface pit sampling programs, heavy mineral analysis will detect only the visible gold. Conventional geochemical analysis should be used if occluded gold targets are expected.

ME4FEB.WR1

OVERBURDEN DRILLING MANAGEMENT LIMITED

TOTAL # OF SAMPLES IN THIS REPORT = 59

LABORATORY SAMPLE LOG

SAMPLE NO.	WEIGHT (KG.WET)			WEIGHT (GRAMS DRY)				AU		DESCRIPTION							CLASS					
	TABLE SPLIT	+10 CHIPS	TABLE FEED	M. I. CONC				NO. V.G.	CALC PFB	CLAST				MATRIX			ST CY COLOR SD CY					
				TABLE CONC	M.I. LIGHTS	CONC. TOTAL	NON MAG			NO. MAG	SIZE	%	S/U	SD	ST	CY			COLOR			
																				LS	OT	SD
ME-87																						
-07	8.6	1.7	6.9	157.7	96.3	61.4	38.2	23.2	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
-08	5.4	1.3	4.1	88.2	63.0	25.2	18.6	6.6	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
-09	6.8	1.5	5.3	199.1	140.0	59.1	21.5	37.6	0	NA	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
52-01	8.3	1.4	6.9	66.2	21.1	45.1	22.5	22.6	1	67	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
54-01	9.3	1.3	8.0	203.2	64.1	139.1	63.4	75.7	8	18	P	95	5	NA	NA	U	Y	Y	Y	GN	GB	TILL
-02	8.9	1.3	7.6	318.2	105.7	212.5	90.7	121.8	2	32	P	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
-03	9.4	2.0	7.4	303.5	102.0	201.5	101.2	100.3	1	0	P	95	5	NA	NA	U	Y	Y	Y	GN	GN	TILL
-04	7.9	1.0	6.9	394.0	96.6	297.4	103.7	193.7	2	10	P	85	15	NA	NA	U	Y	Y	Y	BBN	BBN	TILL
55-01	8.1	1.9	6.2	134.3	89.8	44.5	25.1	19.4	0	NA	P	85	15	NA	NA	U	Y	Y	Y	BBN	BBN	TILL
56-01	4.7	0.5	4.2	71.8	41.4	30.4	22.9	7.5	0	NA	C	99	1	NA	NA	U	Y	Y	Y	GB	GB	TILL
57-01	8.1	1.5	6.6	113.3	57.0	56.3	33.0	23.3	17	6574	P	95	5	NA	NA	U	Y	Y	Y	GG	GG	TILL
58-01	7.8	0.7	7.1	100.7	55.7	45.0	32.5	12.5	8	227	P,C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
59-01	9.1	1.9	7.2	127.1	94.0	33.1	24.8	8.3	0	NA	C	90	10	NA	NA	U	Y	Y	Y	GB	GB	TILL
-02	8.5	2.6	5.9	104.4	70.9	33.5	20.6	12.9	0	NA	C	90	10	NA	NA	U	Y	Y	Y	GN	GN	TILL
61-01	7.8	0.0	7.8	82.4	35.6	46.8	33.3	13.5	0	NA	TR	NA	NA	NA	NA	U	Y	Y	Y	B	B	TILL

OM86-6-C-101



ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2Z2 TELEPHONE (416) 239-3527

OM86-6-C-091

Certificate of Analysis

Certificate No. ODM-13/02/6015 Date: February 27, 1987
Received Feb. 23/87 59 Samples of Concentrates
Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. Elcomb
For Argentex Resources Exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm
ME-87-44-02	100	1115	54	54
45-01	181	556	80	37
02	605	282	98	26
45-03	306	307	126	28
46-01	1349	2319	90	54
47-01	895	320	<1	31
48-01	1749	778	152	56
02	380	907	167	61
03	378	178	118	34
04	289	511	14	40
05	301	357	99	43
06	7750	296	134	35
07	74	145	<1	40
48-08	313	388	116	46
49-01	265	581	147	112
02	2214	982	54	52
03	577	631	128	44
49-04	539	481	96	44
51-01	95	15	44	19
ME-87-51-02	160	13	68	19

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ASSAYERS (ONTARIO) LIMITED

33 CHAUNCEY AVENUE TORONTO, ONTARIO M8Z 2T2 TEL: (416) 298-3527

0M86-6-C-091

Certificate of Analysis

Certificate No. ODM-13/03/6015 Date: February 27, 1987
Received Feb. 23/87 59 Samples of Concentrates
Submitted by Overburden Drilling Management Ltd. c.c. Mr. K. El comb
For Argentex Resources exploration Corp.

