



42A07SW8456 63.6198 LANGMUIR

010

1990/91 UNDERGROUND EXPLORATION PROGRAM SUMMARY

LANGMUIR # 1 Ni-SULFIDE DEPOSIT

TIMMINS NICKEL INCORPORATED

TECHNICAL SUBMISSION TO ACCOMPANY

OMIP GRANT APPLICATION

DESIGNATION # OM90-118

Prepared by David R Melling, January 15, 1991.



42A07SW8456 63.6198 LANGMUIR

010C

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1.0 Summary

Between August, 1990 and January, 1991 Timmins Nickel Inc. completed an underground exploration program on the Langmuir # 1 Ni-sulfide deposit.

The program included dewatering an existing ramp, extending the ramp (400 ft) to the 315 ft level, substantial exploration drifting (930 ft), raising (390 ft), and underground diamond drilling (4652 ft). In addition, metallurgical testing and ore compatibility studies were completed on bulk samples taken from the deposit.

This exploration work represented a critical step needed to evaluate the feasibility of mining the deposit. Based on the results of this program a production decision is expected in 1991. Underground exploration was still in progress while this report was being written.

This report documents the scope of work completed and accounts for expenditures incurred during the program.

2.0 Introduction

Between August, 1990 and January, 1991 an underground exploration program was completed on the Langmuir # 1 Ni-sulfide deposit by Timmins Nickel Inc. (TNI). The property (Figure 1) on which the deposit occurs is owned 100 % by TNI and all costs for the project were born by TNI. It is located in the Porcupine Mining Division, about 30 km southeast of Timmins, Langmuir Township, Ontario (Figure 2).

The exploration program was designed to evaluate the economic feasibility of mining the Langmuir # 1 Ni deposit which has seen substantial previous exploration, but which has lain idle since 1977. The program described here included the following elements:

- 1) Site preparation, temporary construction, installations and dewatering of a partially completed ramp;
- 2) Underground diamond drilling;
- 3) Extending the ramp, drifting and raising;
- 4) Bulk sampling and metallurgical testing; and,
- 5) Geological mapping and sampling.

The purpose of this report is to document the scope of work completed and account for expenditures (Appendix 1) incurred during the program. Background information such as exploration history, property geology/geophysics and deposit geology and reserves were submitted with TNI's "application for designation" and will not be repeated here.

3.0 Site Preparation, Temporary Construction / Installations and Dewatering

Site preparation completed as part of this program included the following activities:

- 1) Resurfacing, grading and maintenance of site roads;
- 2) Site clearing and levelling for temporary facilities and waste/ore pads;
- 3) Construction of 1/4 mile bush road to powder magazine;
- 4) Construction of 1/8 mile bush road to cap magazine;
- 5) Stripping of overburden for vent raise break through; and,

6) Re-excavation of settling pond.

One D-7 bull dozer, one 966 loader and 4 dump trucks were required to complete the site preparation.

Temporary construction and the installation of the following surface facilities was completed:

- 1) Security gate and trailer;
- 2) Combined shifter / 1st aid office trailer;
- 3) Powder magazine;
- 4) Cap magazine;
- 5) Fuel storage tanks (mine fuel, propane);
- 6) Self contained 650 Kw generator;
- 7) 1350 cfm compressor; and,
- 8) 125 Hp ventilation fan.

Dewatering of the existing ramp (1250 ft) driven by Noranda in 1977 was completed in less than 20 days using a 58 Hp pump. Vent tubing, air, water and power lines were installed underground.

4.0 Underground Exploration Drilling

Two phases of underground diamond drilling were completed involving 19 AQ size DDHs (4652 ft).

Phase 1 was completed after dewatering, but prior to extending the ramp. Nine DDHs (LH-1 to LH-9) were completed between August 20 and September 7, 1990 totalling 3627 ft. The purpose of this drilling was to:

- 1) Confirm the continuity of the mineralized structures previously identified by 100 ft centre surface drilling;
- 2) Establish the widths and grades of the mineralization; and,
- 3) Provide geotechnical information and permit the refinement of exploration drifting layouts.

Phase 2 of underground diamond drilling was initiated after exploration drifting on the 250 level centre zone was largely completed. Between October 15 and 25, 1990 ten AQ size drill holes (LH-10 to LH-19) were completed totalling 1025 ft. The purpose of this drilling was to:

- 1) Provide further confirmation of the continuity and grades of the mineralized structures previously identified;
- 2) In some cases attempt to extend zones of known mineralization and expand reserves; and,
- 3) Refine exploration layouts on the East and West zones.

Drill logs for Phases 1 and 2 are included in Appendix II of this report and illustrated in various plans and sections in Appendix III.

In addition, in excess of 30 underground test holes were drilled with jacklegs using 1 1/4" steel. Sludges from these holes were collected and assayed. This work was completed to establish the limits of the mineralization and the results are illustrated on plans in Appendix III.

5.0 Underground Exploration Drifting and Raising

The purpose of the underground exploration was to confirm the continuity and grade of mineralization indicated by diamond drilling. This was accomplished through a program of ramping, drifting and raising. Extension of the ramp began in September, 1990 and was still in progress at the time this report was written.

The equipment required to complete the underground work included:

- 1 - 3 boom air jumbo
- 1 - 13 ton rock truck
- 2 - 5 yard scoop trams
- 6 - jack legs
- 3 - 2 drum 30 Hp slushers
- 1 - 20" x 30" crusher

To date the ramp (12 X 15) has been extended some 400 ft to the 315 level. Drifting (10 X 10) on the 250 level has amounted to approximately 800 ft. About 300 ft of raising (6 X 6) on mineralization has been completed in 9 raises from the 250 level. Roughly 130 ft of sublevel drifting (10 X 10) has been completed on the 200 ft level. About 90 ft of vent raise (8 X 8) was completed to improve the ventilation. Several remucks and sumps were also excavated and services (power, air, water) were installed as required.

Geological mapping and chip sampling of all exploration headings was also completed. The results of this work is illustrated in various plans in Appendix III.

6.0 Metallurgical Testing and Ore Compatibility Studies

Ore compatibility studies were completed on the Langmuir # 1 underground exploration muck and are documented in Appendix IV. The results of this work indicated that an innovative approach would be required. In fact the concentrate produced during the mill test was rejected by the smelter due to its fine grind and moisture content.

Subsequently a used magnetic cobber was obtained to test an alternative processing method. Approximately 5000 tons were crushed to nominal 1 inch size and processed on site. A reject/concentrate split of 40/60 was obtained prior to milling. It appears that this approach will be cost effective and eliminate a third of the material as waste and produce a third of the material as a middling product for milling and a third of the material as a concentrate for shipping without milling (Appendix IV).

7.0 Discussion of Results

The 1990/91 underground exploration program on the Langmuir # 1 Ni-sulfide deposit was successful in confirming the drill indicated reserves. Although no new reserve calculations have yet been completed it would appear that a modest increase in grade and tonnage may be anticipated. In addition, geological data obtained as a result of this program has delineated new exploration targets.

Based on the results of this program a new mining plan is currently being developed. Information gained from exploration drifting and raising on mineralization has proven valuable for selecting various mining methods and assessing dilution factors. The metallurgical test work has been critical in establishing a process flow sheet and evaluating various milling alternatives.

It is anticipated that this exploration program should lead to a production decision for the Langmuir # 1 Ni-sulfide deposit in 1991. The Ontario Mineral Incentive Program has made a significant contribution to the exploration and evaluation process required to justify this decision.

Appendix I
Detailed Expenditure List

Appendix II
Diamond Drill Logs

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY *LA NG 11112*
 HOLE NUMBER *LA-1*
 GRID REFERENCE *-20*
 TOWNSHIP
 AZIMUTH CLAIM
 DIP ANGLE

46 hole drilled from Face of Ramp.

DRILLING COMPANY

FOREMAN

DIP TESTS:

CORE SIZE *AQ*

CORE STORED AT:

LOGGED BY *KHS*

DATE *Aug 22/93*

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS	
				Ni	
<i>0-35</i>	<i>Andesite grey green amygdaloidal</i>	<i>40.5-44</i>	<i>66451</i>	<i>0.05</i>	
<i>34-101.8</i>	<i>Ultramafic Pyroxite - minor - 1% Sulf</i>	<i>44-48</i>	<i>52</i>	<i>0.06</i>	
	<i>41.0 - fault zone 2.5 inches</i>	<i>48-53</i>	<i>53</i>	<i>0.07</i>	
	<i>48-60.0 1% Py in some matrix</i>	<i>53-58</i>	<i>54</i>	<i>0.06</i>	
	<i>60.0-62.5 Magnetite</i>	<i>58-60.6</i>	<i>55</i>	<i>0.06</i>	
	<i>62.5-67.0 core fractured</i>	<i>60.6-62.5</i>	<i>56</i>	<i>0.02</i>	
	<i>69-79 heavy quartz carb inclusions</i>	<i>62.5-67.5</i>	<i>57</i>	<i>0.09</i>	
	<i>79-85.5 3-4% Py</i>	<i>67.5-71.5</i>	<i>58</i>	<i>0.15</i>	
		<i>71.5-75</i>	<i>59</i>	<i>0.09</i>	
<i>101.8-</i>	<i>Andesite</i>	<i>75-79</i>	<i>60</i>	<i>0.11</i>	
<i>216.2</i>	<i>130-131 fault to cement on slip also muggy</i>	<i>79-83</i>	<i>61</i>	<i>0.19</i>	
	<i>the stringer</i>	<i>83-85.5</i>	<i>62</i>	<i>0.01</i>	
	<i>@ 142 core rusty w- 5% Py</i>	<i>85.5-90</i>	<i>63</i>	<i>0.16</i>	
		<i>90-95</i>	<i>64</i>	<i>0.19</i>	
<i>262-280</i>	<i>Dacite - chloritic zone 1-2'</i>	<i>95-100</i>	<i>65</i>	<i>0.11</i>	
	<i>270.5-272.5 massive sulfides 40% Py</i>	<i>100-101.5</i>	<i>66</i>	<i>0.10</i>	
		<i>101.5-105</i>	<i>66467</i>	<i>0.01</i>	
<i>280-304</i>	<i>Mafic Volcanic-chloritic - minor Py</i>				
	<i>285-304 - heavy chloritic 4/M??</i>	<i>270.5-272.5</i>	<i>68</i>	<i>5.58</i>	
<i>304-319.0</i>	<i>Pyroxite - 1% sulf 304-307</i>	<i>282-283</i>	<i>67</i>	<i>0.02</i>	
		<i>282-286</i>	<i>72</i>	<i>0.06</i>	
<i>319.0</i>					
<i>309.7</i>	<i>END OF LOG</i>	<i>304-307</i>	<i>71</i>	<i>0.16</i>	

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1 of 1

Use Hole Drilled from Face of Ramp

PROPERTY *Timmins*
 HOLE NUMBER *L11-2*
 GRID REFERENCE
 TOWNSHIP
 AZIMUTH
 CLAIM
 DIP ANGLE

DRILLING COMPANY
 CORE SIZE *AG*

FOREMAN
 CORE STORED AT: *Recovery*

DIP TESTS:
 LOGGED BY *K Hicks*

DATE *11/15/10*

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0-30	Basalt-talcite - med-dark green, amygy, carb blch local, fq carb stringers 1% py	31-36 36-41 41-46	66489 66490 66491	0.06 0.08 0.08
0-11	waxy leucosarc?	46-51	66492	0.07
11-24	generally more sulph - 1-3% dark cubes, bands	51-56	66493	0.08
15.4-30	becom. more siliceous	56-61	66494	0.09
		61-66	66495	0.07
	var. Basalt-talcite - purish grey to dark green, fq, v soft, talcose	66-71	66496	0.07
	variable amt in carb stringers, carb blch	71-76	66497	0.13
	Py cubes, bands to 5% locally (PN?)	76-79	66498	0.08
	32.5-33.5 broken core	79-82.3	66499	0.14
	37 1.5" crush flt - silic sulphidic increasing downhole			
57.1-117	Diorite - Med grey - tuffaceous / locally amygy coll bed @ 80' tca minor - 2% dark py 1cm py cubes			
122.3	Porphy type / porph tuff shard - ser altg, carb stringers carb blch, minor gv			
134.2	Andesite - tuffaceous banded carb stringers, local blch, to 1% py min chf			
188.8-194.2	Porph dyke - sh - ser altg carb veins - in brick red matrix?			

