Report on the 2002 Check Sampling Program

on the

Brandywine Property
Mining Lease No. 3(Lot 3480)

NTS 92J/3E
50° 05′ N 123° 08′ W
Vancouver Mining Division

for:

Avola Industries Inc.
4420, 60B Street
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December, 2002

Table of Contents

List of Appendices

List of Figures

List of Tables

List of Maps

1.0 Summary

The author was commissioned by Mr. Peter Dasler, President of Avola Industries Inc; the owner of the mining lease and claims which comprise the Brandywine Property, to carry out check sampling on parts of ten diamond drill holes drilled in 1995 and 1996 on the Brandywine Property in the Dave's Pond Zone. The purpose of this program was to confirm significant gold values returned from assays of samples taken from the 1995 and 1996 drill holes. These significant gold values form the core of an unclassified mineral resource of 73,872 tonnes grading 8.87 g/t gold with a 1.0 g/t gold cut off calculated by Kilborn Engineering Pacific Ltd. in 1996. Forty-five samples were taken of quarter core splits and were analysed for gold by fire assay at ALS Chemex's North Vancouver laboratory. Representative sub samples were pulverised and screened for metallics, with all of the +106 micron material assayed and two 30 gram subsamples of the -106 micron material assayed. The 2002 sample results were, in general, less than 10% of the 1995 and 1996 results. Forty-one samples were substantially lower than the 1995 and 1996 results, three samples correlated well and one sample was higher.

The Brandywine property has received considerable past work, beginning in the 1920's when the major known showings were staked. The property is underlain by a Lower Cretaceous, metavolcanic and metasedimentary roof pendant, the Callaghan Creek roof pendant. This unit is probably co-eval with Gambier group, which hosts the Britannia Mine. The roof pendant is enclosed by early to late Cretaceous intrusive rocks of the Coast Plutonic Complex.

There are two distinct styles of mineralization on the property. Massive to stringer base metal sulphides associated with rhyolite tuff horizons, in the Callaghan Creek roof pendant, exemplified by the Tedi Pit and structurally controlled, precious metals mineralization, near the contact of the roof pendant and the surrounding intrusives, exemplified by the Main Showing and Dave's Pond showing. The Main Showing and Dave's Pond are at the junctions of north trending faults with an east-north-east (73°) trending fault.

High grade mineralization has been shipped from both the Tedi Pit and Main Showing areas. Five hundred tons were shipped to the Cominco Smelter in Trail from the Tedi Pit in 1967 which assayed:

Pb - 14.2% Zn - 12.5% Ag - 339gm/ton Au - 2.57gm/ton.

Fifty tons were shipped to the smelter at East Helena, Montana from the Main Showing in 1965 which assayed:

Au - 83.1 gm/ton Ag - 354 gm/ton Pb - 9.9% Zn - 7.4% Cu - 0.30%.

Other structurally controlled showings, the Silver Tunnel, McKenzie Mill and Quartz Tunnel are present on the property. There are also minor skarn showings within limestone units of the Callaghan Creek roof pendant.

2.0 Conclusions

The 2002 Check Sampling Program did not confirm the results of the 1995 and 1996 sample results. The 2002 program revealed an extremely serious contamination problem in the 1995 and 1996 sample results. The mineral resource calculated from the 1995 and 1996 drill results in the Dave's Pond area should be discounted. Gold mineralization does exist in the area but it is neither as high grade nor as continuous as the 1995 and 1996 assay results indicated. To properly evaluate the resource it would be necessary to drill across the mineralizing structure which has an attitude of: Strike 73° Dip 0° to 20° south. Further drilling to test this structure should be drilled at: Azimuth 343° Inclination -45°.

Keeping in mind the above qualification, the area covered by the Brandywine Property is highly mineralized. Sections of the mineralizing structure between the Main Showing and Dave's Pond and to the east of Dave's Pond have not been drill tested. There is also the potential to outline more deposits similar to the Main Showing and Dave's Pond at similar fault intersections. Intersections of this nature exists at Grid Co-ordinates(Placer Dome 1989 grid): 47+15E 66+75N, 48+00E 66+85N, and 50+25E 67+20N.

The potential to find more base metal volcanogenic massive sulphide mineralization north of the Tedi Pit is good. This area, covered by the northern three units of the Brandy 6 claim, has not had any geophysical surveys carried out on it.

3.0 Recommendations

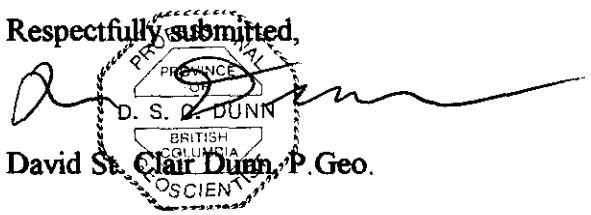
Detailed geological mapping at a scale of 1:1,000 or larger with an emphasis on structures should be carried out in the area of the Main Showing , Dave's Pond and further to the east northeast along the mineralized trend.

Three diamond drill holes, totalling approximately 300 meters, should be drilled to test the fault junctions east north east of the Main Showing and Dave's Pond. Specifically, the holes should be drilled as follows:

Collar	Bearing	Inclination	Length (PD grid 1989)
47+20E 66+40N	343°	-45°	100 m
48+20E 66+40N	343°	-45°	100 m
50+20E 66+70N	343°	-45°	100 m

A pulse-em, UTEM, or gravity geophysical survey should be carried out from Line 85 north to the northern claim boundary. The recommended work is estimated to take four weeks and cost $70,000.

Respectfully submitted,



David St. Clair Dunn, P.Geo.

4.0 Introduction

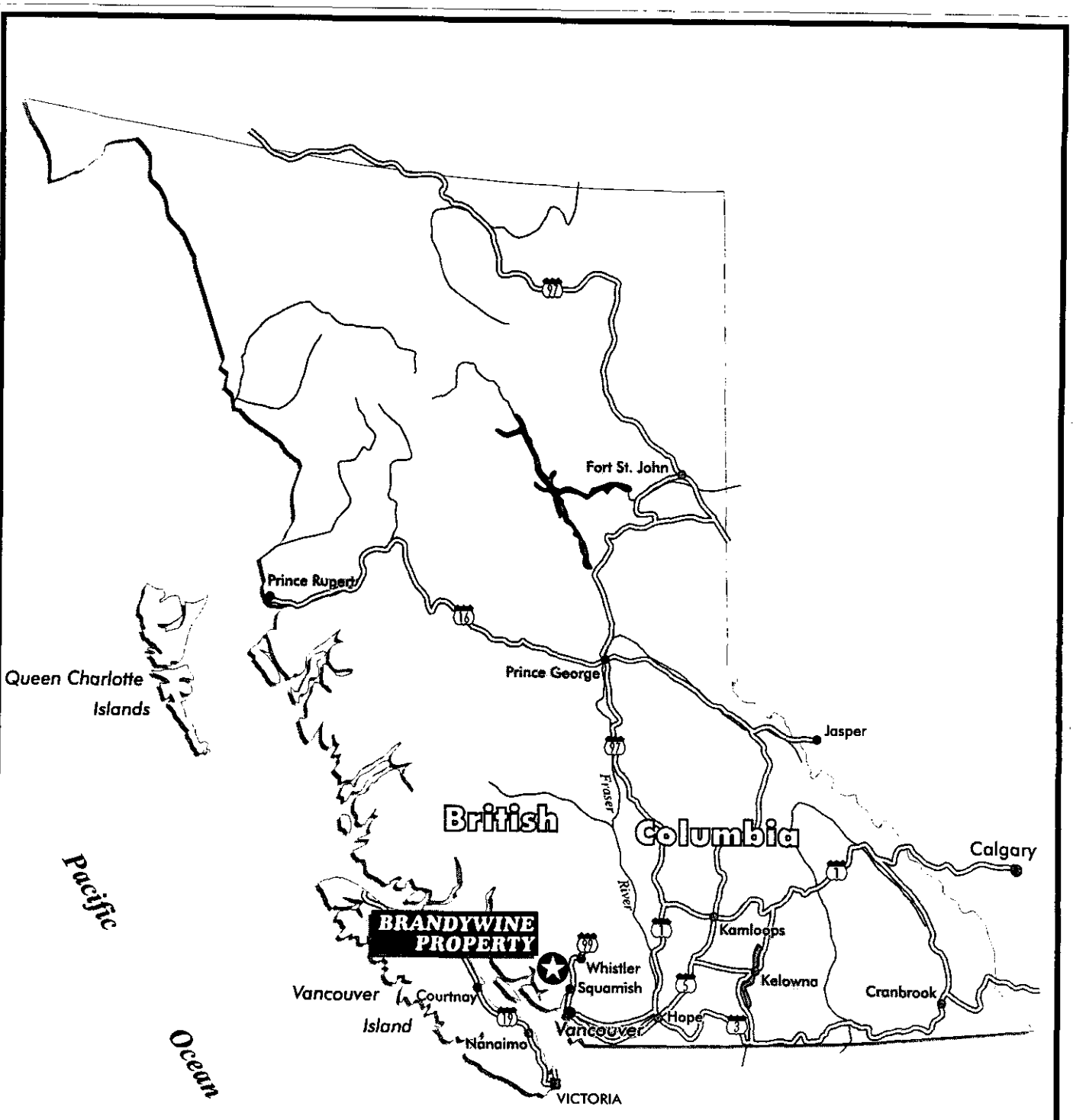
The author was commissioned by Mr. Peter Dasler, President of Avola Industries Inc., to check sample drill core from 1995 and 1996 drill programs carried out on Avola's Brandywine Property(Figs.1,2 and 4). The core is stored on the property, specifically on the Brandy A claim, approximately 20 meters west of the B.C. Rail line at the rail station of McGuire. McGuire consists of one substantial house with year round renters and several shacks and cabins. The core is in racks south and east of one of the shacks.