Sample No.	Au ppb	Cu ppm	As ppm	Zn ppm	Co ppm
ME-87-51-03	198	72	34	25	
04	368	96	44	23	
05	113	15	59	14	5
06	2090	53	68	19	148
07	237	23240	97	119	660
08	29	762	303	26	280
51-09	42	564	<1	28	185
52-01	287	194	195	38	275
54-01	252	714	<1	39	1375
02	85	1089	178	29	1228
03	30	701	105	27	974
54-04	14	375	141	37	2040
55-01	125	146	156	25	307
56-01	267	245	101	45	140
57-01	5938	391	118	48	137
58-01	438	280	65	59	374
59-01	58	239	128	42	362
59-02	16	1116	85	88	76
ME-87-61-01	134	40	105	28	426

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42A01SE0179 2.10303 EBY

900

NOTE: Laboratory Sample log (1 page) and
Certificate of Analysis (2 pages) were
added from OM86-6-C-091 August 1989.
Additions located at back of report.



Ministry of Northern Affairs and Mines

Report of Work
(Geophysical, Geological, Geochemical and Expenditures)

2.103030

#336/87

- Instructions: - Please type or print.
- If number of mining claims traversed exceeds space on this form, attach a list.
Note: - Only days credits calculated in the "Expenditures" section may be entered in the "Expend. Days Cr." columns.
- Do not use shaded areas below.

W8708-336

Mining Act

Type of Survey(s) Reverse Circulation Drilling	Township or Area Eby Twp. Dist. of Timiskaming
Claim Holder(s) Mary Ellen Resources Expl. Corp.	Prospector's Licence No. T-1566
Address P.O. Box 546, Kirkland Lake, Ont. P2H-3L1	
Survey Company Overburden Drilling Management	Date of Survey (from & to) 19 01 87 03 02 87 Day Mo. Yr. Day Mo. Yr.
Name and Address of Author (of Geo-Technical report) G.S. Sheld S.A. Ansell, C.D.M., 107-15 Capella Ct. Nepean Ont., K2E-7x1	

Credits Requested per Each Claim in Columns at right

Mining Claims Traversed (List in numerical sequence)

Special Provisions	Geophysical	Days per Claim
For first survey: Enter 40 days (This includes line cut time)	- Electromagnetic - Radiometric	
For each additional survey using the same grid: Enter 20 days (for each)	- Radiometric - Other	
Man Days Complete reverse side and enter total(s) here	Geophysical - Electromagnetic - Magnetometer - Radiometric - Other Geological Geochemical	Days per Claim
Airborne Credits Note: Special provisions credits do not apply to Airborne Surveys.	Electromagnetic Magnetometer Radiometric	Days per Claim

Mining Claim			Mining Claim		
Prefix	Number	Expend. Days Cr.	Prefix	Number	Expend. Days Cr.
L	618673	6560	L	640896	6560
	618674	6560		640897	6560
	618675	6560		640898	6560
	618676	6560		640899	6560
	618677	6560		640900	6560
	618678	6560		640901	6560
	620743	6560		640903	6560
	620744	6560		640904	6560
	620745	6560		640905	6560
	620746	6560		640906	6560
	620747	6560		640908	6560
	620748	6560		640938	6560
	620750	6560		640939	6560
	620751	6560		640940	6560
	620752	6560		640941	6560
	620753	6560		640942	6560
	620754	6560		640943	6560
	620755	6560		640944	6560
	620756	6560		640945	6560
	620757	6560		640946	6560
	640893	6560		640947	6560
	640894	6560		640948	6560
	640895	6560		640949	6560

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SEP 16 1987

MINING LANDS SECTION

AUG 25 1987
1200 hours
S

Max credits allowed no 60 days per claim

Total number of mining claims covered by this report of work. **57**

Expenditures (excludes power stripping)

Type of Work Performed (**SEE 77-19**)
Reverse Circulation Drilling

Performed on Claim(s)
L-640465, 640962, 640908, 640966, 640464

L-640904, 640901, 640902, 640900, 640899

Calculation of Expenditure Days Credits

Total Expenditures	Total Days Credits
\$ 55,438.04	3696

Instructions
Total Days Credits may be apportioned at the claim holder's choice. Enter number of days credits per claim selected in columns at right.