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY: LANGMUIR #1

PAGE 2
HOLE NUMBER: 48-2

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
	203.6 - 214.1 increasing sct altm - patchy talc			
221.2 - 222	Cmb vn. rusty walls - crds mud			
202 - 222	Doubt l. med gray fq. up tuffaceous sct shavings, vgn blch fr. 2-3% py locally - some seams			
288 - 289.7	2-3% Py	276 - 281	66500	.01
292.2 - 293	UM Pentolite - dark gray - fq. V mag	281 - 286	66401	.01
	- wk fract - carb / local chlorite	286 - 290	66402	.01
292.2 - 309	1-2% sulph	290 - 292.2	66403	.54
292.2 - 293	chlorite altm - prev	292.2 - 296	66404	.47 .34
307 - 347	st mud talc cmb altm	296 - 299	66405	.50 .47
	Exact blch, fr. 1% sulph	299 - 302	66406	.15 .50
		302 - 307	66407	.26 .15
		307 - 309	66408	.53 .26
		309 - 314	66409	.22 22 .33
		314 - 317	66410	.18 18 .22
		317 - 324	66411	.18
		324 - 329	66412	.16
48-2	6-24			

FAULT

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1 of 2

PROPERTY
HOLE NUMBER
GRID REFERENCE
TOWNSHIP
AZIMUTH

L1125
211-3

CLAIM
DIP ANGLE

u/s Hole Drilled from face of Ramp

DRILLING COMPANY
CORE SIZE *AQ*

FOREMAN
CORE STORED AT: *Leostone*

DIP TESTS:
LOGGED BY *X Hicks*

DATE *Aug 30/90*

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0-25.1	<i>Basalt/Andesite - med grey, ch/ amy, carb, carb stringers, th py</i>			
0-25	<i>Leucocryst common - white</i>			
32.3-33	<i>py cubes - 1cm - amy</i>			
		35-40	66413	0.04
		40-45	66414	0.06
35.1-37.1	<i>Andesite - med grey - grey green</i>	45-50	66415	0.05
	<i>U soft - talc carbonate</i>	50-55	66416	0.06
	<i>med - 5/8 mag</i>	55-60	66417	0.09
	<i>2-3% dss py cubes throughout</i>	60-65	66418	0.09
		65-70	66419	0.07
38.6-38.8	<i>perovskite sep altn - flt?</i>	70-75	66420	0.10
		75-80	66421	0.19
37.7-41.1	<i>Diorite - med grey, hard, triffaceous, amy, carb amy, carb stringers/bands</i>			
	<i>2-3% py at upper cut</i>			
41.1-26.8	<i>Andesite - banded / triffaceous?</i>			
	<i>Med - dk grey, to 1% py</i>			
	<i>mag at carb stringers</i>			
191-209	<i>sp. bl. tuff?</i>			
217.5-230.8	<i>shaly py - carb / s1 blch</i>			
229.8-230.3	<i>py cubes</i>			
237-26.8	<i>seph common - hard - dss locally massive</i>			
	<i>mostly py</i>			
136-26.8	<i>amy common to 4mm dia</i>			

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY:

HOLE NUMBER: 231-2

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
268-381	UM peridotite - dross to left wk part local all			
268-270.4	Massive Sulph Zone	268-270.4	66422	1.64
270.4-285.1	Net texture Sulph Zone mineral part in mass slams	270.4-275	66423	2.29
		275-280	66424	3.04
285.1-301.3	3-5% dross Sulph	280-285	66425	3.39
301.3-303.8	3-5% clotty Sulph	285-290	66426	1.62
303.8-310.5	Sulph pods	290-295	66427	1.22
310.5-353	patch to dross sulphs 3-5% local massive blebs	295-300	66428	1.12
		300-303.8	66429	1.04
		310-315	66430	.66
		315-320	66431	.81
		320-325	66432	.48
381	EOH	325-330	66433	.50
		330-340	66434	.78
		340-345	66435	.36
		345-350	66436	.34
		350-355	66437	.24
		355-360	66438	.34
		360-365	66439	.35
		365-370	66440	.33
		370-375	66441	.55

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY: LANEMUIR #1 HOLE NUMBER: #5 PAGE _____

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
256-258.3	MA. SULP. P.O. PN CPY		66478	
258.3-259.3	PERD LARGE BLOB MAS Sup		66479	
259.3-261.9	MA SULP		66480	
261.9-264	20% SULP.		66481	
264-283	~10% +1-5% SULP IN PERID		66482	
283-296	" " " " " "		66483	
296-301	~5%		66484	
301-304	1-2% SULP WEAKLY DIS		66485	
304-317	" " " " "		66486	
317-319.4	25% MA SULP		66487	
319.4-320.4			66488	

Don Mack

705
642 3244

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY
HOLE NUMBER
GRID REFERENCE
TOWNSHIP
AZIMUTH

LAND...
2...

CLAIM
DIP ANGLE

DRILLING COMPANY

FOREMAN

DIP TESTS:

CORE SIZE

CORE STORED AT:

LOGGED BY

JAC

DATE

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
18.2-22	GREEN... CARB... MICRO FRACTURES			
	CONTAINING PY BLUES			
20-21	14.8-17.6... QZ-CARB? - R-SPAR 20%	14.8-17.6	66472	.08
	... JERRY SOFT, TALCOSA	17.6-19.4	66473	0.10
	21% SPARKED SULPH PY PH?	19.4-21.2	66474	.20
		21.2-23.0	66475	.31
	69.6-69.65 QZ URIN, MINOR CARB, TR SULP.	69.6-69.65	66476	.20
	28.4-28.6 SEMI... GOUGEY, TR SERP.			
	44.1-44.1 CARBON			
	57.0-57.3 5% SULP GLASS			
	60.3-60.6 " " "			
	60.5-60.7 NUMEROUS CARB MICRO FRACTURES			
80-81	... RARE PY GLASS.			
	180.3 ... SERP			
101-102	194.3-194.3 CHL/SERP... SULP CORES			
	194.3-194.3 POSSIBLE COALINER			
193.1-198.5	QZ URININE (FRODO...)			Cu
	... 22% SULP RARE SULP CORES			
	198.5-199.4 CHLORITE RICH			
	204.3-205.2 SERP PIT - VERY GREEN.	242-244	66477	0.01
	... 30° TCA	244-246	66478	0.01
	242.5-243.7 FRACTURED - PUSSEY 1% SULP	246-247	66479	3.77 0.27
	246.7-246.7 MA SULP. PO, PH? 10% CPY			0.04
	252.3-252.5, 252.3-252.8 ~10% PY PHO			

ALL P. 2

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY: LAKEVIEW #1

HOLE NUMBER: 2

PAGE 1

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
	24.5-40.0 1.2% SULEP			0.01
	40.0-41.0 2.0% SULEP			
	41.0-42.0 1.0% SULEP			
	42.0-43.0 1.0% SULEP			
	43.0-44.0 1.0% SULEP			
	44.0-45.0 1.0% SULEP			
	45.0-46.0 1.0% SULEP			
	46.0-47.0 1.0% SULEP			
	47.0-48.0 1.0% SULEP			
	48.0-49.0 1.0% SULEP			
	49.0-50.0 1.0% SULEP			
	50.0-51.0 1.0% SULEP			
	51.0-52.0 1.0% SULEP			
	52.0-53.0 1.0% SULEP			
	53.0-54.0 1.0% SULEP			
	54.0-55.0 1.0% SULEP			
	55.0-56.0 1.0% SULEP			
	56.0-57.0 1.0% SULEP			
	57.0-58.0 1.0% SULEP			
	58.0-59.0 1.0% SULEP			
	59.0-60.0 1.0% SULEP			
	60.0-61.0 1.0% SULEP			
	61.0-62.0 1.0% SULEP			
	62.0-63.0 1.0% SULEP			
	63.0-64.0 1.0% SULEP			
	64.0-65.0 1.0% SULEP			
	65.0-66.0 1.0% SULEP			
	66.0-67.0 1.0% SULEP			
	67.0-68.0 1.0% SULEP			
	68.0-69.0 1.0% SULEP			
	69.0-70.0 1.0% SULEP			
	70.0-71.0 1.0% SULEP			
	71.0-72.0 1.0% SULEP			
	72.0-73.0 1.0% SULEP			
	73.0-74.0 1.0% SULEP			
	74.0-75.0 1.0% SULEP			
	75.0-76.0 1.0% SULEP			
	76.0-77.0 1.0% SULEP			
	77.0-78.0 1.0% SULEP			
	78.0-79.0 1.0% SULEP			
	79.0-80.0 1.0% SULEP			
	80.0-81.0 1.0% SULEP			
	81.0-82.0 1.0% SULEP			
	82.0-83.0 1.0% SULEP			
	83.0-84.0 1.0% SULEP			
	84.0-85.0 1.0% SULEP			
	85.0-86.0 1.0% SULEP			
	86.0-87.0 1.0% SULEP			
	87.0-88.0 1.0% SULEP			
	88.0-89.0 1.0% SULEP			
	89.0-90.0 1.0% SULEP			
	90.0-91.0 1.0% SULEP			
	91.0-92.0 1.0% SULEP			
	92.0-93.0 1.0% SULEP			
	93.0-94.0 1.0% SULEP			
	94.0-95.0 1.0% SULEP			
	95.0-96.0 1.0% SULEP			
	96.0-97.0 1.0% SULEP			
	97.0-98.0 1.0% SULEP			
	98.0-99.0 1.0% SULEP			
	99.0-100.0 1.0% SULEP			

227-227 #1562 → 0.258 Ni 47

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY LANDMARK #1
 HOLE NUMBER LH-6
 GRID REFERENCE
 TOWNSHIP
 AZIMUTH
 CLAIM
 DIP ANGLE

DRILLING COMPANY

FOREMAN

DIP TESTS:

CORE SIZE AQ

CORE STORED AT: REDSTONE

LOGGED BY K Hicks

DATE Sept 1 20

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0 - 11	dark ls, med greenish, carb wisps - amygdaloids??, carb stringers to py blks locally chloritic			
35.7 - 51.5	Brownish - Eg dyke?? lower cont associated?			
52.2 - 57.1	above			
11.1 - 14.1	11.1 - 14.1 Pyroclastic - med-dk greenish gray, talcose, carb wk - st. mag			
	11.1 - 12.5 dark silty sh (py) blks - some py			
	local frags to carb ff			
	eg chert at top - very magnetic			
13 - 14.9	Magnetite / Chromite? - massive seam			
72 - 84	carb young common - brown			
94.2 - 94.3	contact shear - sep talcose			
94.6 - 100.5	brn ls - med greenish, med hard carb wisps common, blotchy carb possibly tuffaceous over lam (Pal)			
105.2 - 132.8	11.1 - 12.5 med gray, v hard, seg, chert masses tuffaceous, locally cherty exact in blk halos - some ls med carb veining in cubic py to dyp py			
121.1 - 132.8	ls pyroclastic?			

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY: LANGMUIR #1

PAGE 243
HOLE NUMBER: 211

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
172.2-215.2	Andesite / Diorite med greenish fq buffaceous locally tan / for locally 3-4% py cubes carb in sp's, stringers, chertitic hard			
179.4-180.7	porph dyke ?? sulph common at cnt			
194-214.6	magnetite common wk ferriferous non fm	179.4-180.7	66803	1.80%
214.6-215	magnetite stringers			
215-215.25	massive sulph po, py	214.6-215.3	66804	2.14
215.2-220	Dacite Flow med-dark grey aph fq 219.6-220 - flow to box			
220-232.5	Dacite Flow med-dark grey aph fq very hard, locally porph xtal buff cherty magnetite stringers			
227.5-231.2	shear - possible porph dyke rust carb stringers			
232.5-241.1	F-Porph dyke - minor shear, sci alter carb facts			
241.5-270.4	Andesite med greenish carb amyg up to 5% py cubes locally locally porphyritic as bands	262.8-265.4	66805	2.00
262.8-268.9	up to 5% py cubes as bands			
270.4-271	massive sulphide - po, py fm	270.4-271.6	66806	2.60
271.4-271.6	25% sulph clots			
270.4-281.5	Dacite / Andesite - d grey, very hard - amyg? porph? local carb blsh 1% py blsh			
281.4	1/2 qtz VN			
284-284.2	30% sulph - py some po	283.6-284.6	66807	3.05
284.4-285.2	bx 300 in AU			

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY:

PAGE 2115
HOLE NUMBER: LN-6

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
286.7-293	magnetite seams - massive sections common			
	local massive magnetite			
293.7-296.5	5-10% sulph blubs in massive magnetite	295.7-296.5		
296.5-299.1	thinly bedded magnetite - pe, pm, py in massive magnetite seams / frags	296.5-299.1		2.52
299.1-301.3	cherty rock - unknown origin possible U.M. 2-3% po,	299.1-301.3		0.43
301.3-311.5	11M Peridotite med-dark greyish green talcos. soft - v soft, carb blubs fract in carb			
330.1-335.5	pyroph. dyke 3-4% dens sulph	330.1-335.5	66811	0.27
335.5-338.6	1-2% dens sulph	335.5-338.6	66812	0.29
338.6-343	seep fill zone	338.6-343	66813	0.19
343-352.1	2-3% sulph	343-352.1	66814	0.11
352.1-356.5	2-3% sulph	352.1-356.5	66815	0.22
356.5-366.3	1-2% sulph	356.5-366.3	66816	0.46
366.3-372	2-3% sulph	366.3-372	66817	0.27
372-374.3		372-374.3		
374.3-377		374.3-377		
377	carb			

