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D.K. MARTIN & ASSOCIATES

Mining Development & Administration 4728 N. 21st Avenue Phoenix, Arizona 85015

27 April 1982

LPR Land Company Richard L. Lambert, President 555 South Woodward Avenue Suite 607 Birmingham, Michigan 48011

Dear Mr. Lambert:

Thank you for escorting Mr. Brown, geologist, and myself to your mining property located in Section 2, Township 4 South, Range 5 West, Maricopa County, Arizona, near Gila Bend, and requesting us to submit an evaluation proposal for your consideration.

Open pit mining differs from hard rock vein mining in that it is impossible to ascertain the exact content of precious metal in any one location merely by surface sampling and panning, due to the nature of the deposit and its mobility. Therefore, to attempt to calculate the extent of the deposit from these indications, could be very misleading.

It has been determined a drilling exploration program is necessary to prove the feasibility of operation and calculate reserves and a success ratio.

The geological and physical data is compiled, and targeted areas located to deliniate the extent of minable ore. These areas are then drilled to a depth of 50 feet or more and carefully sampled. The results are then analized and a projection as to the profitability determined.

No two open pit deposits have conditions which are identical and the problems faced by any mine operator change from day to day, even on the same property.

Such conditions require the geologist and engineer possess ingenuity, resourcefulness, expertise, and experience. As has been proven, all successful mining projects have utilized well known facts, and established scientific and geologic principles in their mining operations. It stands to reason, everything else being equal, the miner whose operations are based upon and conform to fundamental laws will make the best profit.

(continued)

(602) 246-9573

4/27/82 Page Two

We are able to accomplish the required evaluation through geological and engineering expertise. We propose to utilize Mr. Frank Clark, of Clark-Oliver Mining Company as the drilling contractor who has prior knowledge of the property. He is licensed, bonded, and performs an excellent job.

Our methods employ a step by step planned approach, of which only actual charges and expenses will be made.

To evaluate, block out ore, probable ore and ore in site, on your Gold Bug Claims, it is estimated will cost somewhere between \$28,916 and \$31,314.

For a breakdown of the estimated expenditures involved to properly evaluate the mining prospect, please refer to the attached Estimate of Expenditures.

Respectfully submitted,

. X. Martin

Douglas K. Martin

DKM:dm Encl: cc: D. Rose



D.K. MARTIN & ASSOCIATES

Mining Development & Administration 4728 N. 21st Avenue Phoenix, Arizona 85015

27 April 1982

GEOLOGICAL PROGRAM PROPOSAL

for

GOLD BUG MINING CLAIMS Maricopa County, Arizona

PROFESSIONAL SERVICES:

Geological and Engineering

7	-	8	days	field time @ \$350/day	\$2450 - \$2800
4	-	5	days	Office, data, reports,	
			maj	pping, sample prep, etc.	1400 - 1750

EXPENSES:

7 -	8 days @ \$65/day	455	-	520
850	- 900 miles @ \$.40/mile	340	-	360
100	+/- assays	750	-	1000

SUBCONTRACT DRILLING:

1900 to	2000 feet, 50 Mobilization	to 200' depth @ \$10/ft	19000 - 20000 750 - 800
CONTINGENCY	15%		3771 - 4084

ESTIMATED COST PROJECTION

\$28916 -\$31314

Note: Only actual charges and expenses will be made.

GEOCHEMICAL EXHALATION SURVEYS 602-994-9964

JOHN E. CHALLINOR SOIL GAS TECHNOLOGY 6360 E. ROSE CIRCLE DR. SCOTTSDALE, ARIZONA 85251

URANIUM OIL & GAS GEOTHERMAL & METALS

GOLD BUG MINING AND MINERALS CO. INC CYANIDE LEACH TEST

This metallurgical report is a preliminary evaluation of the cyanide leachability of a 450 pound gold ore sample provided by Lebern Cox on September 15, 1981. The ore appears as a purple to red colored, highly oxidized, fractured granitic rock. Hematite is abundant, filling cavities in the quartz, and probably represents former pyrite sites prior to oxidation. The gold was likely associated with the pyrite as free gold was readily recovered by panning a handful of hematitic fines.

The ore sample was first screened with a 3/8 inch screen and the plus 3/8 inch product was crushed in a 5×5 laboratory crusher and re-screened. The minus 3/8 product was called " Fine Ore ", and the plus 3/8 product, which graded up to 3/4 inch, was called " Coarse Ore ".

The fine and the coarse samples were then split with a Jones splitter to obtain a representative sample for assaying. The assay sample was then crushed to minus 1/8 inch and sent for commercial assaying by atomic absorption.

