PROGRESS REFORT

SCRANTON MINE EXPLORATION & DEVELOPMENT

PERIOD 1 MARCH - 22 JULY 1978

David Minerals.

WALTER E. CLARKE, B.Sc., P. Eng.

25 July, 1978

To The President & Directors, David Minerals Ltd. (N.P.L.) 315 - 850 West Hastings St., Vancouver, B.C. V60 121

PROGRESS REPORT SCRANTON MINE EXPLORATION & DEVELOPMENT PERIOD 1 MARCH - 22 JULY 1978

INTRODUCTION

The following report summarizes the work performed during the period at the Scranton Mine property, in general following the recommendations in the writer's "Proposed Exploration & Development Program Report" dated 25 October, 1977.

In order to utilize available funds most effectively, portions of both Phase I and Phase II have been completed, the main objective being exploration for the downward extension of the Grandview/Sunrise Basin vein structures by drifting, cross-cutting and diamond drilling on the 5700 Level.

The property has been visited by the writer on a regular basis throughout the period, the latest examination being 20-22 July, 1978.

Attached maps are 1" = 20' scale Geology Plans and a 1" = 200' Composite Plan.

SUMMARY

Drifting southwesterly was commenced 19 March 1978 from the previous south face of the 5700 Level drift, approximately 1,000 feet from the portal. The drift has been advanced 727 feet, and for the last 270 feet has been within a zone of strong multiple shearing and faulting, alteration, mineralization and narrow quartz veining, believed to be the downward extension of the Grandview/Sunrise Basin vein structures.

Diamond drill stations have been cut at approximately 150 foot intervals along the drift and the 8300 crosscut has been extended 100 feet into the west wall, near the south face of the drift, as a drill base to permit diamond drill exploration of the zone above the level. The total drifting, cross-cutting and equivalent slashing footage is 958. Diamond drilling amounts to 1520 feet, with 9 holes completed and the 10th and last hole of the current program in progress.

The present south face of the drift is about 160 feet within the projected target area of the Grandview/Sunrise Basin zone and with the exception of diamond drill holes 78-7 and 78-9, another 300 feet to the southwest remains unexplored.

Funds available for the program amounted to \$251,952.00, of which approximately \$198,000.00 were spent on property operations to 30 June 1978, and \$44,618.00 on related administrative expenses for a total of \$242,618.00. Total direct and indirect property costs for drifting and cross-cutting were \$187.00/foot of advance, and for diamond drilling \$16.96/foot.

Economic mineralization has not been encountered in the drifting, although several narrow sheared quartz veins, cutting the drift at low angles, were mineralized with pyrite, galena and sphalerite, and returned interesting values in gold, silver, lead and zinc. Investigation of these veins outside the drift section is warranted. The first 6 diamond drill holes checked for any parallelling structures up to 200 feet in the east and west walls over a distance of about 500 feet. A strong altered shear zone was intersected in the last 28 feet of hole 78-6, 136 feet east of the drift, but values were low, as were assays on lesser intersections in the

Hole 78-7, drilled from 8300 crosscut encountered the main other 5 holes. structure 130 feet south of the present face and, within the 7.0 foot true width of zone, intersected a very well mineralized 3.1 foot core length (2.2 foot true width) which assayed 0.38 ounces/ton gold, 11.20 ounces/ton silver, 15.705/ton lead and 20.60, ton zinc. Hole 78-8 was directed to intersect the zone about 150 feet above the level on line with the crosscut, but poor ground conditions prevented completion of the hole and the main structure was not reached. Hole 78-9 intersected the main zone approximately 90 feet above hole 78-7. The structure is strong across a comparable width, with scattered pyrite, galena and sphalerite, but all assays were low. Hole 78-10 is currently being drilled to check the structure approximately 100 feet north of the crosscut and 150 feet above the level. completion of hole 78-10, it is planned to resume drifting southwesterly to open up the vein intersected in hole 78-7.