3.33
5.6

0.24 / 10.9
0.11 / 1.7
0.27 / 2.0
0.63 / 7.0

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY
HOLE NUMBER
GRID REFERENCE
TOWNSHIP
AZIMUTH

LANSING
LH-7

CLAIM
DIP ANGLE

UG / fine or Rasp

DRILLING COMPANY
CORE SIZE 118

FOREMAN
CORE STORED AT: Redstone

DIP TESTS:
LOGGED BY R. Hicks

DATE 8/2/57

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0-70	Basalt/Andesite - med green, lg locally diffused local chloritic at top / scuffs mod part w @ carb veining carb wisps, blebs, some blk h trace to 3-4% py locally			
	33.2-34 @ carb vein - to py			
	44.4-45.1 @ carb vein, Blk, to py			
70-77.3	UM Andesite med greenish, soft to coarse wk part in carb ft - 1-2% sulph			
	70-71.5 sph chert masses - massive - some massive magnetite, chlorite			
	81-84 2-3% py	81-84	6600	0.00
	83.2-83.6 sulph E. of ft w 1" chert/sulph mat			
	97.6-99.3 lg chert irregular - non mag			
110-117	Pyrite dyke feld porph			
117-119	Andesite Med/dark green, sph lg buff? carb wisps, stringers 5-10% py as super to 1" - no po, pn some py stringers			
119-121	Dacite lg Mt/Med green, lg sph v hard, subconch possible appt. @ 114-114.3? chloritic nodules - locally porph			
	127.7-145 pyrite common - 10-15% locally some massive stringers			

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY: LANGMILLER 1/1

PAGE 20/27
HOLE NUMBER: 207

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	SAYS
145-207	tuffaceous?? lam/foliation well developed locally porph - possible dykes - contacts unconsolidated			
184.2-185.2	var. altn blk/y zone fault @ 185.5 - 8"			
186.2-198.2	variable slt altn - possible porph dyke zone			
170-172	brick rd QU common - granite??			
202-245.1	Andesite/Dacite Tuff - dark green, fa laminated locally gneiss, porphyritic magnetic rich sections, chloritic py to 5% local clusters			
207-214	Porphy Dyke - sharp cut in chl altn wk porphyritic near lower cut wk part - carb stringers			
217-271.1	Oxide Fe Formation?? very magnetic, possibly a tuff 3-5% cubic py disseminated carb altn + stringers becoming less magnetic & more more tuffaceous	267.1-271.1	1001	GR
		292.1-297.1	1002	GR
		297.1-299.5	1003	GR
271-272.5	Magnetic Sulfide Zone - in magnetic bands/frags minor QU py, py pn??			Shoulder?
272.5-272.7	CHERTY TUFF/SED -			
312.2-312.7	Magnetic/Sulfide Zone - OXIDE Fe FORMATION??	312.2-312.7	1004	GR
312.7-312.8	UM PERIDOTITE med/dark green fa - locally talouse - blotchy carb altn mod part in carb stringers			

407
110
5'

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY: 2000/11/17/1

PAGE 1
HOLE NUMBER: 13

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	SAYS
	316.7 - 318 3-4% Sulph	316.7-318	66925	0.85
	320.3 - 325 2-3% Sulph	320.3-325	66926	0.20
	325 - 328.8 3% dust / not sulph	325 - 328.8	66927	0.53
	328.8 - 357 1-2% Sulph local 2-3%	328.8-334	66928	0.25
	357 - 392.5 - spotty dust sulph	334 - 337	66929	0.27
		337 - 344	66930	0.32
		344 - 349	66931	0.25
		349 - 351	66932	0.29
		351 - 357.3	66933	0.26
		357.3 - 362	66934	0.27
362 - 367	Porphyry - Turb. Dyke Eq. phyn. to 2mm hard, hch cont for 13' hole	362 - 367	66935	0.18
		367 - 372	66936	0.18
		372 - 377	66937	0.27

CHECK TAGS
CHIP TAGS
(DUPLICATE)

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1 OF 3

46/ Face of Ramp

PROPERTY *University*
 HOLE NUMBER *L4-7*
 GRID REFERENCE
 TOWNSHIP
 AZIMUTH
 CLAIM
 DIP ANGLE

DRILLING COMPANY
 CORE SIZE *AG*

FOREMAN
 CORE STORED AT: *KEDSTONE*

DIP TESTS:
 LOGGED BY *K/H*

DATE *5/2/77*

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
<i>0-108.2</i>	<i>Andesite med/dark green Aph-Fg med hard bract amyg minor carb w/ps, stamens, some calc pale blch local py clotting - 3-4%</i>			
<i>0-72</i>	<i>low local - tufaceous?</i>			
<i>5-7</i>	<i>Greened core</i>			
<i>10-11</i>	<i>Greened core</i>			
<i>15.6</i>	<i>1/2" black dyke?</i>			
<i>27.5-29.65</i>	<i>green - brown</i>			
<i>36.6-39.8</i>	<i>aph/amyg</i>			
<i>72.0</i>	<i>Contact</i>			
<i>72-108.2</i>	<i>Very massive - blue</i>			
<i>108.2-217.7</i>	<i>IM PERMITIVE - dark green - black G med hard - 12/15 soft local blch / sep alt - minor 3-4% dark Py/Pa? throughout</i>	<i>108.2-112</i>	<i>66</i>	<i>0.06</i>
		<i>112-114.6</i>	<i>67</i>	<i>0.26</i>
		<i>114.6-117.6</i>	<i>68</i>	<i>0.30</i>
<i>108.2-114.6</i>	<i>blch cont zone - scapalt alt - minor</i>	<i>117.6-121</i>	<i>69</i>	<i>0.12</i>
<i>114.6-147.7</i>	<i>in blch talcose alt - med fract in carb filling</i>			
<i>147.8-213.1</i>	<i>relatively fresh UM - black med fract in carb / G</i>	<i>125-129</i>	<i>66865</i>	<i>0.16</i>
		<i>129-134</i>	<i>66</i>	<i>0.28</i>
<i>174.1-174.6</i>	<i>QU BX</i>	<i>134-139</i>	<i>67</i>	<i>0.15</i>
<i>195-195.5</i>	<i>broken core</i>	<i>139-144</i>	<i>68</i>	<i>0.14</i>
<i>197.2-198.3</i>	<i>@ carb vein</i>	<i>144-149</i>	<i>69</i>	<i>0.16</i>
<i>213.1-219.7</i>	<i>blch cont zone - pervasive sep</i>	<i>149-154</i>	<i>70</i>	<i>0.10</i>
<i>214.8-217.2</i>	<i>possible spur/pt</i>	<i>154-157</i>	<i>71</i>	<i>0.18</i>
		<i>157-162</i>	<i>72</i>	<i>0.13</i>
<i>177.7-232.5</i>	<i>DACITE TUFF? Very magnetic mass lenses cont hard G</i>	<i>162-167</i>	<i>73</i>	<i>0.13</i>
		<i>167-172</i>	<i>74</i>	<i>0.14</i>
<i>222.7-228.6</i>	<i>Pyph dyke, chd locally from alt. vein</i>	<i>172-177</i>	<i>75</i>	<i>0.16</i>

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY: L9N6MUIR #1

PAGE 201-3
HOLE NUMBER: L11-8

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
232.5 - 204	11M PERIDOTITE mid green to dark green, shd / foliated carb streaks, blotches, magnetic	177-182 182-186	76 6877	0.27 0.29
232.5 - 234	semi massive sulphide zone 30-40% sulph ¹⁵	232.5 - 234	66870	4.22
234 - 237.2	5-10% sulph blotches	234 - 237.2	66841	1.84
237.2 - 240.3	5% sulph	237.2 - 240.3	66842	1.10
240.3 - 245	Massive Sulphide	240.3 - 245	66843	6.75
245 - 247	2-4% disint ^d sulph	245 - 247	66844	1.10
247 - 247.7	Massive Sulph	247 - 247.7	66845	6.11
247.7 - 248.2	5% sulph - Massive Magnetite	247.7 - 248.2	66846	1.54
248.2 - 248.6	Massive Sulph	248.2 - 248.6	66847	6.65
248.6 - 254	5% dis sulph	248.6 - 254	66848	1.38
		254 - 259	66849	1.63
		259 - 264	66850	1.51
		264 - 268.0	66851	1.74
		268 - 271.5	66852	0.79
271.5 - 272.2	Mass Sulph	271.5 - 272.2	66853	3.80
272.2 - 276.6	5% disint ^d Sulph	272.2 - 276.6	66854	1.28
276.6 - 278.5	Mass Sulph	276.6 - 278.5	66855	4.97
278.5 - 282	3-5% dis sulph	278.5 - 282	66856	1.59
		282 - 287	66857	1.04
		287 - 292	66858	1.85
		292 - 297	66859	1.29
300.8 - 301.5	Massive Magnetite 3-5% Sulph	297 - 301.5	66860	1.13
301.5 - 304	Massive Sulph	301.5 - 304	66861	1.52
304 - 332	ANDESITE - tuffaceous w/ interbedded oxide Fe fm?? or magnetite rich layers amyg ^{ul} carb. sand streaks	186-191 191-196 196-201 201-206	66878 79 80 81	0.59 0.27 0.25 0.38
332 - 375.5	PORPHYRY DYKE / TUFF - 2-4 mm scattered phenos 10% WR fol locally - massive magnetite (2/M) 4%	206-213 213-217 217-219.7	82 83 84	0.29 0.35 0.29

155.387
2.12%
7.5

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PROPERTY *1111*
 HOLE NUMBER *44 10*
 GRID REFERENCE *3111*
 TOWNSHIP *3111* CLAIM
 AZIMUTH *090* DIP ANGLE *1000*

DRILLING COMPANY
 CORE SIZE

FOREMAN

CORE STORED AT:

DIP TESTS:

LOGGED BY *JAS*

DATE

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0 - 10	<i>EXPOSE ?</i>			
10 - 14	<i>10</i>			
14 - 16.4	<i>Urn 14.15.45 magnetic 10% Sulp.</i>			
	<i>15.45 - 15.95 Ma Sulp.</i>	<i>14.0 - 16.4</i>	<i>1652</i>	
	<i>15.95 - 16.4 chloritic, mt, 2-3% Sulp.</i>			
16.4 - 59	<i>Ua near to and chl</i>	<i>28.5 - 29.5</i>	<i>1453</i>	
	<i>28.6 - 29.1 Ma Sulp.</i>			
59 - 65.5	<i>Dike.</i>			
65.5 - 92	<i>Ua, in all chl (siliceous? locally), occasional mt bands, amycolites py</i>			
92 - 103	<i>Magnetic (unsp.?)</i>			
103 - 110.5	<i>Plastic Ua chl</i>	<i>119.0 - 120.8</i>		
	<i>Siliceous → G-7 E minor mt bands</i>	<i>2' 120.8 - 122.8</i>		
110.5 - 120.8	<i>Plastic Ua chloritic 1-2% py po alloys</i>	<i>205 122.8 - 124.85</i>		
		<i>5.15 124.85 - 130</i>		
120.8 - 124.85	<i>MS, po, py rich near 121, mt rich 123 → 124.85</i>	<i>5 124.85 - 135</i>		
		<i>5.5 135 - 140.5</i>		
124.85 - 135	<i>Urn 124.85 - 135 - 10-15% diss Sulp magnetic</i>	<i>3</i>	<i>1067</i>	<i>1.375</i>
	<i>135 - 140.5 carb rich Ua < 2% Sulp non magnetic</i>	<i>4.5</i>		<i>1.04</i>
	<i>140.5 - 147.5 - 10% Sulp magnetic</i>	<i>5</i>		<i>0.566</i>
	<i>147 - 203 siliceous + carbon tail</i>			
	<i>147 - 203 siliceous carbon tail RRD 30</i>			
	<i>generally diss Sulp locally.</i>		<i>2.17</i>	
	<i>several possible Sulfars</i>	<i>120.8 - 148</i>	<i>27.2</i>	
203	<i>FOH</i>			

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY
HOLE NUMBER 1411
GRID REFERENCE
TOWNSHIP
AZIMUTH

CLAIM
DIP ANGLE

DRILLING COMPANY
CORE SIZE

FOREMAN

DIP TESTS:

CORE STORED AT:

LOGGED BY JAS

DATE

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
10-10	...			
16-55-16	16.55-17.75 ACT/TORON + 4.5% SULEP 17.75-18 ... SULEP	3.20% 1.45	1500	1.34 5.09
15-76	Qst U... RARE CHL RICH FRAGS, 1/2 SULEP ALONG FRACT.			
17-4	280-29 ... BROKEN ...	1.4'	1500	5.465
20-4-71-2	... WAD ... L-SPAR AMYGDULES TOURMALINE LOOKING SECTIONS			
20-4-71-3	... PICKUP ...			
20-4-71-4	Var. 1.7% SULEP BLEND (PI?) RECOMMENDING MORE CHL RICH TOWARDS B1		79.5-81.2	1678 0.03
		4.2	81.8-86	1678 3.146
		4	86-90	1678 0.023
176	... 10-15% SULEP ...	4	90-94	1678 2.557
	... 87% ...	4	94-98	1678 2.732
	... CRACK ...	4	98-102	1678 1.48
112-117	... MASSIVE + GRANULAR DO DA??, CHL ...	4	102-106	1678 0.686
112-118	... 15% ...	4	106-110	1678 2.991
112-122	... MAS ...	4	110-114	1678 7.803
		4	114-118	1678 4.013
		3'	118-122	1678 4.664

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

2042009

PROPERTY
HOLE NUMBER
GRID REFERENCE
TOWNSHIP
AZIMUTH

LH 11

CLAIM
DIP ANGLE

DRILLING COMPANY
CORE SIZE

FOREMAN

CORE STORED AT:

DIP TESTS:

LOGGED BY *JAS*

DATE

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
102-131	15-70% sulf + int (200-250)	5' 122-172	1679	2.835
131-155	1.7% sulf, more sulf, scattered 170-171	4' 172-176	1680	1.9
	133.5 - 163.8 172-173 ?		1681	0.07
			1682	0.18
	163.8 5% sulf		1683	0.12
	170-176 - UNIQUE TEXTURE/FABRIC CRYSTAL ALIGNMENT			
	174.5-176 SPINIFER.			
26.104				
			2.77	
		81.8-131	49.2	

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY *L.A. Nickel*
 HOLE NUMBER *LH 13*
 GRID REFERENCE
 TOWNSHIP
 AZIMUTH
 CLAIM
 DIP ANGLE

DRILLING COMPANY
 CORE SIZE

FOREMAN

CORE STORED AT:

DIP TESTS:
 LOGGED BY

DATE

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0-3	Very chloritic & Pb rich 3-5% of diss sulfa occas. po patch	0-3		0.05
3-11	Very weakly chl 8-9" 1" ground			
11-18	Chl + mt. kaolined matrix (off?)			
18-27.5	Very weakly chl.			
27.5-33.2	Felsic dyke - pinkish stain at fract.			
33.2-41	Very weakly MOD CHL. zones of silicification, po bleiss, mt banding, 'py' cubes			
	38.9-39.4 - Qtz patches of frags vs. .05", subangular, associated with			
	40.2 - Qtz			
	41.1 - 41.2 mt.			
	41.5-43.5 sugary matrix of Qtz, v. fine grained, in 2.5" Iron Form			
43.5-68.4	Matrix - fine gr. po. dy bands ocl to common. - zones of Fe (or Mn) rich - occasional zones of Mn rich Fe (or Mn) - calcification 15-60	53-54		0.013
	Matrix 41-43.5 not sampled - frags in core, such as LH 13 by 2.			
		68.4-72.4	1645	
72.4-74.1	10% mt, 7-15% sulfa, increasing down hole. to semi mass. mt/sulfa	72.4-74.1	1646	
	73.7-74.1 granular mt + po pn	74.1-77	47	
74.1-77	fine sulfa in mt from 74.1-74.2	77-81	48	
	74.1-77 coarse - fine grain sulfa, po, mt, pn? py?	81-86	49	
77-81	Matrix, 10% sulfa			
81-91.2	Um, bluish, talcy (soft) 2% diss sulfa, fractured (carb inclusion)	105.0-108		
	89-89.3 Qtz core vein	108-112		

-120 foot core matrix: less calcification, more massive, harder, weakly magnetic. MIN FRAGMENTS 105.5-112 3-5% SPICULOS po.

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

LANG

MAX. INCL 42°
 $\approx 30^{\circ}$

FACE

PROPERTY
 HOLE NUMBER 4414
 GRID REFERENCE - 3200N
 TOWNSHIP CLAIM
 AZIMUTH 225 DIP ANGLE $+30$

DRILLING COMPANY *Morrisette* FOREMAN
 CORE SIZE CORE STORED AT:

DIP TESTS:
 LOGGED BY *JAS.* DATE *Oct 24*

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0 - 71.3	Ulm.	0 - 5'	1569	0.31
	0 - 21 - Ulm, TALCY, GENERALLY $\approx 2\%$ FINELY DISS SULPH, UP TO 5% (POD - 60) Locally, MODERATE FRACTURING WITH CARB INFILLING	7 - 13	70	0.32
	5:7 - 2' GROUND.	13 - 17	71	0.33
		17 - 21	72	0.22
WEAKLY MAGN.	21 - 68 Ulm - HIGHLY CARBONATED, TALCOSE, NUMEROUS CARB FILLED FRACTURES (POD 60) $\approx 1\%$ FINELY DISS SULP, TRACE SULP BLEBS, MAJOR CARB FILLED FRACTURES @ 24.4 - .03'	21 - 26	73	0.27
		26 - 31	74	0.14
		31 - 36	75	0.28
		36 - 41	76	0.17
		41 - 46	77	0.09
		46 - 51	78	0.12
		51 - 56	79	0.12
		56 - 61	80	0.16
	63 - 68 CONTACT GROUND. - WEAK TO MOD MAGNETIC	61 - 68	81	0.13
	- Ulm - (FRAGILE?) MORE MASSIVE, LESS TALC, NO CARB, SOFT	68 - 70.7	82	1.08
-	68.7 - 71.3	70.7 - 71.3	83	2.09
	< 68.7 - 70.7 - 15-20% FINELY DISS SULPH MOD-STRONG MAGNETIC IN Ulm/MAGNETIC	71.3 - 73.8	84	0.02
	- 70.7 - 71.3 - MASSIVE MAGNETITE WITH PO, PY, PA, COY BLOB WITHIN			
	POSSIBLE GROUND CORE @ 71.3 - CONTACT GROUND			
71.3 - 72.6	Mafic! VERY CHLORITIC, SOFT			
72.6 - 93.0	POSSIBLE Tuff? BANDED - 5% MAGNETIC RICH BANDS DECREASING IN FREQUENCY DOWN HOLE. OCCASIONAL AMBYDYL (BARRAN. EXCEPT PARTIAL PY BAND @ 90.5'			
93.0 - 100	- DYKE. VFG TO MASSIVE, HARD, FELDSPATHIC			
100 - 113	Mafic! - CHLORITIC			
113	FOU.			

LANG #1

TIMMINS NICKEL INC.
DIAMOND DRILL LOG

PROPERTY LANGMOIR #1
HOLE NUMBER LH 15
GRID REFERENCE 3200
TOWNSHIP CLAIM
AZIMUTH 225° DIP ANGLE

lit H₂O @ 30'

DRILLING COMPANY *Morissa* FOREMAN
CORE SIZE *AQ* CORE STORED AT:

DIP TESTS: *16*
LOGGED BY *JAE* DATE *2-11-00*

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0-53.3	U _{in} , TALCOSE, MOD TO STRONGLY CARBONATED, MOD MAGNETIC	0-5	1585	0.55
	3-5% SULP OVERALL, FINELY DISS + SMALL PATCHES	5-10	6	0.38
	ABUNDANT CARB FILLED FRACTURES, HAIRLINE UP TO .1"	10-15	5	0.39
	17.7-17.9 CARB FILLED	15-20	8	0.32
	22.6-22.7 " "	20-25	9	0.30
	24.8-24.9 " "	25-30	90	0.30
	29.6-29.7	30-35	1	0.24
	36.2 SLIP - 1-2mm GOUGS, BECCATED. 20° TCA	35-40	7	0.14
	ONE 1/4 PIECE CORE 39-40 - SHEAR ZONE N15° TCA	40-45	1	0.28
	FOUR 1/4 54 - MAS MAGN + SULP 44-45 ~ 10% SULPH	45-50	6	0.24
	POSSIBLY ON OR PLACE PUT IN 52-53.3. 52-53.3 - 3% SULP BLESS popy in NON CARB U _{in}	50-52	5	0.24
53.3-55	U _{in} CHL MAFIC? NON MAGNETIC	52-53.3	6	0.09
	54.6-55 VERY COARSE CRYSTALLINE UNIT, MAGNETIC	53.3-55	7	0.05
		55-60	1598	0.01
55-75	U _a			
75	EOH			

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY *Lancaster #1*
 HOLE NUMBER *4416*
 GRID REFERENCE *3200W*
 TOWNSHIP
 AZIMUTH *090* CLAIM
270 DIP ANGLE
45

Hit H₂O @ 25 + 35'

DRILLING COMPANY

FOREMAN

DIP TESTS:

CORE SIZE

CORE STORED AT:

LOGGED BY *Jas*

DATE *Oct 26*

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
<i>0 - 72</i>	<i>Um (oxidation) BLUE GRACK, HOMOGENEOUS (MAGNETIC) 5-10% SULPHIDES (DSS), MAGNETIC</i>	<i>0-5</i>	<i>1599</i>	<i>0.45</i>
		<i>5-12</i>	<i>1600</i>	<i>0.26</i>
	<i>6-10 - VERY SHOWN CORE POSSIBLY SLIP/FAULT INDICATED BY</i>	<i>12-17</i>	<i>1601</i>	<i>0.16</i>
	<i>Q17 CARB UP TO 1" WITH PRECIPITATED UM FRACS.</i>	<i>17-22</i>	<i>1602</i>	<i>0.21</i>
	<i>17' - .1 Q17/CARB 32° TCA</i>	<i>22-27</i>	<i>1603</i>	<i>0.39</i>
	<i>19.2 - .25 " " 45° TCA</i>	<i>27-32</i>	<i>1604</i>	<i>0.19</i>
<i>H₂O?</i>	<i>22.6 - .1 " " RUSTY? 30° TCA.</i>	<i>32-37</i>	<i>1605</i>	<i>0.18</i>
	<i>27 .2 CARB TALL Q17 ~20° TCA SWEAT.</i>	<i>37-42</i>	<i>1606</i>	<i>0.94</i>
<i>H₂O?</i>	<i>28 .2 " " 38 " "</i>	<i>42-47</i>	<i>1607</i>	<i>1.63</i>
	<i>31-38 - 2% SULP.</i>	<i>47-52</i>	<i>1608</i>	<i>1.45</i>
<i>A</i>	<i>38-41 - 72 GRAY BLUE, GRAINY UM 3-5% SUMP. ~1-2% nt present</i>	<i>52-57</i>	<i>1609</i>	<i>1.30</i>
	<i>SLIGHTLY ^{MOAN} MAGNETIC TO STRONGLY MAGNETIC</i>	<i>57-62</i>	<i>1610</i>	<i>1.29</i>
		<i>62-67</i>	<i>1611</i>	<i>1.24</i>
		<i>67-72</i>	<i>1612</i>	<i>1.46</i>
<i>72</i>	<i>END</i>			
	<i>WORK STOPPED SHORT DUE TO EXCESS H₂O</i>			

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY *Timmins*
 HOLE NUMBER *2478*
 GRID REFERENCE *33000*
 TOWNSHIP _____ CLAIM
 AZIMUTH *090* DIP ANGLE *0*

DRILLING COMPANY
 CORE SIZE

FOREMAN

CORE STORED AT:

DIP TESTS:
 LOGGED BY

DATE

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
<u>2-6</u>	Um <i>40%</i> ~1% SULPH.	0-4	1659	0.14
	3' - 1' TALL/CARB FRACT 25°-CA	4-6	1660	0.20
	3.5 - 4.2 SOAPSTONE TALL FRACTURE?			
	4.2 - 4.5 5% SULPH.			
	4.5 - 7.6 BROKEN + RUBBLE CORE.	13-14.6	1631	0.02
	→ 5.6-5.65 MAS mt + po pr?	14.6-15.0	1632	0.20
	5.7-6.0 CHL	15.0-19	1633	0.28
6.0-14.6	- MAFIC → ULTRAMAFIC UNIT?			
	6.8 - 0.15 QTZ UN.			
	9.7 0.1 QTZ UN.			
	11-11.6 RUBBLES.			
	11.6-13 1.4" GROUND CORE			
	13-14.2 - VERY SILICEOUS, HARD.			
	14.2-14.6 - BECOMING CHL RICH			
<u>14.6-19.0</u>	Um - MORE CRYSTALLINE TEXT - POSSIBLE POOR SPINIFEX			
	14.6-14.8 - 90% MA MT, THIN SULPHID BAND AT BOTH WALL ROCK SIDES.			
	CONTACT WHERE 90° TCA.			
<u>19.0-26.6</u>	DKK: QTZ, F SPAN, A FEW POORLY DEVELOPED PORPHYRY, PINKISH ALONG FRACTURES.			
	19.0-20.2 ALYAGATION ZONE	38.5-40.5	1634	0.01
	19.7 - CHL + MT	40.5-45	1635	0.01
<u>26.6-38.5</u>	Um ?? WEAK CHL + CARB, WEAK LINATION OR BANDING (TUFF?) BARRER, NON-MAGNETIC			
<u>38.5-61</u>	CHL MAFIC ROCK, HAS A VERY FINE GRAINED TEXTURE, VERY MAGNETIC, SOME CARB.	45-50	1636	0.01
	45.6 .1 Q.V.	50-55	1637	0.06
	50.3-59 - 5% PATCHY SULPHID INCREASING TO ~10% FROM 55.5 → 58	55-59	1638	0.48
	mainly po, py, mt, some pr?, to cpy.	59-64	1639	0.08
<u>61-64</u>	possibly Um - BARRER.			
	60-61 GRADATIONAL CONTACT.			

64 EOH

TIMMINS NICKEL INC.