The author went to the site on the 18th and 19th of November, 2002 and resampled the core with the assistance of A. J. Muirhead, a geological technician. Parts of ten holes were resampled(Table 1). The core that was resampled was NQ and had been previously sampled using a diamond saw. Recovery is estimated to have been greater than 90%, judging by the amount of core remaining in the boxes. The core was mechanically quartered over the same intervals as originally sampled. Hole 96-1 had been previously resampled with a diamond saw and all of the remaining one quarter of the core was taken.

5.0 Location and Access

The Brandywine property is located approximately 100 km. north of Vancouver (Fig. 1). A major communication corridor crosses the south eastern part of the property. Provincial Highway 99, the B.C. Rail mainline and three major power lines are in this corridor.

Access from Vancouver is via Highway 99, 92 kilometres to the property. Access from Highway 99 to the western and northern parts of the property is possible via a network of logging and mine access roads. Logging mainlines follow both Brandywine and Callaghan Creeks across the property(Fig. 2).



Avola Industries Ltd.

BRANDYWINE PROPERTY

Vancouver Mining Division, B.C.

Location & Access

Scale	as shown	Date	December, 2002
N.T.S.	92J/3	By	d.s.d. / a.g.b.
Figure	1		David Dunn P. Geo.

6.0 Physiography and Climate

Topography on the property is moderate with some rugged areas. The junction of Callaghan Creek and the Chekamus River valleys is located in a large gently east sloping area, two kilometres east-west by five kilometres north-south, located in the eastern third of the property, with the western two thirds of the property dominated by the steep east-facing slopes of Mount Brew and Metal Dome Mountain and the Brandywine Creek valley. Elevations range from 400 meters ASL near Daisy Lake in the south-east corner of the property to 1400 meters ASL on the flank of Metal Dome Mountain on the western edge of the property.

The climate is West Coast Marine with generally mild temperatures, heavy spring and fall rains, and heavy winter snowfall. Much of the property(~50%) has been clear cut. The remainder is covered by mature cedar, hemlock, spruce, and Douglas fir.

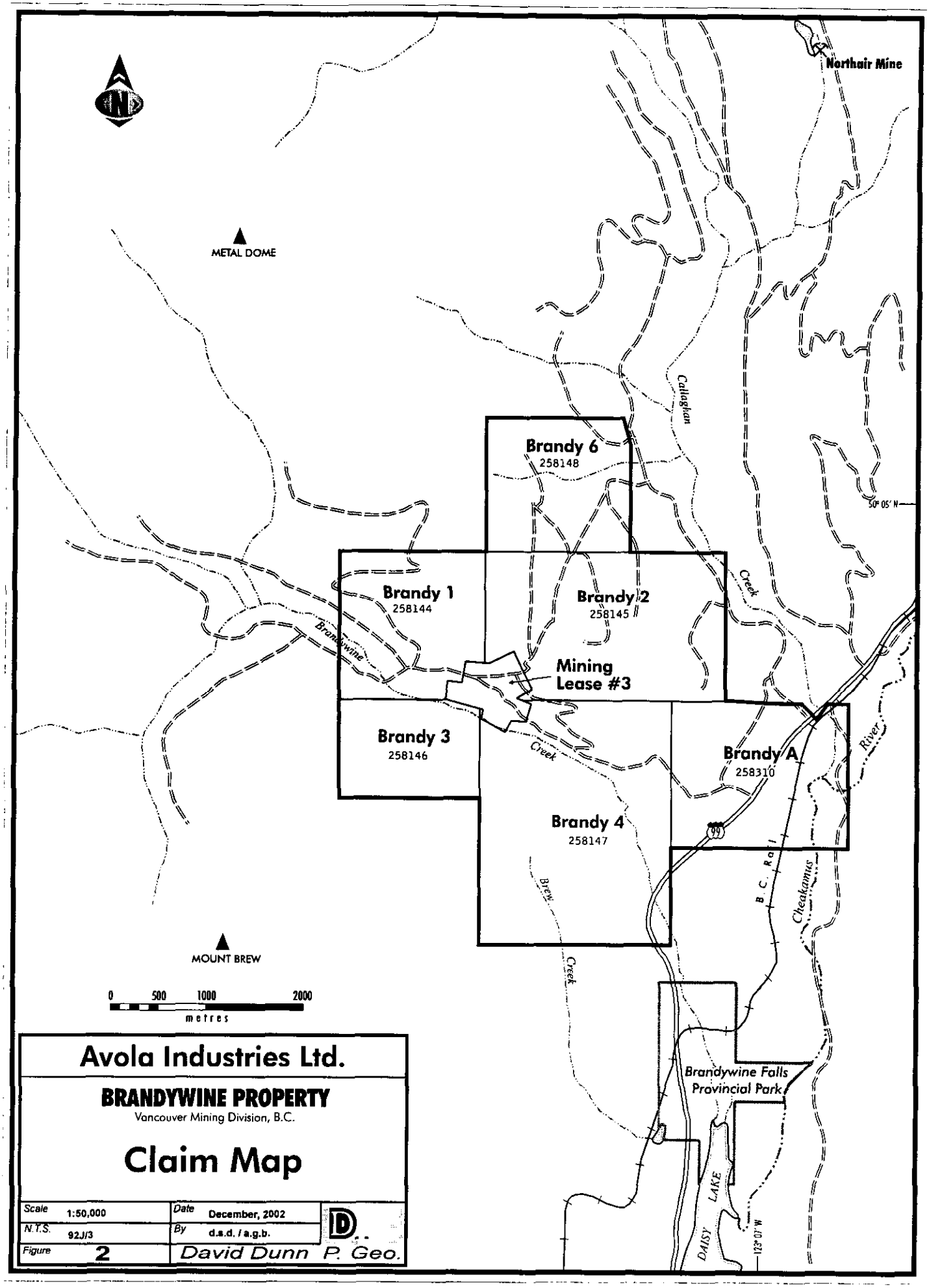
7.0 Property Status and Ownership

The Brandywine property is owned by Avola Industries Inc. with a 0.5% NSR royalty payable to Consolidated Silver Tusk Mines Ltd. The property consists of six mineral claims totalling 71 units and Mining Lease No. 3(Fig. 2). The entire property is approximately 1775 hectares in area. The following table lists claims and their status:

NAME	UNITS	RECORD NO.	EXPIRY DATE
Mining Lease No. 3	6	3480	2/9/03
Mineral Claims			
Brandy 1	9	258144	14/4/03
Brandy 2	15	258145	14/4/03
Brandy 3	6	258146	14/4/03
Brandy 4	20	258147	14/4/03
Brandy 6	9	258148	14/4/03
Brandy A	12	258310	14/4/03

8.0 Exploration History

The main showings were initially staked in the 1920's. A description of the showings appears in the 1936 Report to Minister of Mines under the names Astra, Cambria(Tedi Pit), and Blue Jack



Avola Industries Ltd.