For Office Use Only

Total Days Cr. Recorded 3420	Date Recorded Aug 25, 1987	Mining Recorder M. C. W. ...
	Date Approved as Recorded 1987-10-02	Branch Director M. Chamesky

R.M.

Date Aug. 19/87	Recorded Holder or Agent (Signature) S. J. Carmichael
---------------------------	---

I hereby certify that I have a personal and intimate knowledge of the facts set forth in the Report of Work annexed hereto, having performed the work or witnessed same during and/or after its completion and the annexed report is true.

Name and Postal Address of Person Certifying S. J. Carmichael, B.Sc., Mary Ellen Resources Expl. Corp.	Date Certified Aug. 21/87	Certified by (Signature) S. J. Carmichael
--	-------------------------------------	---



Ministry of
Northern Affairs
and Mines
Ontario

Report of Work
(Geophysical, Geological,
Geochemical and Expenditures)

Instructions: - Please type or print.
- If number of mining claims traversed
exceeds space on this form, attach a list.
Note: - Only days credits calculated in the
"Expenditures" section may be entered
in the "Expend. Days Cr." columns.
- Do not use shaded areas below.

Mining Act

②

Type of Survey(s)		Township or Area	
Claim Holder(s)		Prospector's Licence No. 7-1566	
Address			
Survey Company		Date of Survey (from & to)	Total Miles of line Cut
		Day Mo. Yr.	Day Mo. Yr.
Name and Address of Author (of Geo-Technical report)			

Credits Requested per Each Claim in Columns at right

SPECIAL PROVISIONS For first survey: Enter 40 days (this includes line cutting) For each additional survey: using the same grid: Enter 20 days (for each)	Geophysical - Electromagnetic - Magnetometer - Radiometric - Other	Days per Claim
	Geochemical	Days per Claim
MAN DAYS Complete reverse side and enter total(s) here	Geophysical - Electromagnetic - Magnetometer - Radiometric - Other	Days per Claim
	Geological Geochemical	Days per Claim
AIRBORNE CREDITS Note: Special provisions credits do not apply to Airborne Surveys.	Electromagnetic	Days per Claim
	Magnetometer	Days per Claim
	Radiometric	Days per Claim

Mining Claims Traversed (List in numerical sequence)

Mining Claim			Mining Claim		
Prefix	Number	Expend. Days Cr.	Prefix	Number	Expend. Days Cr.
L	640950	6560			
	640951	6560			
	640952	6560			
	640960	6560			
	640961	6560			
	640962	6560			
	640963	6560			
	640964	6560			
	640965	6560			
	640966	6560			
	760582	6560			

RECEIVED
SEP 10 1987
MINING LANDS SECTION

RECEIVED
AUG 25 1987
12:00 noon

Expenditures (excludes power stripping)

Type of Work Performed
Reverse Circulation Drilling

Performed on Claim(s)
640998, 618673, 618674

Calculation of Expenditure Days Credits

Total Expenditures ÷ 15 = Total Days Credits

Instructions
Total Days Credits may be apportioned at the claim holder's choice. Enter number of days credits per claim selected in columns at right.

Date Recorded Holder or Agent (Signature)

For Office Use Only

Total Days Cr. Recorded Date Recorded Mining Recorder

Date Approved as Recorded 1987.10.02 Branch Director R.M. Charnesky

Certification Verifying Report of Work

I hereby certify that I have a personal and intimate knowledge of the facts set forth in the Report of Work annexed hereto, having performed the work or witnessed same during and/or after its completion and the annexed report is true.