DIAMOND DRILL LOG

PAGE 1

PROPERTY *L'Anse-au-Loup*
 HOLE NUMBER *LM 19*
 GRID REFERENCE *3200*
 TOWNSHIP CLAIM
 AZIMUTH *090* DIP ANGLE

DRILLING COMPANY

FOREMAN

DIP TESTS:

CORE SIZE

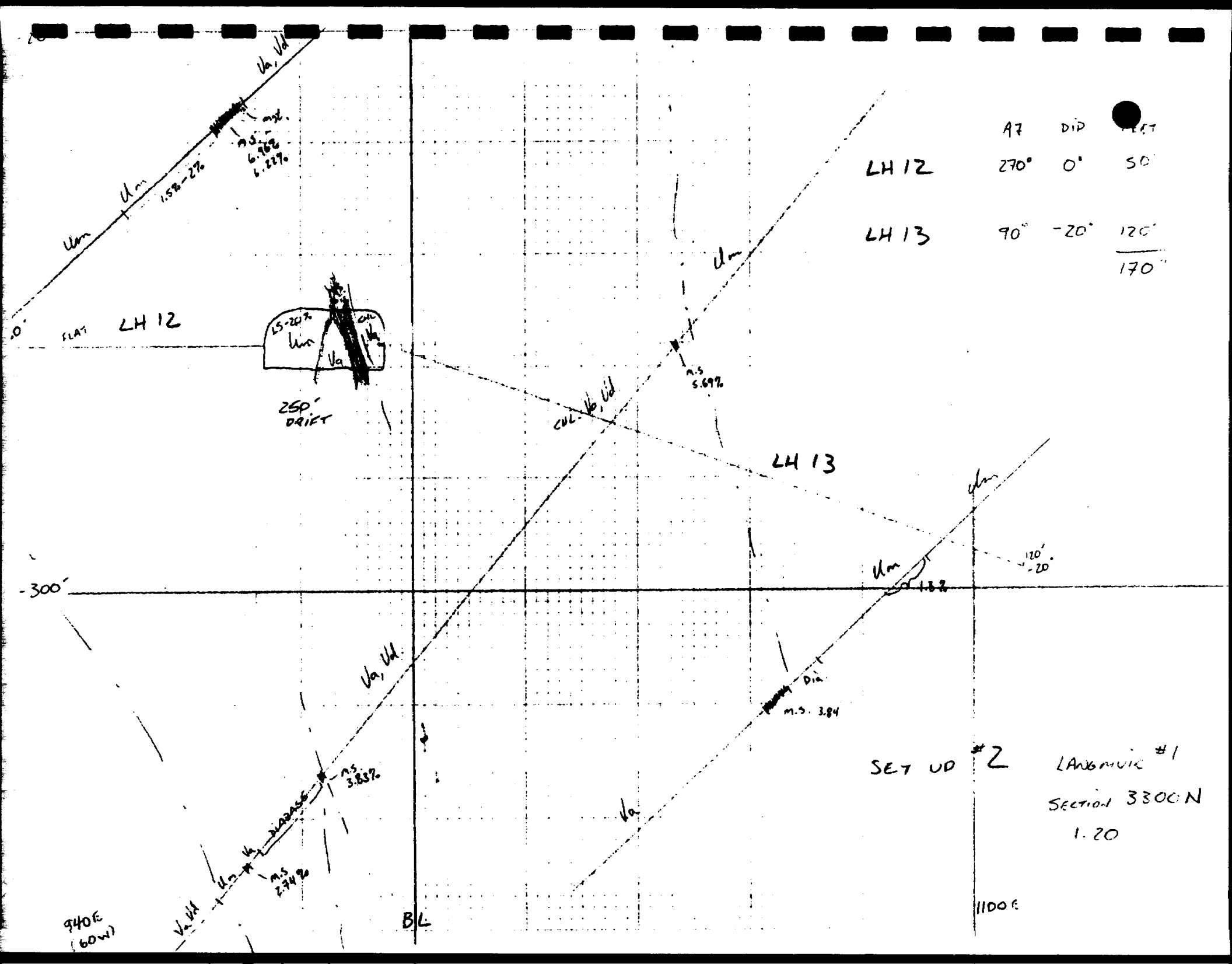
CORE STORED AT:

LOGGED BY *IAS*

DATE *Oct 76*

FOOTAGE	DESCRIPTION OF CORE	SAMPLE INTERVAL	SAMPLE NUMBER	ASSAYS
0-5.5	Ugn, BLuish, MOD TALEY 27% FINELY DISS SULF, WHARFY MAGN.	0-5.5	1613	0.20
5.5-23.3	VERY CHLORITIC IN CONTACT BECOMING LESS CHLORITIC	5.5-11	14	0.05
	9.5-10.5 - 2' GROUND. 6.5-8.5 - RUBBLE.	11-15	15	0.01
	VERY HARD CHEST LINE UGNI WITH 10-70% DISPY TO POORLY SANDED mt	15-20	16	0.01
	TO SEMI MASSIVE (60-60%) mt	20-23.3	17	0.01
	TO 23.3 - 23.3' CHLORITIC TOWARDS 20'-23'	23.3-27	18	2.06
		27-30	19	0.77
23.3-30	SUBSIDING MASSIVE, SEMI MASSIVE + 15% DISS IN Ugn, VERY MAGNETIC	30-34.5	20	0.03
	23.3-23.8 80% M.S. & 20% PY?, CPY			
	24.2-24.5 40% " " 20% PY			
	24.5-29.3 Ma mt.			
30-32	CHL RICH MAFIC ± 5% mt BANDING			
32-34.5	GREENISH FAIRLY SOFT Ugn??			
34.5-40.5	DYKE ^{FALSPATHIC} OCCASSIONAL PORPHYRITIC OR QTZ FYSS. MORE COMMON 42-49.5			
44.5-69	MAFIC? CHL CHLORITIC ZERRAN. MINOR SULPHIDE 62.5-67.55			
	65.4 .05' QTZ UGNI 50° TCA	63-65.4	21	0.01
	65.4-68 - ~5% SULPHIDE PY PO.	65.4-68	22	0.41
	68-69 - 41% SULF	68-69	23	0.11
69-	SULPHIDE ZONE	69-70	24	0.84
	69.0-69.3 - GOUGGY/FALCINTRO ZONE (POSSIBLY FAULT OR MACHINERY)	70-73	25	1.22
	69.3-69.7 - CHL Ugn? 5% SULPH	73-76.5	26	1.05
	69.7-70 MAS TO SEMI MAS SULF. PO PY MT? PY?	76.5-80	27	0.28
	70-79 30-10% FINELY DISS SULF.	80-85	28	0.29
	76.5 QTZ? FILLW FALT .07'			
79.4-	Ugn 5-10% SULF			

Full 1-21 (in core)



LANGMUIR #1
AS OF OCT 12.

DRILLING - SEE
SECTIONS

N1300' AQ
TO BEGIN MONDAY 15TH

2 SHIFTS
8-4
4-12

LH 14
F.S. LH 15

LH 16

F.S. SET UP #3

LH 17
LH 18
LH 19

B.S.
LH 14
LH 15

LH 14, 15. REVISED
OCT. 16.

LH 12

SET UP #2

LH 13

SET UP #1

LH 10
LH 11

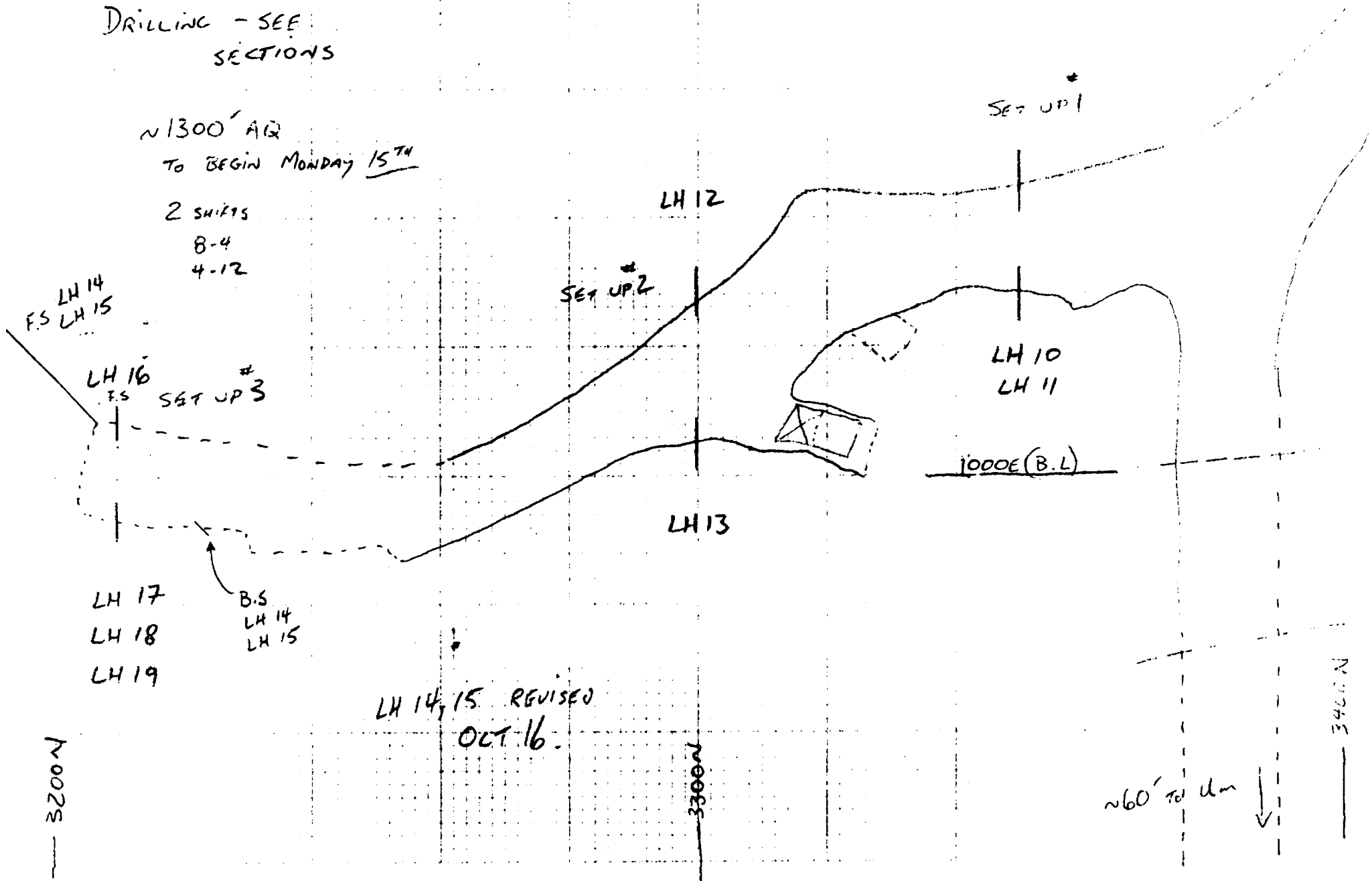
1000E (B.L.)

3200N

3300N

3400N

~60' to down



Appendix III

Underground Plans and Sections

Appendix IV
Metallurgical Studies

MEMO

LANGMUIR TEST RUN (Preliminary Report)

To: Don MacKinnon
From: Tony Lipiec

SUMMARY:

The test run on the Langmuir ore started at 2:30 a.m. on Saturday, October 6, and finished at 12:30 p.m. on October 9. Approximately 1250 wet tons were milled with an preliminary recovery of 81.5 % on a 1.80 % Ni feed grade. Concentrate grade produced was on the average in the range of 11 - 11.5 % Ni.

INTRODUCTION:

The purpose of this test run was to determine how well the Langmuir No. 1 ore would react in the flotation circuit presently used for the Redstone ore. In order to accomodate this test, it was arranged with Giant Yellowknife Mines to make room for this run by increasing at their expense the tonnage run through of the Redstone ore.

The test run was conducted by Tony Lipiec, the TNI production metallurgist, with the assistance of Mike Yu, metallurgical engineer with BHP-Utah Mines International. Extra manpower was employed to produce good quality sampling through out the day shift runs.

TEST OBSERVATIONS

Immediately as the run began, it became apparent that the muck was significantly different from the Redstone ore. The principal difference is the much higher pyrrhotite content in the Langmuir No. 1 ore. This was apparent both visually in the crushed muck, in the flotation froth, by the X-met samples, and by the lab assays. Unlike the Redstone, where the Fe:Ni ratio is approximately 2.5:1, the Langmuir ore is at 10.0:1. The consequences of this difference is the relative difficulty of making a high nickel grade concentrate, due to the problem of rejecting the pyrrhotite. Unlike the Redstone where all the sulphides are basically pulled into the concentrate, to do this with this ore would produce a Langmuir con under 8 % Ni.

After the problem was determined, it was decided to follow a flotation strategy different from the Redstone in that both the xanthate, frother and copper sulphate were cut back substantially. Keeping the latter two reagents at approximately half

the Redstone dosage allowed a significant increase in con grade to over 11 % Ni. The other change was the running of the floats sufficiently slow as not to overload the cleaning circuit. Tonnage was run at 330 tpd to determine the recovery at these dosages and grind. On the second day of the test it was decided to run the ore through at 410 tpd to determine the importance of grind. Recovery suffered at the higher throughput, decreasing from the low 80s to the high 70s.

Among other criteria looked at was how magnetic the various products were. Using a bar magnet both the nickel bearing minerals and the gangue were very magnetic. Another point, more positive, is the relative consistency of feed grade compared to the Redstone ore which fluctuates greatly. The Langmuir ore also is relatively easy to crush while in grinding a relatively fine grind will have to be employed. Environmentally, the Langmuir ore poses a problem because of its high pyrrhotite content. Although an acid generating test has not been done yet, it will likely be required before the MOE will allow its continued disposal into the ballpark.