BRANDYWINE PROPERTY

Vancouver Mining Division, B.C.

Claim Map

Scale	1:50,000	Date	December, 2002
N.T.S.	92J/3	By	d.s.d. / a.g.b.
Figure	2	David Dunn P. Geo.	

(Silver Tunnel, Main Showing).

Recent exploration includes a 50 ton bulk sample shipped to the smelter in East Helena, Montana in 1965 by Mr. Martial Levasseur.

Van Silver Explorations Limited carried out some work around the Main Showing and Silver Tunnel between 1965 and 1969.

Between 1967 and 1969 Barkley Valley Mines conducted a drilling and pitting program in the area of the Tedi Pit.

In 1969 Noranda Exploration optioned the property and completed soil geochemical, geophysical, and geological surveys over much of the property.

In 1977 Van Silver Mines Ltd. built a 150 ton per day mill, which treated mineralization from the Main Showing and Tedi Pit for a few months in the fall of 1977.

In 1978 the property was optioned to Cominco, who drill tested the Silver Tunnel and Main Showing areas.

Surface exploration was carried out in 1979,1981, and 1983 by Brandy Resources, part of the Northair group of companies.

In 1988 Placer Dome optioned the property and carried out geological, geochemical, and geophysical surveys in 1988 and 1989.

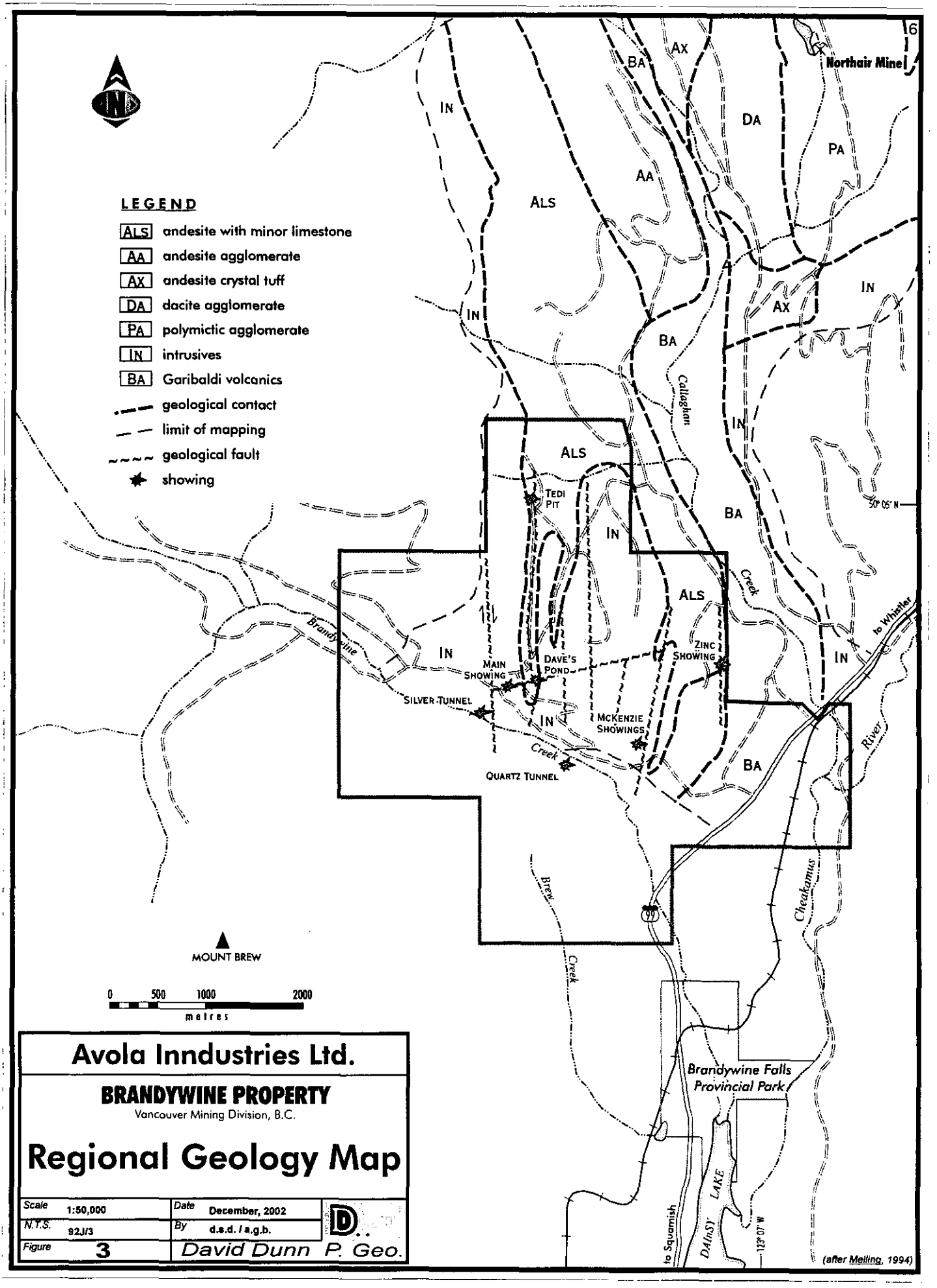
In 1991 La Rock Mining Corp. acquired the property and carried out additional geophysical surveys in 1991 and 1992.

From 1992 to the 1997 La Rock and has been drill testing the property. A total of 134 recorded diamond drill holes totalling 9892.5 meters have been drilled on the Tedi Pit, Dave's Pond, Main Showing, Zinc Zone, Little Lake, and other targets.

9.0 Regional Geology

The Brandywine Property lies within the Coast Plutonic Complex of the Canadian Cordillera(Fig.3). The property covers part of the Callaghan Creek roof pendant, a Lower Cretaceous metamorphosed volcanic-sedimentary package surrounded by Early to Late Cretaceous intrusives of the Coast Plutonic Complex. The roof pendant is probably co-eval with the Gambier Group, the host to the Britannia Mine, a volcanogenic massive sulphide deposit. This mine operated from 1905 to 1975 and produced 55 million tons of ore, grading 1.1% copper, 0.65% zinc, 0.2 oz./ton silver, and 0.02 oz./ton gold.

Structurally, the Brandywine Property covers the junction of three major regional faults trending 0°, 70°, and 108°(Pinsent, R., Personal Communication, 1998). Structurally controlled precious metals mineralization has been exploited in the region, notably at the Northair Mine, four kilometres north of the Brandywine Property. This mine operated from 1974 to 1982 and produced 582,923 tons of ore grading 0.31 oz./ton gold, 1.62 oz./ton silver, 1.13% lead and 1.54% zinc.



Avola Inndustries Ltd.

BRANDYWINE PROPERTY

Vancouver Mining Division, B.C.

Regional Geology Map

Scale	1:50,000	Date	December, 2002
N.T.S.	92J/3	By	d.s.d. / a.g.b.
Figure	3	David Dunn P. Geo.	

10.0 Property Geology and Mineralization

The northern half of the Brandywine Property covers part of the Callaghan Creek roof pendant which interfingers with intrusives of the Coast Plutonic Complex in the central and southern parts of the property(Map 1). The eastern edge of the property is covered by a thin veneer of very recent vesicular basalt, part of the Garibaldi Volcanics. The roof pendant consists mainly of andesite to dacite flows and pyroclastics, with minor rhyolite and limestone. These rocks strike north-westerly and dip steeply both east and west. They have been altered to green schist facies on a regional scale, with more intense alteration near the larger structures. Within this sequence, volcanogenic massive sulphide mineralization can be found, spatially, and probably genetically, associated with a rhyolite dome and tuff horizon. The most developed showing of this type is the Tedi Pit. The mineralization consists of massive to stringer pyrite-galena-sphalerite-chalcopyrite. Five hundred tons, grading 14.2% lead, 12.5% zinc 339 grams/tonne silver, and 2.57 grams/tonne gold are reported (Melling,1994) to have been shipped to the Cominco Smelter in Trail in 1967. Fifty-nine recorded diamond drill holes totalling over 3000 meters have been drilled in the immediate area Of the Tedi Pit. This work has not succeeded in outlining a significant mineral resource.