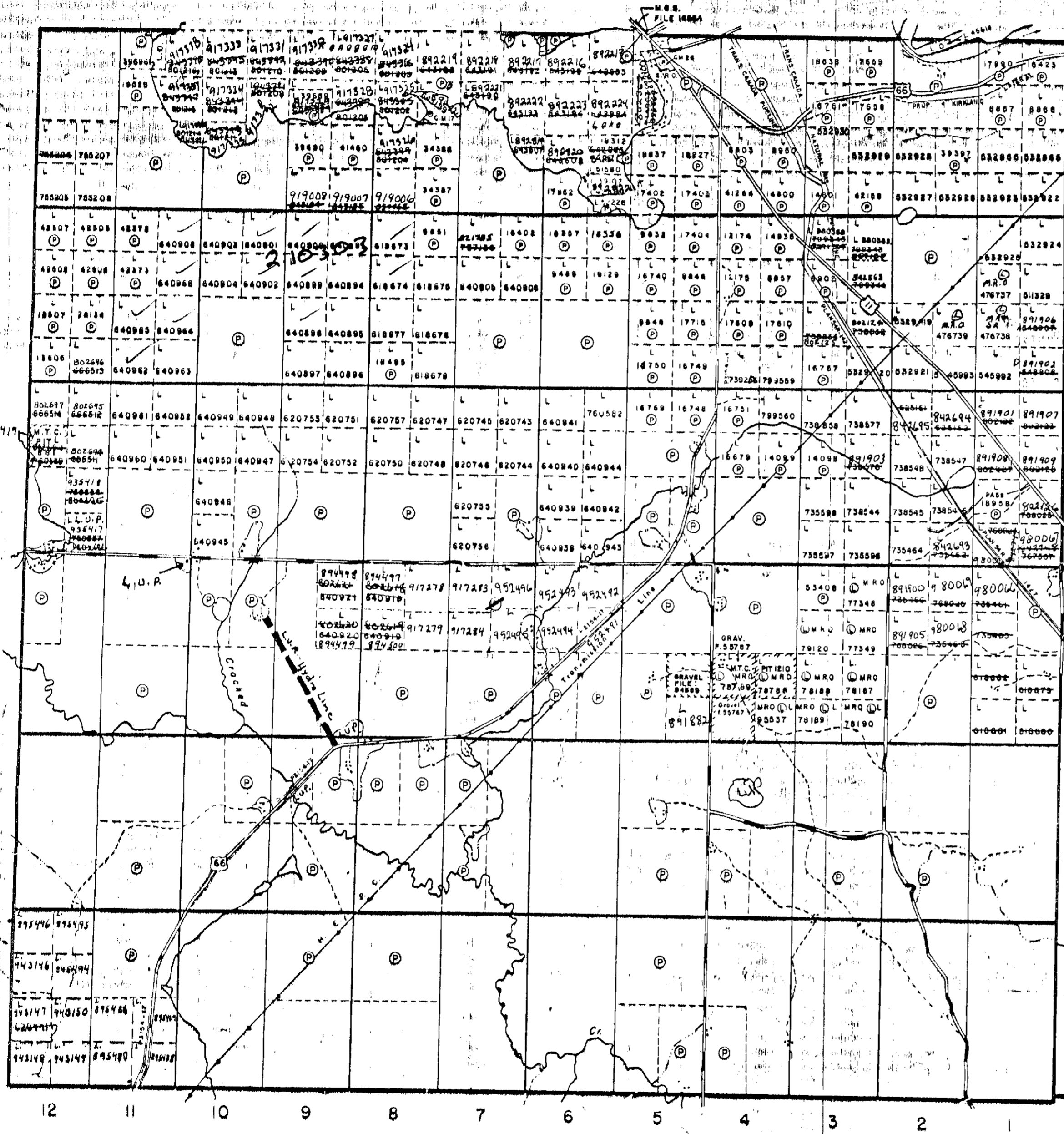
Name and Postal Address of Person Certifying

Date Certified Certified by (Signature)