CONCLUSIONS

work to date has demonstrated the presence of a strong multiple stranded shear and fault system on the 5700 Level, which is presumed to be part of the regional structure that has been intermittently identified on surface for over 7000 feet. In the vertical plane, mineralization is now known to extend at least 800 feet below surface, with important gold and silver values associated with lead and zinc, a prerequisite for an economic operation on a narrow vein system. The attitude and continuity of individual fault strands and their control over mineralization is not clear. So ore shoot has as yet been developed, but opening up of the strong mineralization in diamond drill hole 78-7 by drifting, initially, is of the

utmost immediate importance to help determine mineralization controls and continuity. The results of this work will have a major influence on the conduct of future exploration in this area.

DRIFTING

A line drive \$550W was commenced from the south face of the 5700 Level. designed to intersect the downward extension of the Grandview/Sunset Basin vein structures. For the first 470 feet the heading was in relatively fresh granodiorite, moderately sheared and fractured, with some of the fractures quartz filled and sparsely mineralized. A marked increase in the fracture pattern from 300 to 425 feet suggested proximity to a major structure. At the 470 foot mark, a strong zone of shearing, faulting and accompanying alteration of the granodiorite entered the drift at a low angle from the west wall, dipping 650 easterly, and remained in the drift for approximately 120 feet, at which point it was terminated at a strong east-west fault, dipping 200 to the south. The strongest shearing varied from one to four feet wide, within an overall zone width of alteration in excess of Quartz veining and sulphide mineralization were variable, but in places amounted to 30 to 40% pyrite and lesser quantities of galena and sphalerite. The best assays were obtained from samples at the junction of the shear zone with the cross-fault and 25 feet north of the junction. The former assayed 0.078 ounces/ton gold, 38.10 ounces/ton silver, 0.28%/ton lead and 0.14%/ton zinc, while the latter assayed 0.13 ounces/ton gold, 30.42 ounces/ton silver, 0.50 /ton lead and 0.23%/ton zinc. For a 30 foot length and 2.4 foot width the average assays from four chip samples were 0.079 ounces/ton gold and 18.5 ounces/ton silver with low lead and zinc.

Beyond the cross-fault relatively fresh granite was encountered for the next 50 feet, at which point a strong steeply dipping shear and fault structure entered the drift section, again from the west wall, which gradually angled across the drift so that at the present face, 100 feet from the point of entry, it has almost passed out of the drift in the east wall. Within the zone, two mineralized quartz-filled shears, dipping 60° to 65° easterly, and striking across the drift at an angle of 20°, were exposed. The best assay on the first vein was 0.22 ounces/ton gold, 3.57 ounces/ton silver, 3.90%/ton lead and 0.03%/ton zinc, over a width of 0.5 feet, and on the second vein 0.12 ounces/ton gold, 3.20 ounces/ton silver, 16.65%/ton lead and 0.06%/ton zinc. Additional exploration on these shears into both walls and in the vertical plane is warranted.

In 8300 crosscut, a prominent shear was exposed after 18 feet of advance with a mineralized quartz filled shear a further 5 feet west. The latter vein shear assayed 0.15 ounces/ton gold, 2.00 ounces/ton silver and 2.24%/ton lead and may be related to the intersection in diamond drill hole 78-7.

DIAMOND DRILLING

The following tabulation summarizes the diamond drilling to date.