The Coast Plutonic Complex intruding and surrounding the Callaghan Creek roof pendant consists of at least eight different units on the property including diorite, granodiorite, and late stage felsic and andesitic dikes. Most contacts between the intrusives and volcanic-sedimentary rocks are metasomatic contacts, making exact lithologic boundaries difficult to determine. Near the contacts, precious metals showings are present, localized at the junctions of north trending faults and an east-north-east trending fault. The rocks near the structures exhibit a higher level of alteration, including silicification and argillic alteration. The showings of this type that have been best developed to date are the Silver Tunnel, Main Zone, and Dave's Pond. A 50 ton bulk sample grading 83.1 grams/tonne gold, 354 grams/tonne silver, 9.9% lead, 7.4% zinc, 0.30% copper is reported to have been shipped to East Helena, Montana (Melling,1994). Seventy-five recorded diamond drill holes, totalling more than 6,000 meters have been drilled on these three zones. The Main Showing and Dave's Pond appear to be steeply dipping pencil shaped mineralized bodies, following the fault junctions. Drilling has shown these deposits are of limited size.

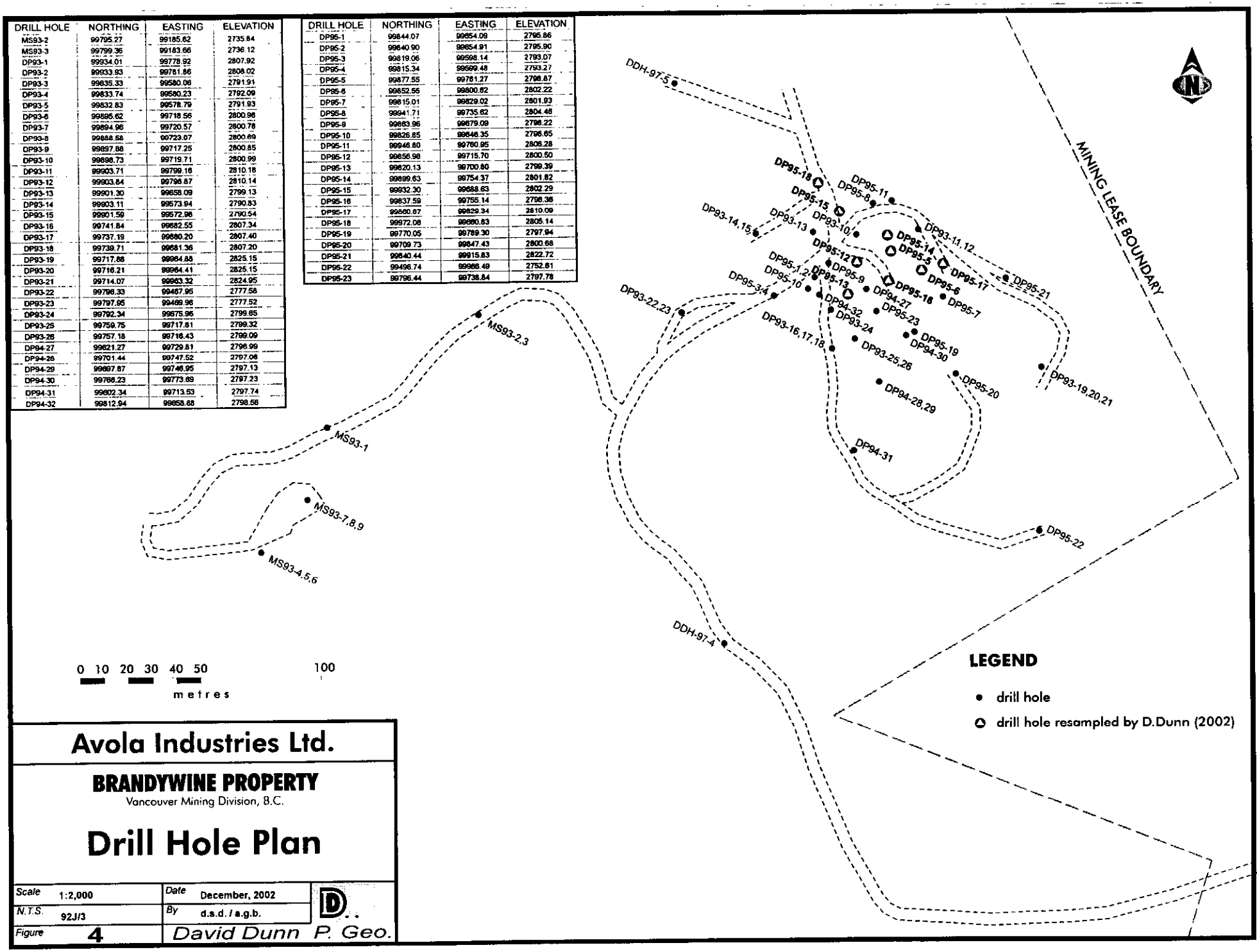
The only reliable resource developed by the drilling is 73,872 tonnes grading 8.87 grams/tonne with a 1.0 g/t cut off, calculated by Kilborn Engineering Pacific Ltd. in 1997 to be present at Dave's Pond(Kilborn,1997). The work that is the subject of this report was designed to check the accuracy of the original assays that formed the core of the Kilborn resource estimate.

11.0 2002 Check Sampling Program

The object of the 2002 program was to check assays from core samples taken from 1995 and 1996 drill holes(Fig. 4). Forty-five quarter core splits were taken over the same intervals as 45 of the 1995 and 1996 samples(Table 1). The core was washed in water before quartering in an attempt to remove any possible surface contamination. Quatered samples were placed in a

DRILL HOLE	NORTHING	EASTING	ELEVATION
MS93-2	99795.27	99185.62	2735.84
MS93-3	99799.36	99183.66	2736.12
DP93-1	99934.01	99778.92	2807.92
DP93-2	99933.93	99761.86	2808.02
DP93-3	99835.33	99580.06	2791.91
DP93-4	99833.74	99580.23	2792.09
DP93-5	99832.83	99578.79	2791.93
DP93-6	99895.62	99718.56	2800.96
DP93-7	99894.96	99720.57	2800.78
DP93-8	99888.58	99723.07	2800.69
DP93-9	99897.88	99717.25	2800.85
DP93-10	99898.73	99719.71	2800.99
DP93-11	99903.71	99799.16	2810.18
DP93-12	99903.84	99796.87	2810.14
DP93-13	99901.30	99658.09	2799.13
DP93-14	99903.11	99573.94	2790.83
DP93-15	99901.59	99572.96	2790.54
DP93-16	99741.84	99682.55	2807.34
DP93-17	99737.19	99680.20	2807.40
DP93-18	99739.71	99681.36	2807.20
DP93-19	99717.88	99864.88	2825.15
DP93-20	99716.21	99864.41	2825.15
DP93-21	99714.07	99863.32	2824.95
DP93-22	99796.33	99467.95	2777.58
DP93-23	99797.95	99469.96	2777.52
DP93-24	99792.34	99675.96	2799.85
DP93-25	99759.75	99717.81	2799.32
DP93-26	99757.18	99716.43	2799.09
DP94-27	99821.27	99729.81	2796.99
DP94-28	99791.44	99747.52	2797.06
DP94-29	99887.87	99746.95	2797.13
DP94-30	99766.23	99773.69	2797.23
DP94-31	99802.34	99713.53	2797.74
DP94-32	99812.94	99658.88	2798.56

DRILL HOLE	NORTHING	EASTING	ELEVATION
DP95-1	99844.07	99854.09	2795.86
DP95-2	99840.90	99854.91	2795.90
DP95-3	99819.06	99598.14	2793.07
DP95-4	99815.34	99599.48	2793.27
DP95-5	99877.55	99781.27	2798.87
DP95-6	99852.55	99800.82	2802.22
DP95-7	99815.01	99829.02	2801.93
DP95-8	99941.71	99735.62	2804.48
DP95-9	99883.96	99679.09	2796.22
DP95-10	99826.85	99846.35	2796.65
DP95-11	99946.80	99780.95	2806.28
DP95-12	99856.98	99715.70	2800.50
DP95-13	99820.13	99700.80	2799.39
DP95-14	99889.63	99754.37	2801.82
DP95-15	99932.30	99688.63	2802.29
DP95-16	99837.59	99755.14	2798.36
DP95-17	99880.67	99829.34	2810.09
DP95-18	99972.08	99880.83	2805.14
DP95-19	99770.05	99789.30	2797.94
DP95-20	99709.73	99847.43	2800.68
DP95-21	99840.44	99915.83	2822.72
DP95-22	99496.74	99866.49	2752.61
DP95-23	99796.44	99738.84	2797.78



Avola Industries Ltd.