Grenfell Twp. M.351

Burt Twp. M.334

Otto Twp. M.379



THE TOWNSHIP OF 210303

EBY

DISTRICT OF TIMISKAMING

LARDER LAKE MINING DIVISION

SCALE: 1-INCH=40 CHAINS

LEGEND

- PATENTED LAND Ⓟ
- CROWN LAND SALE C.S.
- LEASES Ⓛ
- LOCATED LAND Loc.
- LICENSE OF OCCUPATION L.O.
- MINING RIGHTS ONLY M.R.O.
- SURFACE RIGHTS ONLY S.R.O.
- ROADS —
- IMPROVED ROADS —
- KING'S HIGHWAYS —
- RAILWAYS —
- POWER LINES —
- MARSH OR MUSKEG —
- MINES Ⓜ
- CANCELLED C.

NOTES

400' surface rights reservation along the shores of all lakes and rivers.

AREAS WITHDRAWN FROM STAKING under Sec.43 of The Mining Act (R.S.O.1970).
 Order No. File Date Disposition

Not Not? Con. 3 - Cert. of Forfeiture - Sept. 17/85

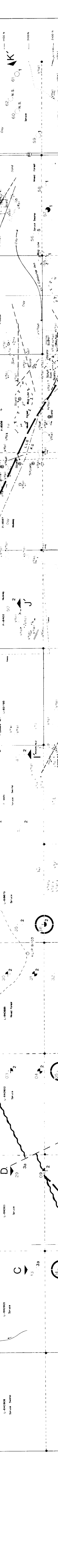
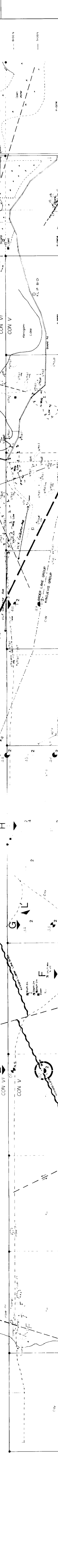
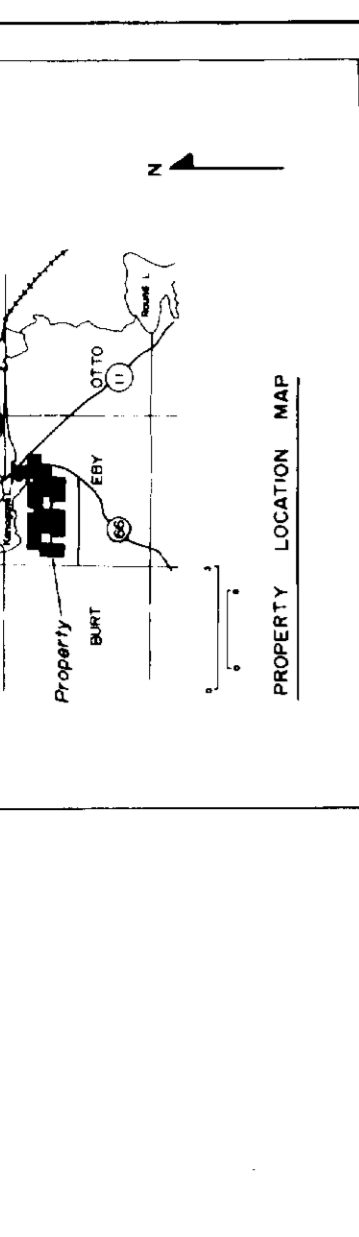
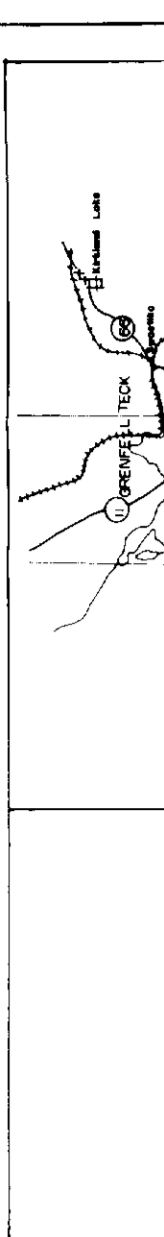
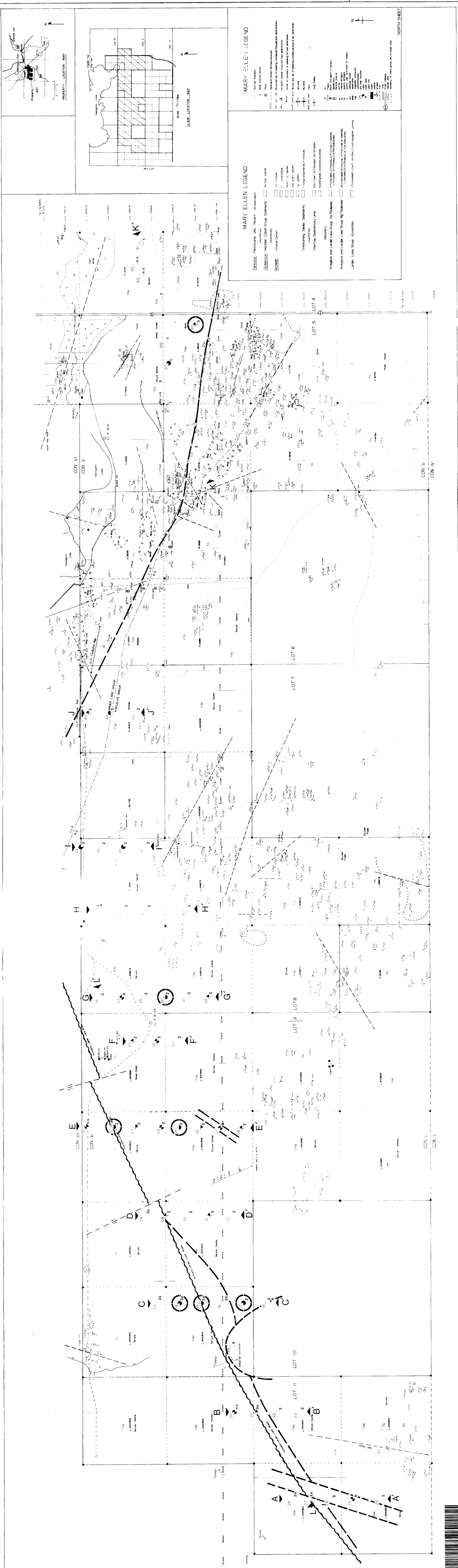
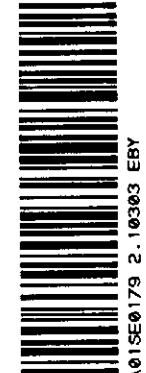
DATE OF ISSUE
 JUL 17 1987
 LARDER LAKE
 MINING RECORDER'S OFFICE