The prepared ore samples were now weighed and charged into seperate eight inch diameter PVC leaching columns. The fine ore column received 200 pounds, filling it to a height of 73 inches and the coarse ore column received 180 pounds, filling it to the same height. All water used in the tests was demineralized. Calcium and sodium hydroxide were added for pH control and the solution circulated until it stabilized before the cyanide was added. There was approxamately 15 pounds of excess leach solution in each column and an electric pump would cycle the solution to a spray at the top of the column each two hours according to an electric timer. After a prescribed number of leach cycles are accomplished, the columns are drained, the pregnant solution weighed, and assayed by a commercial assayer using atomic absorption. The results obtained by the subsequent filling, leaching, and draining of the leach solutions can be converted to approximate times and recoveries that may be expected from actual heap leaching.

Leaching data and assays for ore and solutions are attached as is a graphical representation of the gold recovery versus time for the first five leach tests. An evaluation of the entire test will follow the completion of testing.

John E. Challinor September 30, 1981

FINE ORE LEACH TEST

.

Sep	pt. 1981								
Tes	<u>st No. 1</u>	(8 c	ycles)						
17	Time	рН 8 2	NaCN gpl	Weighed 200 lbs -3/8 ore and 39.51bs of water and loaded fine ore column. Add 15g CaO	с У	s a	Gold troy	& Silver oz / ton	
	8:00p 9:00p	8.6	, - , ,	Add 15g CaO Add 15g CaO	1	p 1	OISC	olution.	
	11:00p	8.6	- ,	Add 15g CaO	s	e			
18	6:00a 7:30a	8.4	-	Add 30g NaOH Add 16 5g of NaCN					
	9:30a 12:00a	11.3	0.70	Add 10.5g Of Mach	1	1	0.093		
	3:30p 6:30p	11.5 11.3	0.67		4	3	. 24		
	8:30p 10:00p	11.2 11.0	0,65 0,62	Let the column drain overnight.	6 7				
	This ro	10,8	0.55	Recovered 14.0 lbs of preg, solution.	8	4	. 35	. 32	
	TUT2 16	covery	represe	ils 0.00245 oz Au; Or a total recovery	OI	12.25%	Irom	the ore	

Test No. 2 (16 cycles)

19	8:30a			Restart co	lumn wit	h add	ition	of 16.3					
				lbs of wat	er, 1g Na	aOH a	nd 10.1	2g NaCN					
	11:00a	11.1	0.95						2				
	3:30p	11.0	0.80						4	5	.26		
	11:00p	11.1	0.80						8	6	. 29		
20	8:45a	11.1	0.75						13				
	2:00p			Drained th	ne column	afte	r the 2	2 pm cycle	e				
		11.0	0,70	Recovered	14.0 lbs	of p	reg, so	olution	16	7	.33		27
	This re	ecovery	represe	nds 0.00231	. oz Au;	Or a	total	recovery	of	11.55%	from	the o	ore

Test No. 3 (32 cycles)

20	8:00p			Restart column with addition of 16.7				
				lbs of water, 2g NaOH and 10g NaCN.				
	10:00p	11.3	0.90		1			
21	11:45a	11.1	0.80		8	8	.22	
22	7:45a	11.1	0.75		18	9	. 24	
23	11:00a			Drained the column after 10am cycle.				
		10.9	0.55	Recovered 16.3 lbs of preg. solution	32	10	. 24	.32
	This re	covery	represe	nts 0.00196 oz Au; Or a total recovery	of	9,78%	from	the ore

Tests 1 thru 3 represent approximately 5 1/3 days of leaching for a total recovery of 33.58% of the gold in the sample, and 14.97% of the silver.