D.D.H. No.	FOOTAGE	LATITUDE	CATION E DEPAR	TURE	BEARING .	DĪÞ	REMARKS
78-1 78-2 78-3	209.0 190.0 186.0	9101 N 9124 N 9037 !	8781 8770 8547	E	\$35°11'E N24°13'W N36°30'W	+00491	Nothing of interest. Nothing of interest. Altered zone to 30 feet. Main Zone? No values.
78-4 78-5 78-6	167.0 95.0 164.0	8946 N 8924 N 8822 N	8523 8533 8392	E	N31°41'W \$21°01'E \$30°51'E	+1°23'	Nothing of interest. Weak zone 68-74. No values. Shear zone at 82 and 125 feet Alteration & fault zone 136 to 164 feet. No values.
78-7	203.0	8948 N	8249	E	S15°22'W	+0°51'	Main zone 153-166 feet. See
78-8	106.0	8851 N	8252	E	539 ⁰ 49'E	+66 ⁰ 291	Fault zone 94-106. Hole abandoned. Main zone not reached. No values.
78-9	200.0	8850 N	8249	E	211°38 'W	+46009	Main zone 111-121 feet. No values.
Total :	1520.0	(a) 32	ai a		•	*	
D.D.H.	78-7			ASSA	YS		REMARKS
Fox	otage	Cono	Gold	Silv	er Lead	Zinc	
From	Fo	Core Length	oz/ton	oz/t	on $3/to$	n 3/ton	
153.2 154.5 157.6 154.3	157.6 164.3	1.3 3.1 6.7 1.5	0.015 0.380 0.036 0.009	0.2 11.2 0.4 0.3	0 15.70 1 N/A	0 20.60 N/A	True width 2.2 feet 7.5% core recovery
	2000		0000	, 00)	, 0.0	0.71	

12.6 - True width 7.0 feet

COSTS

PROPERTY OPERATING

	DRIFTING &	CROSS_CUTT]	ING DIAMOND	DRILLING	TOTAL
				**************************************	EXPENDITURE
Footage for		18		34	\$
period	95	8	11	14	
por rou					
es es .	EXPENDITURE	COST/FOOT	EXPENDITURE	COST/FOOT	
	\$	\$	\$	\$	4) 4
	*			•	
Direct					
Contract	87.709.00	91.55	11,247.00	10.10	98.956.00
Cost Plus	0/1/0/000	72477	11,20	20120	, , , , , , , o o o
Labor	20,205.00	21.09			20,205.00
Materials	15,069.00	15.73			15,069.00
Overhead	7.471.00	7.80			7,471.00
Sub Total Cost Plus	42.745.00	44.62		14	7,471.00
	643.00	0.67	592.00	0.53	7 225 00
Assaying	043.00	0.07	792.00	0.53	1,235.00
TOTAL DIRECT	131,097.00	136.84	11,839.00	10.63	142,936.00
TOTAL DIRECT	1)1,097.00	1)0.04	11,000,000	10.37	172,950.00
Indirect					
Room & Board	3.300.00	3.45	484.00	0.44	3.784.00
	17,219.00	17.97	2,528.00	2.27	19,747.00
Power & Air	17,219.00	14.71	2,520.00	2021	17,14/.00
Transportation,	77 70/ 00	11.67	1.641.00	1.47	10 010 00
Equipment, Rentals	11,176.00	4.93	693.00	0.62	12,817.00
Repairs	4,725.00				5,418.00
Insurance, Taxes	11,631.00	12.14	1,707.00	1.53	13.338.00
TOTAL INDIRECT	48,051.00	50.16	7,053.00	6.33	55,104.00
TOTAL INDICATE	.0,0,100	70020	11000		
TOTAL PROPERTY					
OPERATING	179,148.00	187.00	18,892.00	16.96	193,040.00
OI BICKI III	2//12/000				=
ADMINISTRATIVE		g ³⁰			
EXPENSES		e e			44,618.00
ZAI BRODO			•		

The above property operating costs are about 40% higher than estimated in the writer's October 1977 report. The most unpredictable cost item is in the area of "cost plus" work and this was higher than anticipated due to heavy (ground conditions

in the latter 250 feet of advance which required extensive back and wall support. Further cost escalation was in the negotiated contract price per foot for drifting and cross-cutting, partially as the result of inflationary pressures in labour and material costs. For similar reasons indirect costs for power and air were higher than anticipated. Breakdowns increased the equipment repair and rental items and mill insurance had to be borne by this phase of the operation.

Respectfully submitted,

Willer Gillark

WALTER E. CLARKE, B.Sc., P. Eng.