FURTHER WORK

Further work should be done to both increase recovery and to improve the concentrate grade on the Langmuir ore. Sufficient samples were taken to aid in environmental work, X-met calibration models (for circuit control), mineralogical work, and some lab work. Possible methods to improve the circuit performance would be to try alternate collectors (pyrrhotite rejecting) and possibly pH control in the cleaning circuit.

CONCLUSIONS

At this point, the test run has produced some preliminary data and pointed out the need for improving certain aspects of the circuit. It also produced sufficient concentrate to allow Sherritt Gordon an idea of its refining characteristics.

From the way it was necessary to run the Langmuir flotation, it would be inappropriate to consider mixing the ore at any stage of flotation. It might be beneficial to the Redstone concentrate to mix it with Langmuir concentrate if the arsenic content proves to be low. It would certainly be beneficial to the Langmuir tailings to mix it with the more benign Redstone tailings.

REDSTONE JOINT VENTURE - METALLURGICAL BALANCE

October 6L

PRODUCTS	Weight Tons	Weight %	Assay % Ni	Pounds Ni	Dist Ni
FLOT FEED	189	100.0	1.77	6678	100.0
FLOT CONC	26	13.6	11.20	5732	85.8
FLOT TAIL	163	86.4	0.29	946	14.2

PRODUCTION FOR OCTOBER

FLOT FEED	189	100.0	1.77	6678	100.0
FLOT CONC	26	13.6	11.20	5732	85.8
FLOT TAIL	163	86.4	0.29	946	14.2

MOISTURE: 2.26 % H2O
FINAL CONC D/S 11.2 % Ni 25.4 % Fe
FINAL CONC N/S 11.2 % Ni 25.4 % Fe

REAGENT CONSUMPTIONS:

XANTHATE 0.40 lbs/T
COPPER SULFATE 0.40 lbs/T (CuSO4*5H2O)
GUARTEC 2.00 lbs/T

DOWN TIME 0.0 Hours

COMMENTS:

I. A. Lipiec

REDSTONE JOINT VENTURE - METALLURGICAL BALANCE

October 7L

PRODUCTS	Weight Tons	Weight %	Assay % Ni	Pounds Ni	Dist Ni
FLOT FEED	331	100.0	1.76	11665	100.0
FLOT CONC	43	13.1	11.10	9617	82.4
FLOT TAIL	288	86.9	0.36	2048	17.6

PRODUCTION FOR OCTOBER

FLOT FEED	520	100.0	1.76	18343	100.0
FLOT CONC	69	13.3	11.07	15349	83.7
FLOT TAIL	451	86.7	0.33	2994	16.3

MOISTURE:	2.26 % H ₂ O		
FINAL CONC D/S	11.0 % Ni	28.2 % Fe	
FINAL CONC N/S	11.2 % Ni	23.9 % Fe	

REAGENT CONSUMPTIONS:

XANTHATE	0.25 lbs/T
COPPER SULFATE	0.20 lbs/T (CuSO ₄ *5H ₂ O)
GUARTEC	0.50 lbs/T

DOWN TIME	0.0 Hours
-----------	-----------

COMMENTS:

I. A. Lipiec

REDSTONE JOINT VENTURE - METALLURGICAL BALANCE

October 8L

PRODUCTS	Weight Tons	Weight %	Assay % Ni	Pounds Ni	Dist Ni
FLOT FEED	412	100.0	1.86	15325	100.0
FLOT CONC	55	13.3	11.13	12214	79.7
FLOT TAIL	357	86.7	0.44	3111	20.3

PRODUCTION FOR OCTOBER

FLOT FEED	932	100.0	1.81	33668	100.0
FLOT CONC	124	13.3	11.12	27563	81.9
FLOT TAIL	808	86.7	0.38	6105	18.1

MOISTURE:	2.38 % H2O		
FINAL CONC D/S	11.7 % Ni	24.8 % Fe	
FINAL CONC N/S	10.7 % Ni	24.5 % Fe	

REAGENT CONSUMPTIONS:

XANTHATE	0.25 lbs/T	
COPPER SULFATE	0.20 lbs/T (CuSO4*5H2O)	
GUARTEC	0.50 lbs/T	

DOWN TIME	0.0 Hours
-----------	-----------

COMMENTS:

I. A. Lipiec

REDSTONE JOINT VENTURE - METALLURGICAL BALANCE

October 9L

PRODUCTS	Weight Tons	Weight %	Assay % Ni	Pounds Ni	Dist Ni
FLOT FEED	299	100.0	2.05	12267	100.0
FLOT CONC	38	12.7	12.87	9758	79.5
FLOT TAIL	261	87.3	0.48	2510	20.5

PRODUCTION FOR OCTOBER (LANGMUIR ONLY)

FLOT FEED	1231	100.0	1.87	45935	100.0
FLOT CONC	162	13.2	11.52	37321	81.2
FLOT TAIL	1069	86.8	0.40	8615	18.8

MOISTURE:	2.22 % H ₂ O	
FINAL CONC D/S	14.6 % Ni	26.4 % Fe
FINAL CONC M/S	11.6 % Ni	25.6 % Fe

REAGENT CONSUMPTIONS:

XANTHATE	0.25 lbs/T
COPPER SULFATE	0.20 lbs/T (CuSO ₄ *5H ₂ O)
GUARTEC	0.50 lbs/T

DOWN TIME 0.0 Hours

COMMENTS: The run finished 18 hrs into the day.

I. A. Lipiec

15.00
 30.00
 45.00
 11.00

LAKEFIELD RESEARCH

185 Concession Street
 Postal Bag 4300
 Lakefield, Ontario
 K8L 2J1 0

Facsimile No. (705) 652-6365
 Telephone No. (705) 652-3341

To: Mike Ross

Company: Timmins Nickel

From: Dave Evans

Fax No.: (705)264-2170

Date: 1990 11 28

Reference: LR#4075, Dome Mountain

This transmission consists of---8--- page(s) including this one.

Enclosed are the metallurgical results from Tests 5 and 6. Sorry for the delay but our XRF unit was down on Monday.

Also summarized is the preliminary results of the magnetic pre-concentration of Langmuir ore.

Product	Weight %	Assays, %		Distribution, %	
		NI	S	NI	S
Mags	25.5	-	7.00	-	77.6
Non-mags	74.5	-	0.69	-	22.4
Feed(calc)	100.0	-	2.30	-	100.0
Feed(assay)	-	1.33	2.19	-	-

Nickel assays will be available tomorrow.

Sue Parker has been in contact with Tony Lipiec about the flotation conditions, etc.

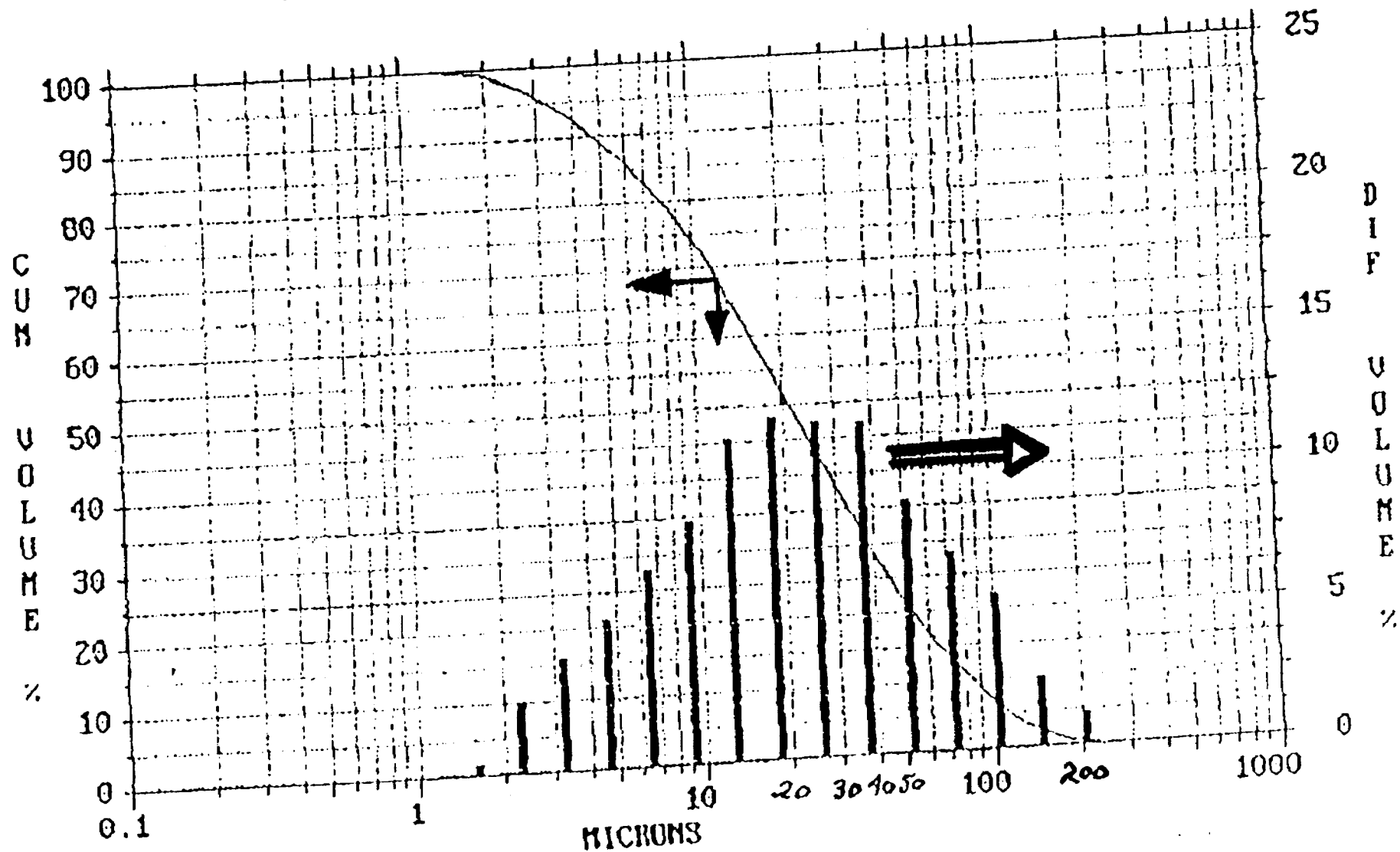
FROM SHERRITT GORDON - ANALYTICAL SERVICES

11.02.1990 10:11

NO. 11

P. 2

CM 137675 (07-90-01)
MT-785
Record Number: 550



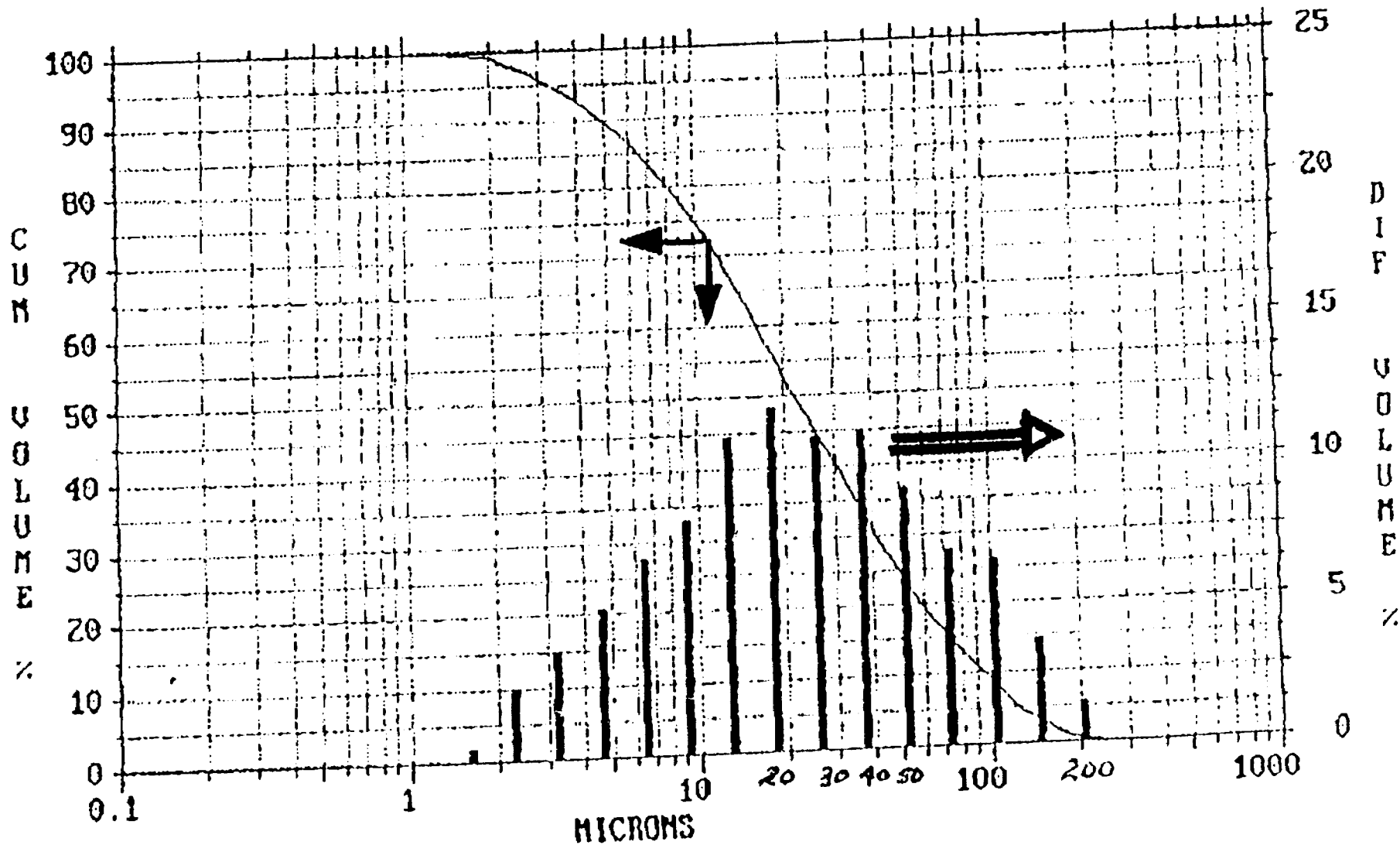
FROM SHERRITT GORDON - ANALYTICAL SERVICES

11.02.1990 19:12

NO. 11

P. 3

CN 136743 (07-90-02)
MT-786
Record Number: 554





November 11, 1990

Mr. S. McIntyre
Timmins Nickel, Inc.
Suite 205, 155 University Avenue
Toronto, Ontario M5H 3B7

Dear Steve:

I have attached a report from Mark Benz, our Refinery Production Superintendent, regarding the two trial railcars of Langmuir concentrate. As advised previously, we are already experiencing severe problems due to the high Mg content of your regular production and cannot consider any action that might further disrupt an already serious situation.