BRANDYWINE PROPERTY

Vancouver Mining Division, B.C.

Drill Hole Plan

Scale	1:2,000	Date	December, 2002
N.T.S.	92J/3	By	d.s.d. / a.g.b.
Figure	4		David Dunn P. Geo.

DP Zone ******** **TABLE1** ********

HOLE	SAMPLE	INTERVAL	GRADE(oz/t)	< to CA		1995-96 Grade(g/t)	2002 Grade(g/t)	2002 SAMPLE
95-5	11286	103-105.5	2.12	5 to 10		65.9	2.16	198951
	11291	118-120.5	1.43	30-Jan		44.5	0.6	952
	11286	120.5-123	2.4	?		74.64	1.89	953
95-6	11227	66-71	1.851	II to 60		57.57	2.73	954
	11228	71-76	1.293		90	40.21	1.85	955
	11229	76-81	0.302		90	9.39	0.16	956
	11246	93-98	1		30	31.1	3.39	957
	11247	98-103	1.429	brxx		44.44	3.85	958
	11248	103-108	1.35		30	41.98	0.67	959
	11249	108-113	0.342	II to CA		10.64	0.08	960
95-12	12377	28-33	2.393	II to CA		74.42	0.17	961
	12378	33-38	0.239	II to CA		7.43	0.26	962
	12392	90.5-93	1.82		60	56.6	1.89	963
	12402	133-138	2.248		80	69.91	<0.05	964
95-13	12451	18-28	2.355		30	72.24	0.05	965
	12452	28-33	0.394		30	12.25	0.32	966
	12453	33-38	1.328		25	41.3	1.14	967
	12454	38-43	0.967		30	30.07	0.5	968
	12455	43-48	1.554		30	48.33	<0.05	969
95-14	12506	68-73	0.432		15	13.44	1.97	970
	12509	73-78	0.945		20	29.39	0.07	971
	12511	83-88	0.492		20	15.3	0.08	972
	12513	93-98	0.566	II to 20		17.6	12.7	973
	12514	98-103	0.168	II to 20		5.22	2.52	974
	12515	103-108	0.34	II to 20		10.57	10.5	975
	12515	108-113	0.953	II to 20		29.64	1.46	976
	12516	113-118	0.172	II to 20		5.35	5.26	977
95-15	12549	27-33	0.133	II		4.14	2.26	978
	12565	98-103	0.607		20	18.88	0.15	979
	12568	113-118	0.318		15	9.89	0.07	980
	12569	118-123	0.006		15	0.187	0.27	981
	12570	123-128	0.81		25	25.19	0.06	982
95-16	12618	88-93	0.341	II to 20		10.6	0.15	983
	12619	93-98	0.222	II to 20		6.9	0.16	984
	12625	123-128	1.038		40	32.28	0.35	985
	12626	128-133	0.523		40	16.26	0.09	986
95-17	12669	82-88	0.034	15 to 30		1.06	0.09	987
	12670	88-93	0.375	15 to 30		11.66	36.9	988
	12674	108-113	0.552	15 to 30		17.17	0.84	989
95-18	12723	63-68	0.597	40 to 70		18.57	<0.05	990
96-1		153-158	0.541			16.82	<0.05	991
		158-163	1.8			55.98	<0.05	992
		178-183	0.668			20.77	<0.05	993
		188-193	1.309			40.71	<0.05	994
		209-214	1.418			44.1	<0.05	995

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2.28

1.85

0.15

30.38

DP Zone ******** TABLE1 ********

HOLE	SAMPLE	INTERVAL	GRADE(oz/t)	< to CA	1995-96 Grade(g/t)	2002 Grade(g/t)	2002 SAMPLE
95-5	11286	103-105.5	2.12	5 to 10	65.9	2.16	198951
	11291	118-120.5	1.43	30 to CA	44.5	0.6	952
	11286	120.5-123	2.4	?	74.64	1.89	953
95-6	11227	66-71	1.851	ll to 60	57.57	2.73	954
	11228	71-76	1.293	90	40.21	1.85	955
	11229	76-81	0.302	90	9.39	0.16	956
	11246	93-98	1	30	31.1	3.39	957
	11247	98-103	1.429	brxx	44.44	3.85	958
	11248	103-108	1.35	30	41.98	0.67	959
	11249	108-113	0.342	ll to CA	10.64	0.08	960
95-12	12377	28-33	2.393	ll to CA	74.42	0.17	961
	12378	33-38	0.239	ll to CA	7.43	0.26	962
	12392	90.5-93	1.82	60	56.6	1.89	963
	12402	133-138	2.248	80	69.91	<0.05	964
95-13	12451	18-28	2.355	30	72.24	0.05	965
	12452	28-33	0.394	30	12.25	0.32	966
	12453	33-38	1.328	25	41.3	1.14	967
	12454	38-43	0.967	30	30.07	0.5	968
	12455	43-48	1.554	30	48.33	<0.05	969
95-14	12506	68-73	0.432	15	13.44	1.97	970
	12509	73-78	0.945	20	29.39	0.07	971
	12511	83-88	0.492	20	15.3	0.08	972
	12513	93-98	0.566	ll to 20	17.6	12.7	973
	12514	98-103	0.168	ll to 20	5.22	2.52	974
	12515	103-108	0.34	ll to 20	10.57	10.5	975
	12515	108-113	0.953	ll to 20	29.64	1.46	976
	12516	113-118	0.172	ll to 20	5.35	5.26	977
95-15	12549	27-33	0.133	ll	4.14	2.26	978
	12565	98-103	0.607	20	18.88	0.15	979
	12568	113-118	0.318	15	9.89	0.07	980
	12569	118-123	0.006	15	0.187	0.27	981
	12570	123-128	0.81	25	25.19	0.06	982
95-16	12618	88-93	0.341	ll to 20	10.6	0.15	983
	12619	93-98	0.222	ll to 20	6.9	0.16	984
	12625	123-128	1.038	40	32.28	0.35	985
	12626	128-133	0.523	40	16.26	0.09	986
95-17	12669	82-88	0.034	15 to 30	1.06	0.09	987
	12670	88-93	0.375	15 to 30	11.66	36.9	988
	12674	108-113	0.552	15 to 30	17.17	0.84	989
95-18	12723	63-68	0.597	40 to 70	18.57	<0.05	990
96-1		153-158	0.541		16.82	<0.05	991
		158-163	1.8		55.98	<0.05	992
		178-183	0.668		20.77	<0.05	993
		188-193	1.309		40.71	<0.05	994
		209-214	1.418		44.1	<0.05	995

plastic sample bag with the corresponding sample tag and delivered to ALS Chemex's laboratory in North Vancouver. Samples were crushed to -10 mesh and a representative 500 gram split taken. The 500 gram split was then pulverised to -100 mesh. Two representative 30 gram splits were then screened at 106 microns. All of the oversize was fire assayed and two 30 gram fire assays were carried out on the undersize(Appendix A). Results were then calculated to give a representative assay for the whole sample.

Correlation with the 1995 and 1996 assays was extremely poor. The 2002 samples averaged less than 5% of the gold values of the 1995 and 1996 samples. Forty-one samples were much lower than the 1995-1996 results, three samples correlated well and one was higher. The only reasonable conclusion is that the 1995 and 1996 samples were systematically contaminated. All of the 1995 and 1996 assay results should be discounted and the resource calculated from these results likewise discounted.

The parts of the core that were resampled were also relogged(Appendix B). Relogging the core showed that the 1995 and 1996 drill holes, drilled vertically, were drilled down or within 15° of a near vertical structure, making analysis of the results and resource calculations very difficult.