FINE ORE LEACH TEST

Sep	t. 1981							ii San sannar
Tes	t No. 4	(22 c	ycles)		с У С	s a m	Gold troy of so	& Silver oz / ton lution.
24	Time	рН	NaCN gpl	Restart column with addition of 16.7 lbs of water, $1\frac{1}{2}g$ NaOH and 11g NaCN.	1 e s	P 1 e		
25	7:45a 11:00p	10.9 10.9	0,80 0,75		10 18	11 12	.15 .15	. 22
26	6:00a 7:00a	10.8	0,75	Drained the column after 6 am cycle. Recovered 12.5 lbs of preg. solution	22	13	.15	. 23
	This rec	overy	represe	nts 0.00094 oz Au, or a total recovery	of <u>4.</u>	<u>69%</u> fi	rom the	ore.
		-						
Tes	t No. 5	(22 c	ycles)					
26	11:00a			Restart column with addition of 16.7				
27	1:55a 3:55p	10.9 10.9	0.85 0.80	IDS OF WALLET, F25 NACH and F15 Mach.	8 15	14 15	.093 .098	
28	5:55a 8:00a	10.9	0.75	Drained the column after 4 am cycle. Recovered 14.5 lbs of preg. solution	22	16	.10	.17
	This re	covery	repres	ents 0.00072 oz Au, or a total recovery	of <u>3</u>	.62%	from th	e ore,
Tes	t No. 6	(22 c	ycles)					
28	11:00a			Restart column with addition of 16.7				
29	11:00p 7:45a	10.7 10.7	0,80 0,80	and 50ml 3% Hydrogen Peroxide.	7 11	- 17	.072	
30	5:55a	10,7	0.75	Drained the column after 4 am cycle. Recovered 16 lbs of preg. solution.	22	18	.079	. 20
	This rec	covery	represe	nts 0.00063 oz Au, or a total recovery	of 3.	16% f	rom the	e ore,

Tests 1 thru 6 represent approximately 10 days of leaching for a total recovery of 45.06% of the gold in the sample, and 24.12% of the silver.

COARSE ORE LEACH TEST

Se	pt,	1981

Tes	t No. 1	(8 c)	cles)					
17	Time	pH	NaCN	Weighed 180 lbs +3/8 ore and 25 lbs	с	S	Gold &	Silver
		- ×	gpl	of water and loaded the ore column.	у	а	troy c	z/ ton
	6:00p	8.4	-	Add 10g CaO	С	m	of sol	ution.
	8:00p	11.1	-	Add 5g CaO	1	р		
	10:00p	11.0	-	P	е	1		
	12:00p	10.9	-		s	e		
18	6:00a	10.6	-					
	7:30a	10.6	-	Add 15g NaCN				
	9:30a	11.1	1.00		1	1	.032	
	12:00a	11.0	1.10		2	2	.053	
	3:30p	10.8	1.00		4	3	. 097	
	6:30p	10.8	1.00		5			
	8:30p	10.8	1.00		6			
	10:00p	10.8	0.95	Drained the column after 10 pm cycle.	7			
	•	10.6	0.80	Recovered 15.6 lbs of preg. solution	8	4	.13	1.4
	This re	ecovery	represe	nts 0.00101 oz Au; Or a total recovery	of	4.02%	from th	le ore

Test No. 2 (16 cycles)

19	8:30a			Restart column with addition of 15.8				
				lbs of water, 1g NaOH and 7.4g NaCN.				
	11:00a	10.8	0.90		2			
	3:30p	10,5	0.80		4	5	.075	
	11:00p	10.7	0.72		8	6	.094	
20	8:45a	10.7	0.65		13			
	2:00p			Drained the column after 2 pm cycle.				
	-	10.6	0.65	Recovered 15 lbs of preg. solution.	16	7	.13	.43
	This re	covery	represe	nts 0.000975 oz Au: Or a total recover	y of	3.87%	from	the ore

Test No. 3

20	8:00p			Restart column with addition of 16.7				
	-			lbs of water, 2g NaOH and 8g NaCN.				
	10:00p	11.5	0.90		1			
21	11:45a	10.9	0,75		8	8	.069	
22	7:45a	10.8	0,65		18	9	.088	
23	11:00a			Drained the column after 10 am cycle.				
		10.6	0.45	Recovered 16 lbs of preg. solution.	32	10	.11	.31
	This rec	overy	represe	nts 0,00088 oz Au; Or a total recovery	of	3.49%	from the	ore

Tests 1 thru 3 represent approximately 5 1/3 days of leaching for a total recovery of 11.38% of the gold in the sample, and 30.78% of the silver.

COARSE ORE LEACH TEST

Ser	t 1981				ċ	e	Cold & Silve	r
000						3	trou or / to	1
Too	t No /	(22 0			У	a		
res	L NO. 4	(22 0)	ycres)		c	m	of solution.	
					1	P		
					e	1		
24	Time	pН	NaCN	Restart column with addition of 16.7	S	е		
	1:00p			lbs of water, $1\frac{1}{2}g$ NaOH and 7g NaCN.				
25	7:45a	10.8	0.60		10	11	.061	
	11:00p	10.7	0,55		18	12	.072	
26	6:00a			Drained the column after 6 am cycle.				
	7:00a	10.6	0.50	Recovered 13 lbs of preg. solution.	22	13	076 22	
		19-32 (DA)						
	This rea	overy	represei	nts 0.000494 oz Au, or a total recovery	of <u>1</u>	L.96% f	from the ore.	
Tes	t No. 5	(22 cy	ycles)					
26	11:00a			Restart column with addition of 16.7				
27	1.55a	10.8	0 75	100 of water, 126 nuon and 06 nuon,	8	14	038	
~/	3.555	10.0	0.70		15	15	.050	
20	5.550	10,0	0.70	Desired the select of the large selection	13	10	,045	
28	5:55a			Drained the column after 4 am cycle.				
	8:00a	10.8	0,60	Recovered 13.8 lbs of preg. solution	22	16	.053 .19	
	This re	covery	represe	ents 0.000366 oz Au, or a total recover	y of	<u>1.45%</u>	from the ore.	