I am therefore confirming our previous advise that we are rejecting the two railcars. I realize that this material is not necessarily representative of the concentrate that would be produced from Langmuir on a production basis. Hopefully after your current milling problems are resolved, a more suitable concentrate would be available.

We have contacted Inco and requested that they consider the purchase of this material.

Yours truly,

Brian

B.J. Shea
Manager, Metals Marketing
Metals Division
Sherritt Gordon Limited

sherritt

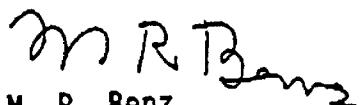
inter-office letter	copies to
date November 5, 1990	RMGarvey Tor. MDDay #11 PAmarnath #99 RSabourin #57 Gingram-Johnson #57 File: Timmins Nickel - Tech
to B. J. Shea	
from M. R. Benz	
subject TIMMINS NICKEL - LANGMUIR CONCENTRATE	

Please find attached the physical, chemical and moisture analyses of Lots 07-90-01, 02 (CN cars 137675 and 136743). Also attached are photographs of the material. The cars are currently still at Fort Saskatchewan and are not yet unloaded.

Unfortunately, we must reject these cars for the following reasons.

1. High Moisture Content. These cars are at 14.7 and 16.2% H₂O. The material in both cars has the consistency of gumbo (see attached photographs). Only one car could even be considered for feeding, and then only in a "teaspoon" fashion. Thus piloting these cars is out of the question. In the winter these cars would freeze and be impossible to unload and to sample representatively.
2. Fines Analysis. The medium particle size in these cars is only about 20 microns. In fact, only about 20% is plus 200 mesh. This probably contributes to the consistency of the material. Feeding this material with its high Mg content (almost 6%) will cause solid-liquid separation problems in our primary leach circuit. This would seriously risk the copper boil heat exchanger to scaling, causing complete production interruption as well as sliming of the copper sulphide. The copper sulphide could well become so wet as to cause severe problems in shipping this important by-product.

Note that the low nickel grade, high sulphur content and high Mg content make this a very marginal concentrate to treat, even without the above problems.


M. R. Benz
Production Superintendent,
Metals Refinery

MRB*bp
Attachment

LAKEFIELD RESEARCH

185 Concession Street
P.O. Bag 4300
Lakefield, Ontario
KOL 2H0

Facsimile No. 705-652-6365
Telephone No. (705) 652-3341

To: Tony Lipiec and Coos Schippers

Company: Timmins Nickel

From: Sue Parker

Fax No.: 705 268 0455 and 416 367 8965

Date: Nov 30, 1990

Reference: 4082

This transmission consists of 1 page including this one.

One 10 kilogram charge of minus 1 inch Langmuir Ore was crushed to minus 10 mesh and a head sample riffled out. The sample confirms the low head grade. The % Ni was 0.84. The original head analyses on the minus 1 inch material will be deducted from your account.

We will repeat the testwork with the new sample as soon as it arrives.

LANGMUIR ORE PROCESS CHART

Feed - crushed to
450 tpd - $\frac{7}{8}$ "

Rougher
Cobber

↓ Conc

Cleaner
Cobber

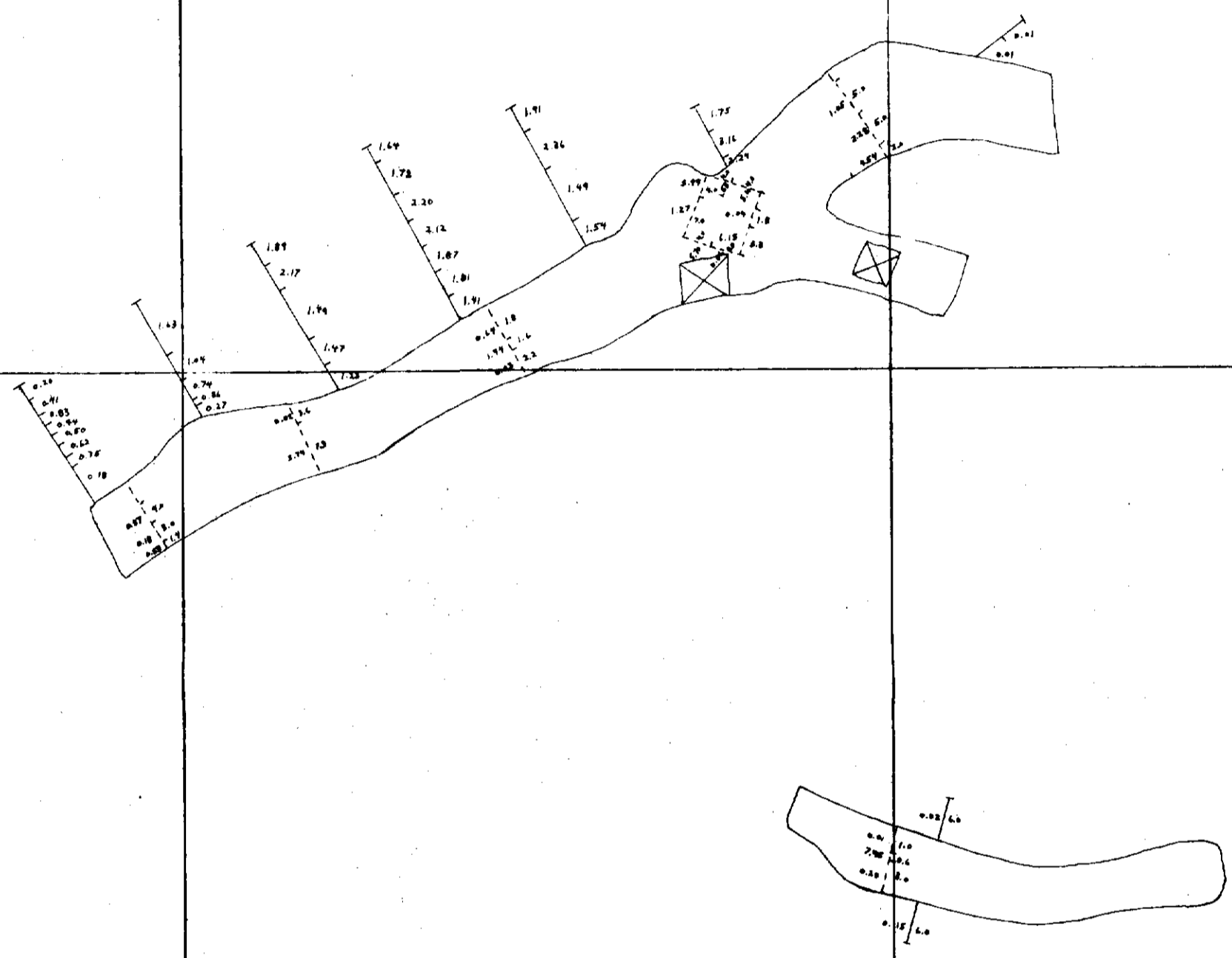
150 tpd.
Waste reject
0.25% Ni.

150 tpd. V
Middling
1.0-1.5% Ni
to Mill for
Flotation

Final Conc 150 tpd
3.5-4.0% Ni shipped
to Fairbridge.