12.0 Bibliography

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Appendix A

Sample Results and Analytical Methods

ALS Chemex
EXCELLENCE IN ANALYTICAL CHEMISTRY
ALS Canada Ltd.
212 Brooksbank Avenue
North Vancouver BC V7J 2C1 Canada
Phone: 604 984 0221 Fax: 604 984 0218

To: AVOLA INDUSTRIES
750 GRAND BLVD.
NORTH VANCOUVER BC V7L 3W4

Date : 4-Dec-2002
Account: AVOIND

CERTIFICATE VA02006203

Project : Brandy Wine

P.O. No:

This report is for 45 ROCK samples submitted to our lab in North Vancouver, BC, Canada on 22-Nov-2002.

The following have access to data associated with this certificate:

PETER DASLER
PETER DASLER
DAVID DUNN

SAMPLE PREPARATION

ALS CODE	DESCRIPTION
WEI-21	Received Sample Weight
LOG-22	Sample login - Rcd w/o BarCode
CRU-31	Fine crushing - 70% <2mm
SPL-21	Split sample - riffle splitter
PUL-32	Pulverize 1000g to 85% < 75 um
SCR-21	Screen to -100 um

ANALYTICAL PROCEDURES

ALS CODE	DESCRIPTION	INSTRUMENT
Au-SCR21	Screen Fire Assay Au - 100 um	WST-SIM
Au-GRA21	Au 30g FA-GRAV finish	WST-SIM
Au-GRA21d	Au 30g FA-GRAV finish - DUP	WST-SIM

To: **AVOLA INDUSTRIES**
ATTN: DAVID DUNN
1154 MARINE DRIVE
GIBSONS BC V0N 1V1

This is the Final Report and supersedes any preliminary report with this certificate number. Results apply to samples as submitted. All pages of this report have been checked and approved for release.

Signature:

ALS Chemex
EXCELLENCE IN ANALYTICAL CHEMISTRY
ALS Canada Ltd.
212 Brooksbank Avenue
North Vancouver BC V7J 2C1 Canada
Phone: 604 984 0221 Fax: 604 984 0218

To: AVOLA INDUSTRIES
750 GRAND BLVD.
NORTH VANCOUVER BC V7L 3W4

Date : 4-Dec-2002
Account: AVOIND

Project : Brandy Wine

CERTIFICATE OF ANALYSIS VA02006203

Sample Description	Method Analyte Units LOR	WEI-21 Recvd Wt kg 0.02	Au-SCR21 Au Total ppm 0.05	Au-SCR21 Au (+) F ppm 0.05	Au-SCR21 Au (-) F ppm 0.05	Au-SCR21 Au (+) m mg 0.001	Au-SCR21 WT. + Fr g 0.01	Au-SCR21 WT. - Fr g 0.1	Au-GRA21 Au ppm 0.05	Au-GRA21d Au ppm 0.05
198951		1.02	2.16	19.00	1.76	0.410	21.57	912.0	1.76	1.74
198952		0.82	0.60	0.28	0.61	0.005	17.66	746.1	0.55	0.66
198953		0.86	1.89	5.04	1.80	0.116	23.03	747.2	1.56	2.03
198954		1.62	2.73	58.1	1.78	1.514	26.04	1505.5	1.30	2.25
198955		1.80	1.85	25.1	1.37	0.789	31.43	1522.0	0.99	1.75
198956		1.56	0.16	0.64	0.16	0.003	4.68	1316.0	0.07	0.25
198957		1.24	3.39	19.35	3.00	0.525	27.12	1108.0	2.97	3.03
198958		1.68	3.85	8.69	3.74	0.295	33.94	1489.5	3.70	3.78
198959		1.26	0.67	1.69	0.64	0.050	29.61	1119.0	0.56	0.72
198960		1.76	0.08	0.12	0.08	0.003	24.44	1630.0	0.06	0.10
198961		1.50	0.17	<0.05	0.17	<0.001	4.91	1326.5	0.13	0.22
198962		1.54	0.26	0.18	0.26	0.004	21.67	1353.5	0.30	0.23
198963		0.92	1.89	8.65	1.75	0.146	16.87	843.9	1.87	1.63
198964		1.68	<0.05	<0.05	<0.05	0.001	24.96	1517.0	<0.05	<0.05
198965		1.48	0.05	0.15	0.05	0.004	27.21	1293.5	<0.05	0.07
198966		1.44	0.32	0.27	0.32	0.005	18.41	1210.0	0.20	0.43
198967		1.56	1.14	1.29	1.13	0.046	35.59	1323.5	1.10	1.17
198968		0.90	0.50	0.60	0.49	0.021	34.93	789.4	0.44	0.55
198969		1.56	<0.05	0.14	<0.05	0.005	36.46	1356.5	<0.05	0.06
198970		2.08	1.97	4.89	1.89	0.246	50.27	1791.5	1.92	1.87
198971		1.62	0.07	0.11	0.07	0.002	17.49	1512.5	0.06	0.07
198972		1.92	0.08	<0.05	0.08	<0.001	16.33	1758.5	0.07	0.10
198973		1.92	12.70	84.1	12.00	1.518	18.05	1761.5	12.10	11.90
198974		1.98	2.52	3.57	2.50	0.076	21.30	1817.5	2.47	2.54
198975		1.96	10.50	97.7	9.69	1.551	15.88	1724.5	10.00	9.37
198976		1.82	1.46	1.58	1.45	0.028	17.69	1589.0	1.44	1.46
198977		1.94	5.26	18.90	5.13	0.324	17.16	1799.0	5.38	4.88
198978		1.22	2.26	51.3	1.68	0.642	12.51	1071.0	1.73	1.64
198979		1.40	0.15	<0.05	0.15	<0.001	15.85	1258.5	0.15	0.16
198980		1.68	0.07	<0.05	0.07	<0.001	16.73	1574.5	0.11	<0.05
198981		1.70	0.27	3.71	0.24	0.052	14.03	1552.5	0.28	0.19
198982		1.36	0.06	<0.05	0.06	<0.001	17.63	1219.5	0.10	<0.05
198983		1.28	0.15	0.61	0.14	0.009	14.76	1190.0	0.07	0.21
198984		1.88	0.16	0.15	0.16	0.004	26.28	1608.0	0.15	0.17
198985		1.90	0.35	<0.05	0.35	<0.001	9.49	1793.0	0.37	0.33
198986		1.86	0.09	<0.05	0.09	<0.001	16.89	1737.0	0.11	0.07
198987		1.98	0.09	0.41	0.09	0.005	12.24	1864.0	0.07	0.11
198988		1.60	36.9	582	28.8	12.406	21.31	1447.0	29.3	28.3
198989		2.02	0.84	1.14	0.84	0.019	16.69	1761.5	0.79	0.88
198990		1.56	<0.05	<0.05	<0.05	<0.001	18.75	1348.5	<0.05	<0.05

ALS Chemex
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To: AVOLA INDUSTRIES
750 GRAND BLVD.
NORTH VANCOUVER BC V7L 3W4

Date : 4-Dec-2002
Account: AVOIND

Project : Brandy Wine

CERTIFICATE OF ANALYSIS VA02006203

Sample Description	Method Analyte Units LOR	WEI-21 Recvd Wt kg 0.02	Au-SCR21 Au Total ppm 0.05	Au-SCR21 Au (+) F ppm 0.05	Au-SCR21 Au (-) F ppm 0.05	Au-SCR21 Au (+) m mg 0.001	Au-SCR21 WT. + Fr g 0.01	Au-SCR21 WT. - Fr g 0.1	Au-GRA21 Au ppm 0.05	Au-GRA21d Au ppm 0.05
198991		1.80	<0.05	<0.05	<0.05	<0.001	20.89	1662.0	<0.05	<0.05
198992		1.92	<0.05	<0.05	<0.05	<0.001	29.60	1567.0	<0.05	<0.05
198993		1.68	<0.05	<0.05	<0.05	<0.001	16.82	1540.5	<0.05	<0.05
198994		1.90	<0.05	<0.05	<0.05	<0.001	8.58	1793.0	<0.05	<0.05
198995		2.02	<0.05	<0.05	<0.05	<0.001	6.65	1676.0	<0.05	<0.05

ALS Chemex

Fire Assay Procedure – Au-SCR21
Precious Metals Analysis – Screen Metallics Gold, Double Minus

Sample Decomposition: Fire Assay Fusion
Analytical Method: Gravimetric

The final prepared pulp is passed through a 106 micron (Tyler 150 mesh) screen to separate the oversize fractions. Any +106 micron material remaining on the screen is retained and analyzed in its entirety by fire assay. The -106 micron fraction is homogenized and two subsamples are analyzed by fire assay with gravimetric finish. The average of the two results is used in calculating the combined gold content of the plus and minus fractions.