25 July, 1978.

CERTIFICATE

- I. Walter Ernest Clarke, of the City of Victoria, British Columbia, do hereby certify that:
- 1. I am a consulting goological and mining engineer with an office at 1362 Dallas Road, Victoria, British Columbia. V8S 1AL.
- 2. I am a graduate of Queen's University (1939) with a B.Sc. degree in Goology and Mineralogy.
- 3. I have practiced my profession continuously since graduation.
- 4. I am a member in good standing of the Association of Professional Engineers in the Provinces of British Columbia and Ontario.
- 5. I have no interest, either direct or indirect, in the properties or securities of David Minerals Ltd. (N.P.L.), nor do I expect to acquire any such interest in the future.

Walter Shelork

WALTER ERNEST CLARKE, P. Eng.

July 1978

.	To:	Dav.	Minerals	Ltd.
1	PAGE	No	11	

BONDAR-CLEGG & COMPANY LTD.

DATE: June 6, 1978

315 - 850 West Hastings Street Vancouver, B.C. V6C 1E1

CERTIFICATE OF ASSAY

Samples submitted: May 29, 1978 Results completed: June 6, 1978

I hereby certify that the following are the results of assays made by us upon the herein described ore / core sample

MARKED	GC	LD	SILVER	Cu	Рь	Zn					TOTAL VALU
9	Ounces per Ton	Value per Ton	Ounces per Ton	Percent	PER TON (2000 LBS.)						
10718*	0.075		0.65	-	0.05	0.02	DUFA	NGE 42	AMPLES	*	1-30'
10719 🟲	0.035		0.65	-	-	-					111-2.4
10720	0.002		0.68	-	-	-		TH TT		6	10-0.0790 WER-18.50.
10721 ^K	0.078		38.10	0.04	0.28	0.14	200	57-23	711011	21.	KIER- 18.50.
10722	0.010		0.98	-	-	-		27-23			
10723	0.052		0.90	<0.01	0.05	0.01					54
10724	0.003		0.49	-	-	-					
10725	0.002		0.81	-		-	6				
10726*	0.13		30.42	-	0.59	0.23	-				
. 10727	0.056		1.75	-	-	-					
10728	0.011		0.47	-	-	-		100			
10729	0.064	1	0.80	<0.01	0.10	0.04	(±)				
10826	0.004		0.03	-	-	-					7.0
10827	<0.002		0.08	-	-	_					
10828	<0.002			-	_	_		-			
10829	<0.002		0.05	-	-	-					
10830	<0.002	154	0.05	-	-	-					
10831	<0.002 <0.002		0.06	_							
10832 10833	<0.002	0	0.04			-					
7.0										i	
10834.	<0.002		0.02	-	7	-					
10835	0.006		0.23	_]				1	* "
10836 10837	0.002		0.07	-	_	_					
10837	0.003		0.07	-	-	-					
10839	<0.002		0.02	-	-	-					

To:	David	Minerals
DAGE	No	1

Vancouver, B.C.

315 - 850 West Hastings tigreet

BONDAR-CLEGG & COMPANY LTD.

DATE: June 9, 1978

REPORT No A28 - 244

Samples submitted: June 6, 1978

CERTIFICATE OF ASSAY

Results completed: June 9, 1978

I hereby certify that the following are the results of assays made by us upon the herein described

samples

MARKED	GC)LD	SILVER	Pb	Zn						TOTAL	
	Ounces per Ton	Value per Ton	Ounces per Ton	Percent	Percent	Percent	Percent	Percent	Percent	Percent		TON LBS:)
10730	0.026		3.32	•	•							
10731	0.16		0.47	•	•	25	EAT JO	074/ 37	MON	57-25		
10732	0.086		0.55	•	•							
10733	0.030		0.32	-	•							
10734	0.22		3.57	3.90	0.03	35/	CERT 30	oTH 2	MOTTON	57.75		
10847	0.003	•	0.38							3.2		
10848	0.006		0.18	•	•			500	,3 7 19	3		
10849	<0.002	9	0.06		•			2	75		.TI	
.0 - 78 - 4	0.023		0.32	0.03	0.04				7.			
				*							8	
										*		
									0 N2			
cc Mr. W. Clarke												

# 80 25								* .				
		<u>.</u>										
**************************************	th										2 1 2	

To:	Davis	Minerals
17/		

PAGE No. _

BONDAR-CLEGG & COMPANY LTD.