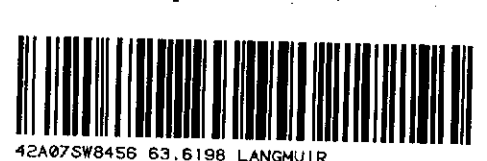
Flotation
Mill
batch process

15 tpd Conc @ 15% Ni
shipped to Sherritt Gordon

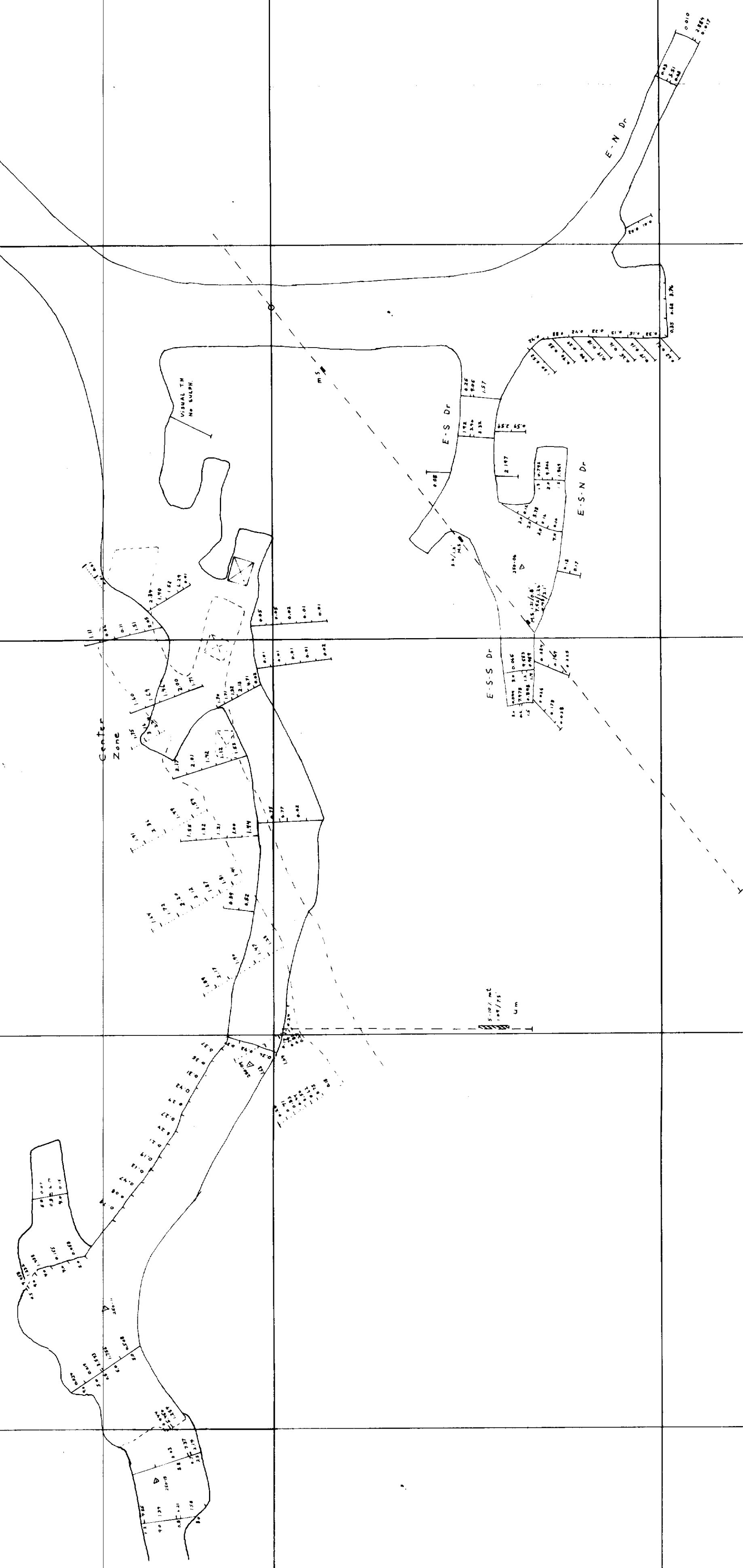


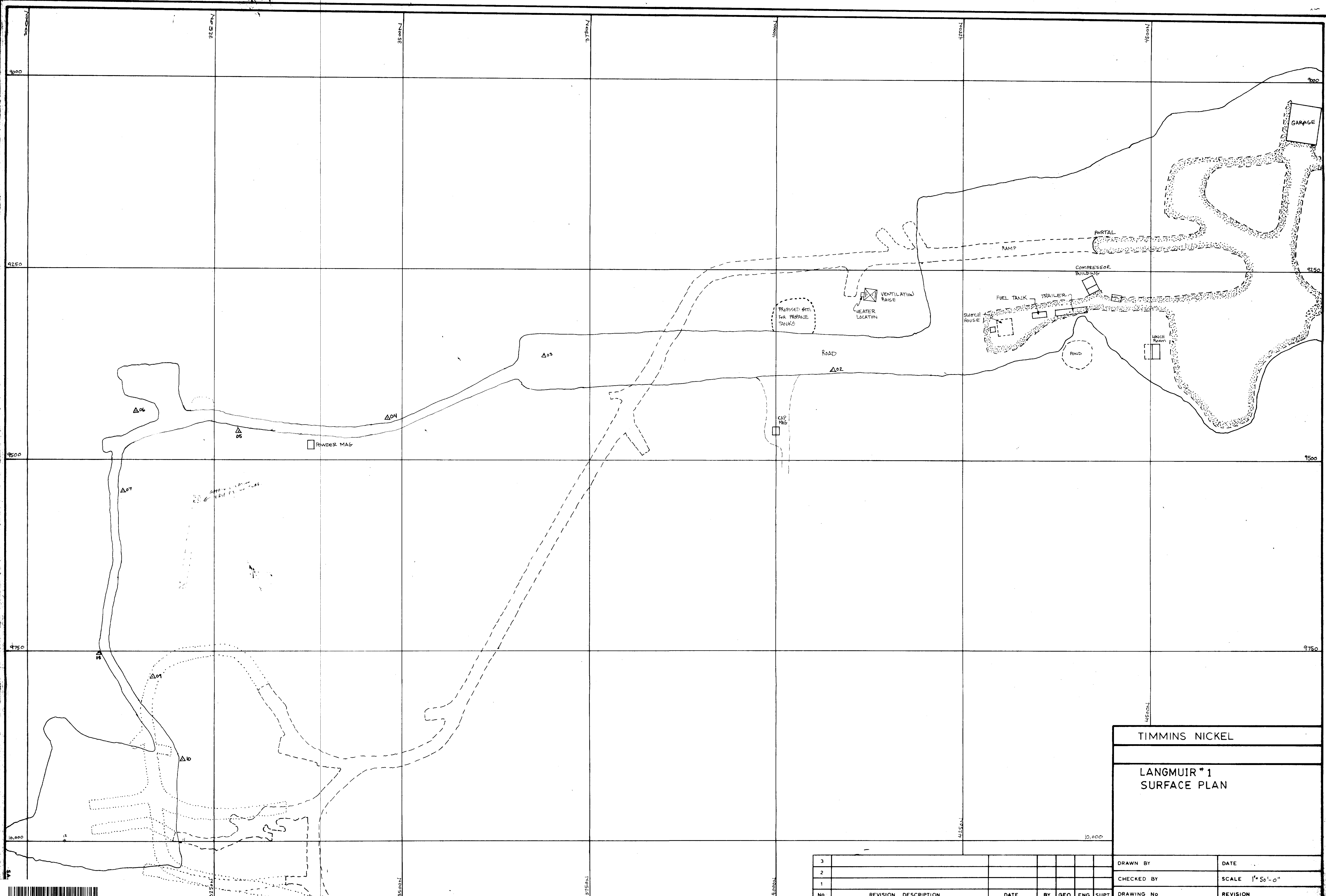
LANGMUIR ASSAY PLAN
200 SUBLEVEL

70-1000 Jan 2/41



LANGMUIR
ASSAY PLAN
250 LEVEL





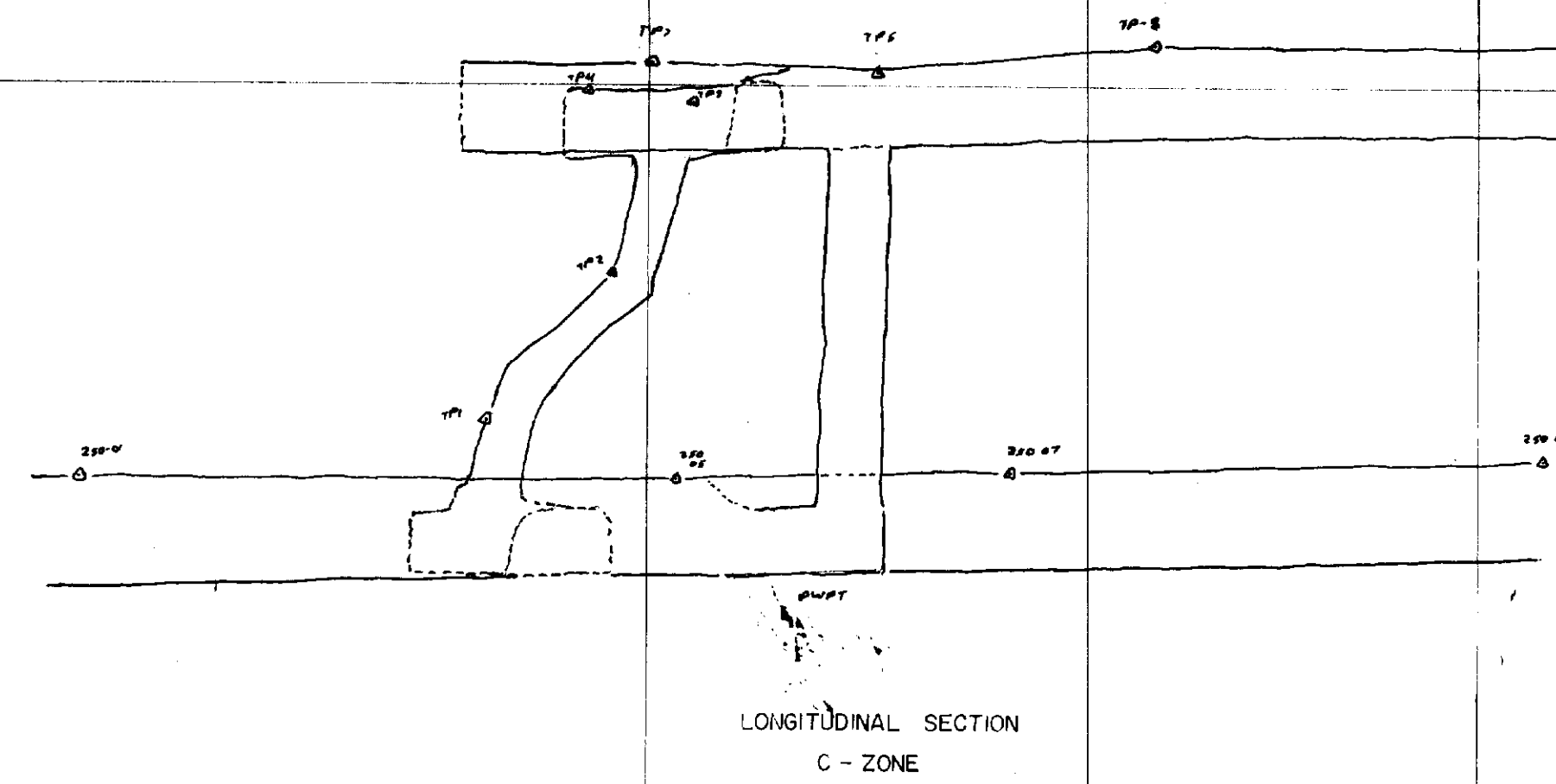
TIMMINS NICKEL

LANGMUIR #1
SURFACE PLAN

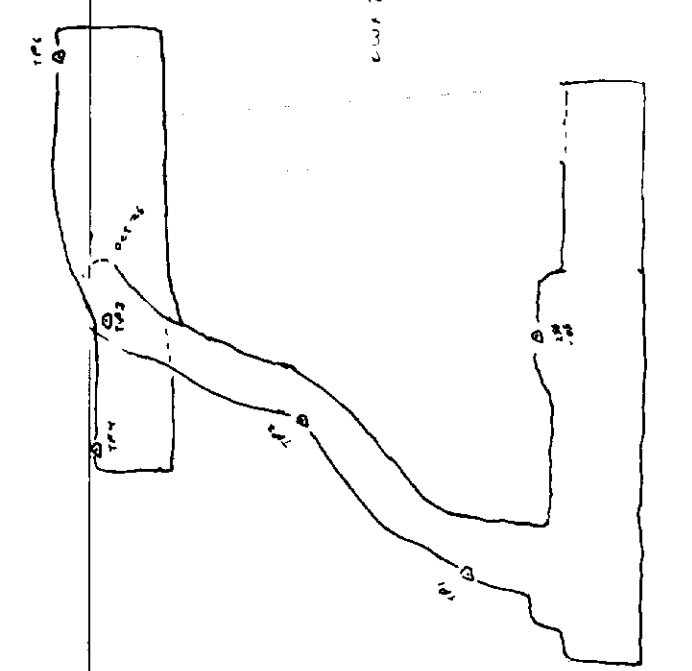
NO	REVISION	DESCRIPTION	DATE	BY	GEO	ENG	SUPT
3							
2							
1							

DRAWN BY	DATE
CHECKED BY	SCALE 1" = 50'-0"
DRAWING No	REVISION

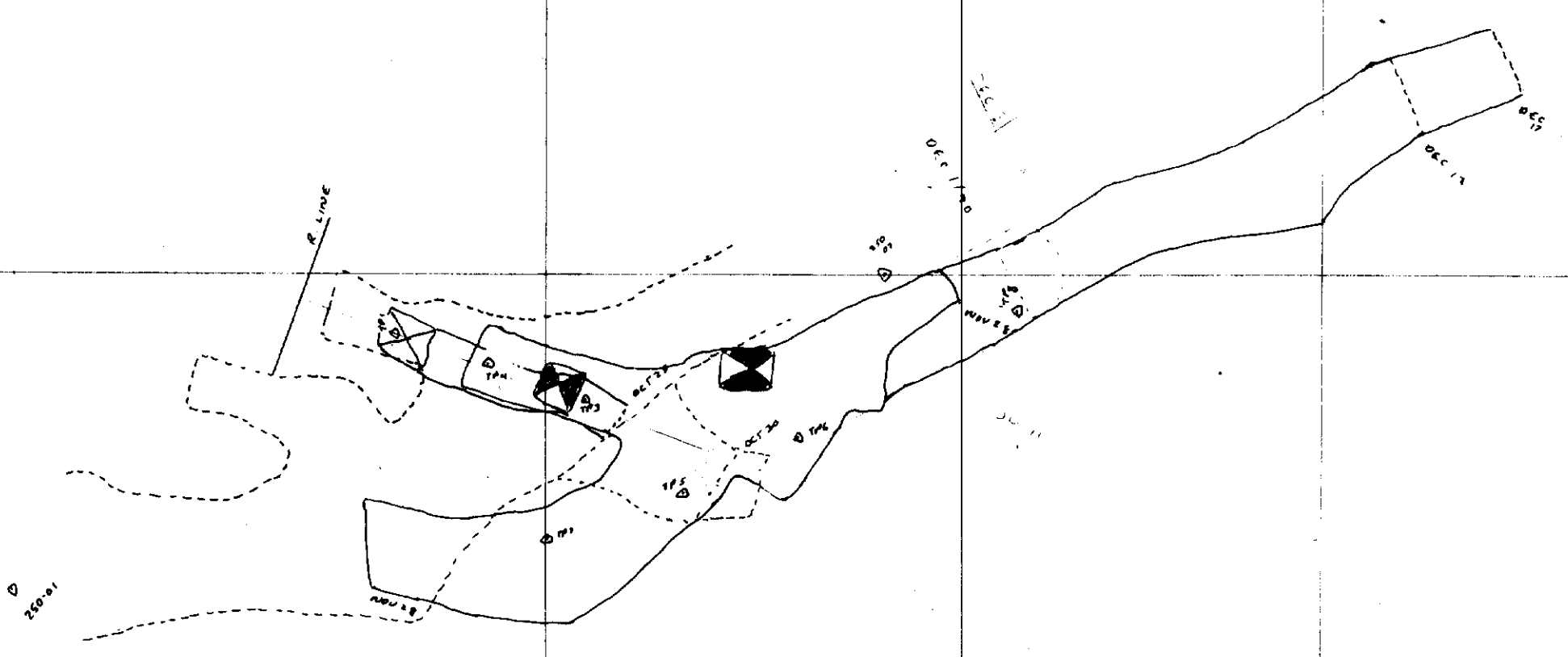




LONGITUDINAL SECTION
C - ZONE



ONE SIDE VIEW SECTION
LOOKING EAST
3200 C - 47 / 456



200 C - ZONE SUB DR.
LANGMUIR
PLAN VIEW

TIMMINS NICKEL-BHP UTAH
REDSTONE PROJECT

LANGMUIR CPE RAISE

No	REVISION	DESCRIPTION	DATE	BY	GEO	ENG	SUPT	DRAWING No	REVISION
3									
2									
1									