In the fire assay procedure, the sample is fused with a mixture of lead oxide, sodium carbonate, borax, silica and other reagents as required in order to produce a lead button. The lead button, containing the precious metals, is cupelled to remove the lead and the resulting precious metal bead is parted in dilute nitric acid, annealed and weighed to determine gold content.

The gold values for both +106 and -106 micron fractions are reported together with the weight of each fraction as well as the calculated total gold content of the sample.

Calculations:

$$\text{Au}^{-}\text{avg} = \frac{\text{Au}^{-}(1) + \text{Au}^{-}(2)}{2}$$

$$AuTotal(g/t) = \frac{(Au-avg(g/t) \times Wt.Minus(g) \times 10^{-6} t/g) + (\text{Weight Au in Plus(mg)} \times 10^{-3} g/mg)}{(Wt.Minus(g) + Wt.Plus(g)) \times 10^{-6} t/g}$$

December 5, 2002

Appendix B

Diamond Drill Logs

Diamond-Drill Record

Property Brandywine Hole No. 95-5, 95-6 Sheet No. 1 of 6

DIP TEST		
	ANGLE	
Depth	Reading	Corrected
0	90°	95-5
0	90°	95-6

UTM _____ TOTAL DEPTH 404.00 ft, 318.00 ft. DATE BEGUN _____

AZIMUTH 0 GRID LOCATION 99761.27E, 99877.55N, 99852.35N 2802.62E DATE FINISHED Oct. 1995

INCLINATION -90° CROSS SECTION Fig. 4 DATE LOGGED 18/11/02

COLLAR ELEVATION 2798.87ft, 2802.22ft. CORE SIZE NQ LOGGED BY David Dunn

Check Sampling

Depth Feet From	To	App. Width	Description	Sample No.	From	To	App. Width	1995 Sample	Au ppb	Cu %	Ag ppm	Zn %
95-5, 103'	105.5'	2.5'	Dark Green Andesite	198 951	103	105.5	2.5	11286				
			Strongly schistose									
			sub // to CA.									
			Non Magnetic. Moderate									
			Carbonate alteration.									
			Abundant epidote,									
			chlorite. 1-2%									
			Sulphides: pyrite,									
			arsenopyrite, minor									
			chalcopyrite.									
118'	120.5'	2.5'	As above but more brecciated	952	118'	120.5	2.5'	11291				
			Strong Carbonate alteration					~~11291~~				
120.5'	123.0'	2.5'	As above.	953				11292				
95-6			sub // to CA									
66'	~~76~~ 71'	~~5~~ 5'	Strongly foliated, light green	954	66	71'	5'	11227				
71'	76'	5'	Andesite Porphyry. Intense	955	71'	76'	5'	11228				
76'	81'	5'	propylitic alt. 1-2% sulphides	956	76'	81'	5'	11229				

Diamond-Drill Record

Property Brandywine **Hole No.** 95-6, 95-12 **Sheet No.** 2 of 6

DIP TEST		
	ANGLE	
Depth	Reading	Corrected
0	-90°	95-12

UTM _____ TOTAL DEPTH 352.00ft DATE BEGUN _____

AZIMUTH 0 GRID LOCATION 99856.8N 99715.7E DATE FINISHED Oct. 95

INCLINATION -90° CROSS SECTION Fig. 4 DATE LOGGED 18/11/02

COLLAR ELEVATION 2800.50' CORE SIZE NQ LOGGED BY David Dunn

Check Sampling 1

Depth From	Depth To	App. Width	Description	Sample No.	From	To	App. Width	1995 ~~Rec.~~ Sample	Au ppb	Cu %	Ag ppm	Zn %
95-6				198 957	93'	98'	5'	112246				
93'	113	20'	Dark green Andesite porphyry	958	98'	103'	5'	112247				
			Intense Propylitic Alteration	959	103'	108'	5'	112248				
			Strongly schistose 0-20° to CA.	960	108'	113'	5'	112249				
			Strong Carbonate Alteration									
			Moderate Silicification. 1-2% Py.									
95-12												
28'	38'	10'	Light green, strongly silicified,	961	28'	33'	5'	112277				
			weak carbonate, brecciated	962	33'	38'	5'	112278				
			~~Andesite~~ Dacite. 3% Sulphides.									
			Siderite Str. Foliation 0 20° to CA									
90.5'	93.0'	2.5'	Dk. gr. str. sil., mod. carb.	963	90.5'	93.0'	2.5'	123092				
			1% Py. Foliation sub // to CA									
				964	133'							
133'	138'	5'	Dk. gr. schistose And. Porph.	~~131~~	~~133~~	138'	5'	124002				
			2-3% Py. Aspy, gz									

Diamond-Drill Record

Property Brandywine **Hole No.** 95-13, 95-14 **Sheet No.** 3 of 6

DIP TEST		
	ANGLE	
Depth	Reading	Corrected
0	-90°	95-13
0	-90°	95-14

UTM _____ **TOTAL DEPTH** 308' **DATE BEGUN** _____

AZIMUTH 0 **GRID LOCATION** 99820.13N 99879.63N / 99700.80E 99754.37E **DATE FINISHED** Oct., 1995

INCLINATION -90° **CROSS SECTION** Fig. 4 **DATE LOGGED** 18/11/02

COLLAR ELEVATION 2799.37', 2810.82' **CORE SIZE** NQ **LOGGED BY** David Dunn

Check Sampling

Depth From	Depth To	App. Width	Description	Sample No.	From	To	App. Width	1995 Sample	Au ppb	Cu %	Ag ppm	Zn %
95-13			18'-28' Oxidized. Grey to	965	18'	28'	10'	124 51				
18'	48'	30'	grey brown. Strongly schistose	966	28'	33'	5'	124 52				
			// to CA. Talc on fractures	967	33'	38'	5'	124 53				
			Very broken ≈ 50% recovery.	968	38'	43'	5'	124 54				
			28'-35' Oxidized Grey brown	969	43'	48'	5'	124 55				
			Schistose // to CA. Much talc.									
			Silicified + brecciated.									
			1% Sulphides.									
			35'-48' Oxidized. Grey Brown									
			Talc. Carbonate weathered out.									
95-14												
68'	~~78~~ 78'	10'	Diorite Brxx. Strong	970	68'	73'	5'	125 06				
			propylitic alt. Qtz. Carb.	971	73'	78'	5'	125 09				
			Stringers // to CA.									
			3% Sulphides. Py, Cupy,									
			Aspy. Stringers of									
			Grey Qtz // to CA.									

Diamond-Drill Record

Property Brandywine **Hole No.** 95-14, 95-15 **Sheet No.** 4 of 6

DIP TEST		
	ANGLE	
Depth	Reading	Corrected
0'	-90°	95-14
0	-90°	95-15

UTM _____ **TOTAL DEPTH** 348.00' **DATE BEGUN** _____

AZIMUTH 0 **GRID LOCATION** 99932.32 / 99688.63 **DATE FINISHED** Oct. 95

INCLINATION -90° **CROSS SECTION** Fig. 4. **DATE LOGGED** 18/11/02

COLLAR ELEVATION 2802.29 **CORE SIZE** NQ. **LOGGED BY** David Dunn

Check Sampling

Depth From	Depth To	App. Width	Description	Sample No.	From	To	App. Width	1995 Sample	Au ppb	Cu %	Ag ppm	Zn %
95-14			Light green, strongly silicified	198 922	83'	88'	5'	125 71				
83'	118'	35'	qtz-chlorite schist. 1854	973	93'	98'	5'	125 73				
			Schistosity 11 to 10° to CA	974	98'	103'	5'	125 74				
			Many Qtz stringers	975	103'	108'	5'	125 75				
			white to dark grey 11 to CA	976	108'	113'	5'	125 76				
				977	113'	118'	5'	125 77				
95-15												
27'	33'	6'	Grey green, v. broken,	978	27'	33'	6'	125 99				
			sand sections. Strong									
			schistosity 11 to CA. Talc.									
			siderite. Fault zone									
			11 to CA.									
98'	128'	30'	Dark green, sil, brxx.	979	98'	103'	5'	125 65				
			Diorite. Str. Propylitic	980	103'	108'	5'	125 68				
			7 lt. 1-2% sulphides py,	981	108'	123'	5'	125 69				
			cpy. Qtz stringers +	982	123'	128'	5'	125 70				
			foliation 11 to CA.									