PLAN NO. - M-345

ONTARIO
 MINISTRY OF NATURAL RESOURCES
 SURVEYS AND MAPPING BRANCH

Blain Twp. M.418





MARY ELLEN LEGEND

Geologic Bedrock and Recent Sediments

- Diabase
- Syenite
- Granodiorite
- Metasediments (conglomerate (3a); graywacke (3b))
- Mafic volcanics (basalt)
- Ultramafic volcanics (komatiite flows)

Symbols

- Reverse circulation drill hole no. ME-87-12 with bedrock intersection of Unit 2
- Major lithologic contact identified or confirmed by reverse circulation drilling

Gold Anomalies

- Greater than 1,000 pps gold
- Greater than 10 visible gold grains
- Stratigraphic continuity
- Pathfinder elements
- Potentially significant gold anomaly

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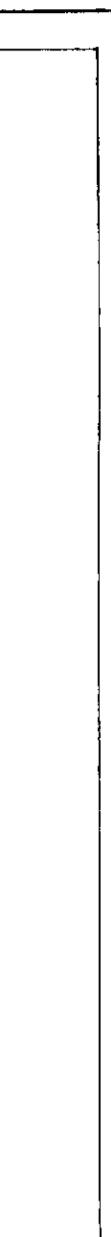
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MARY ELLEN RESOURCES LTD.
EBY PROPERTY
 Eby Twp.
 2

HEAVY MINERAL GOLD ANOMALIES

BY OVERBURDEN DRILLING MANAGEMENT LIMITED JUNE 1997



NORTH SHEET