DATE: June 22, 1978

315 - 850 West Hasings Street Vancouver, B.C.

CERTIFICATE OF ASSAY

Samples submitted: June 16, 1978

Results completed: June 22, 1978

V6C 1E1

I hereby certify that the following are the results of assays made by us upon the herein described ore/core sample

MARKED	GC	DLD	SILVER	Cu	Pb	Zn				e)	TOTAL VA
	Ounces per Ton	Value per Ton	Ounces per Ton	Percent	(2000 LBS						
10705 6	0.004		0.14								
10735 -	0.004		0.14	•							
10736	0.002		0.24	<0.01	0.02	0.04					
10737 ~ 10738 ~	0.010		0.34	<0.01	0.14	0.08					
107392	0.009		3.20	-	16.65	0.06	62 F	EFT JOO	77 27	4 TON	57-25
10740	0.058		0.39	-	0.72	0.30	8 6				
10741	0.004		0.03	<0.01	0.07	-					
10742 × 10743 ·	0.010 0.21		0.19 0.53	-	_	-	11 7	ET JO	TH 77	MONT	57-25
10744	0.014		0.08		-	-					
10851 '	<0.002		0.02	-	-	-		x *			
10852 :	<0.002		0.02	-	-	-			15 3 1	v: 30	Control of the Contro
10853	<0.002		0.05	-	-	-			·		
10854 [*] 10855 [*]	<0.002 0.002		0.02	-	-					-	
10856	0.004		0.26	- 4	-	-					
10857	<0.002		0.05	<0.01	-	-					
10859	0.002		0.05	-	~	-		83.2			
			1						*	• •	
cc Mr. W. Clarke										**	
										2	

To:	D-via	Marala	Ltd.
	••••		

PAGE No. 1

BONDAR-CLEGG & COMPANY LTD.

REPORT No A28 - 3...

DATE: July 14, 1973

Samples submitted: July 11, 1973 Results completed: July 14, 1973

315 - 850 West Hastings Street Vancouver, B. C.

Vancouver, B. C. V&C 1D1

CERTIFICATE OF ASSAY

I hereby certify that the following are the results of assays made by us upon the herein described ore / core sample

	MARKED	GO	LD	SILVER	Pb	Zn						TOTAL VALU
	· .	Ounces per Ton	Value per Ton	Ounces per Ton	Percent	Percent	Percent	Percent	Percent	Percent	Percent	PER TON (2000 LBS.)
	10760	0.15		2.00	2.24	49	8300	X-607	2.0 F	ET END	274770	N 57-27.
	10761	0.005		0.06	•	-		% .				
	10762	<0.002		0.09	67	0.01						
	10763	<0.002		0.02	•							
	10764	0.002		0.02	•	0.04						
	10765	<0.002		<0.02		0.01		*				
	10766	0.045		0.14	•	63						
	10767	0.003		0.10	0.01							
	10763	<0.002		0.04	• .	0.01	8 I ₂					
	10769	0.002		0.17	•	0.01						
	10366	<0.002		0.04	•	-				•		
	10366	0.002		0.12	. 60		1					
	10363	<0.002		0.02	•	-						
	10869	<0.002		0.02	-	•						
	10870	<0.002		0.03	•	•						
		<0.002		0.02			5					
(4)	10371 10372	<0.002		0.02		•						
	10872	0.015		0.23	0.13	0.10			2.D.H.	78-7		
	10373	0.38		11.20	15.70	20.50	VEIN	ZONE	V. U. H.	, ,		
	10875	0.036		0.41		-						
		1 1	÷.	0.39	0.80	0.84						
	10376	0.009		0.03	0.12							
	10377	0.003		0.85	0.96	1.38						
	10378	0.020		0.03	0.70	20,00				1 1		
				3	18.							
cc lir.	W. Clarke				,		,					