Diamond-Drill Record

Property Brandywine **Hole No.** 95-16, 95-17 **Sheet No.** 5 of 6

DIP TEST		
	ANGLE	
Depth	Reading	Corrected
0	-90°	95-16
0	-90°	95-17

UTM _____ **TOTAL DEPTH** 318 ft. **DATE BEGUN** _____

AZIMUTH 0 **GRID LOCATION** 99755.13E, 99837.59N, 99886.67, 99839.34 **DATE FINISHED** Oct. 1995

INCLINATION -90° **CROSS SECTION** Fig. 4 **DATE LOGGED** 19/11/02

COLLAR ELEVATION 2798.36, 2810.09 **CORE SIZE** NQ **LOGGED BY** David Dunn

Check Sampling

Depth From	Depth To	App. Width	Description	Sample No.	From	To	App. Width	1995 Rec Sample	Au ppb	Cu %	Ag ppm	Zn %
95-16				198 983	89	93'	4'	126 18				
89'	98'	9'	Si.I. brxx Diorite, Dk. green	984	93'	98'	5'	126 19				
			Sto. propylitic alt. Foliation									
			11 to 10° to CA. Qtz. Stringers									
			grey to blue 11 to 10° to CA									
			1% sulphides									
123'	133'	10'	123'-126' - Strongly bleached	985	123'	128'	5'	126 25				
			+ broken. Foliation 30° to CA	986	128'	133'	5'	126 26				
			126'-133'. Diorite w/ strong									
			propylitic alteration - 0.5%									
			Dark green. Minor qtz stringers									
			30° to CA.									
95-17												
82'	93'	11'	Bleached, foliated Diorite?	987	82'	88'	6'	126 69				
			Grey to white qtz stringers	988	88'	93'	5'	126 70				
			45° to 10° to CA.									

Diamond-Drill Record

Property Brianty wine **Hole No.** 95-17, 95-18, 96-1 **Sheet No.** 6 of 6

DIP TEST		
	ANGLE	
Depth	Reading	Corrected
0	−90°	95-18
?	?	96-1

UTM _____ **TOTAL DEPTH** 368', ? **DATE BEGUN** _____

AZIMUTH 0° **GRID LOCATION** 99660.83, 99572.08, ? **DATE FINISHED** Oct. 1995, 1996

INCLINATION 90° **CROSS SECTION** Fig. 4 **DATE LOGGED** 19/11/02

COLLAR ELEVATION 2810.09', ? **CORE SIZE** N.Q. **LOGGED BY** David Dunn

Check Sampling

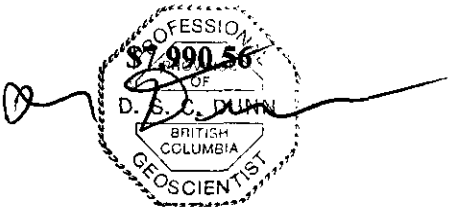
Depth From	Depth To	App. Width	Description	Sample No.	From	To	App. Width	1995 Rec. Sample	Au ppb	Cu %	Ag ppm	Zn %
95-17			Dark green Sil. And.	988 989	108	113	5'	126 74				
108'	113'	5'	Porphyry. Foliation + Qtz.									
			Str. 11° to 10° to CA. Mod.									
			propylitic alt. Wk. carb.									
			alt. 0.5% py mainly in									
			qtz stringers									
95-18												
63'	68'	5'	Dark Green, Sil. And.	990	63'	68'	5'	127 23				
			Foliation + Qtz Stringers 25°									
			to 45° to CA. Strong propylitic									
			alt. 0.5% py									
96-1												
153'	214'	61'	153-163 Dk. gr. And. Porph.	991	153'	158'	5'	508 691				
			Str. foliation 15°-25° to CA. 0.5% py	992	158	163'	5'	692		Core quartered		
			Sil. Brx. Mod. prop. alt.	993	~~163~~	~~168~~	5'	695		previously.		
			178-183 As Above, 188-193 Str.	994	188'	193'	5'	696		Remainder		
			bleached. foliated 11° to CA.	995	209'	214'	5'	700		Taken.		
			209-214 Str. Bleached, fol 20° to CA.									

Appendix C

Statement of Costs

Statement of Costs

Consulting Geologist-(D. Dunn) 5 days @ $500/day + GST	$2,675.00
Geological Technician-(A. J. Muirhead) 1.5 days @ $250/day + GST	401.25
Transportation, Communications, and Food.	396.71
Assays: 45 core samples screened for metallics. Three fire assays + prep. 45 samples @ $49/sample + GST	2,362.61
Drafting	963.00
Printing:	150.00
15% Management Fee:	1,042.29
Total Costs:	**$7,990.56**



Appendix D

Statement of Qualifications

Statement of Qualifications

I, David St. Clair Dunn, Professional Geoscientist, with a business address at 1154 Marine Drive, Gibsons, British Columbia, Canada certify that:

1. I am a graduate of the University of British Columbia and hold a degree of Bachelor of Science in Geology.

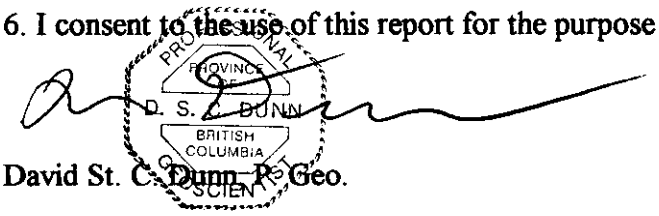
2. I am registered as a Professional Geoscientist with the Association of Professional Engineers and Geoscientists of British Columbia (Reg. # 18479).

3. I have practised my profession as a prospector and geologist for 33 years.

4. I have based my conclusions and recommendations in this report on a review of all available reports and direct supervision of the 2002 Check Sampling Program.

5. I have not directly or indirectly received, nor do I expect to receive, any interest in the subject property or in Avola Industries Inc.

6. I consent to the use of this report for the purpose of private or public financing.



David St. C. Dunn, P. Geo.

December, 2002

GEOLOGY & SYMBOL LEGEND *(after Sutherland Brown, 1989)*

POST PLUTONIC
- 17 Post plutonic basalt flows
- 16 Late andesite dyke
- 15 Late dacite dyke
- 14 Late felsite dyke
- 12 Early andesite dyke
- 11 Early feldspar dyke

PLUTONIC ROCKS
- 10 Aplitic granite
- 9 Granite
- 8 Mixed gneiss
- 7 Granodiorite
- 6 Porphyritic diorite
- 5 Hornblende quartz diorite
- 4 Hornblende diorite
- 3 Hybrid diorite

PRE PLUTONIC
- 2 Limestone
- 1 Greenstone

- geological contact
- shear fault
- showing
- 1997 drill hole
- proposed drill hole
- vein, showing dip
- creek
- logging road
- highway

contours are at 50m intervals

AREA OF PROPOSED GEOPHYSICAL SURVEY

Placer Dome 1989 soil grid

TEDI PIT

97-1 97-2 97-3 97-4 97-5

Martial Lake

DAVE'S POND

MAIN SHOWING

SILVER TUNNEL

QUARTZ TUNNEL

MCKENZIE SHOWINGS

ZINC SHOWING

Proposed

Cullachan Creek

Brandywine Creek

baseline 50E

baseline 55E

99

0 50 100 150 200 250 500
metres

La Rock Mining Corporation

BRANDYWINE PROPERTY
Vancouver Mining Division, B.C.

Property Geology Map

Scale	1:5,000	Date	December, 2002
N.T.S.	92J/3	By	d.s.d. / a.g.b.
Map	1		David Dunn P. Geo.

GEOLOGICAL SURVEY BRANCH
ASSESSMENT REPORT

27,116