Tes	t No. 6	(22 c3	cles)					
28	11:00a			Restart column with addition of 16.7 lbs of water, 1½g NaOH and 8g NaCN				
	11:00p	10.7	0,70	and 50ml 3% Hydrogen Peroxide.	7	-		
29	7:45a	10.8	0.65		11	17	.030	
30	5:55a			Drained the column after 4 am cycle.				
		10.7	0.55	Recovered 16.3 lbs of preg. solution	22	18	.040	.095

This recovery represents 0.000326 oz Au, or a total recovery of 1.29% from the ore Tests 1 thru 6 represent approximately 10 days of leaching for a total recovery of 16.08% of the gold in the sample, and 37.28% of the silver.

817 West Madison · Phoenix, Arizona 85007 · Telephone 254-6181

For: Mr. John Challinor 6360 E. Rose Circle Drive Scottsdale, Arizona 85251

Date: September 18, 1981 Lab. No.: 3347

Sample: Solutions Marked: Lee Cox

Received: 9/18/81

Submitted by: Same

REPORT OF LABORATORY TESTS

troy oz/ton of solution

Samp	le Marked	Gold
# 1	Fine Coarse	0.093 0.032
#2	Fine Coarse	0.14 0.053
#3	Fine Coarse	0.24

Respectfully submitted,

ARIZONA TESTING LABORATORIES

Robert J. Drake

817 West Madison · Phoenix, Arizona 85007 · Telephone 254-6181

For: Mr. John Challinor 6360 East Rose Circle Drive Scottsdale, Arizona 85251

September 21, 1981 3353 Lab. No.:

Solutions Sample:

Marked: See Below Lee Cox project

Date:

9-21-81 Received:

Submitted by: Mr. Challinor

REPORT OF LABORATORY TESTS

Samples Marked	troy oz/ton	of solution
	Gold	Silver
Cox Lot No. 1:		
4C 4F	0.13 0.35	1.4 0.32
5C 5F	0.075 0.26	
6C 6F	0.094 0.29	
7C 7F	0.13 0.33	0.43 0.27

Respectfully submitted,

ARIZONA TESTING LABORATORIES

Claude E. McLean, Jr.

817 West Madison · Phoenix, Arizona 85007 · Telephone 254-6181

For: Mr. John Challinor 6360 East Rose Circle Drive Scottsdale, Arizona 85251

Date: September 24, 1981 Lab. No.: 3410

Sample: Solutions

Marked: See Below

Received: 9/24/81

Submitted by: Same

REPORT OF LABORATORY TESTS

	troy oz/t	<u>on of solution</u>
Sample Marked	Gold	Silver
8F	0.22	
80	0.069	
9 F	0.24	
9 C	0.088	
10F	0.24	0.32
10C	0.11	0.31

Respectfully submitted,

ARIZONA TESTING LABORATORIES

isilte Cland

817 West Madison · Phoenix, Arizona 85007 · Telephone 254-6181

Date: September 28, 1981 For: Mr. John Challinor 6360 East Rose Circle Drive Scottsdale, Arizona 85251 Lab. No.: 3452

Sample: Solution

Marked: Cox Lot #1

Received: 9/28/81

Submitted by: Same

REPORT OF LABORATORY TESTS

troy oz/ton of solution

Sample Marked	Gold
llC	0.061
llF	0.15
14C	0.038
14F	0.093
15C 15F	0.045

Respectfully submitted,

ARIZONA TESTING LABORATORIES

MI Fear aul

817 West Madison · Phoenix, Arizona 85007 · Telephone 254-6181

For: Mr. John Challinor 6360 East Rose Circle Drive Scottsdale, Arizona 85251 Date: September 28, 1981 Lab. No.: 3453

Sample: Solutions

Marked: Cox Lot #1

Received: 9/28/81

Submitted by: Same

REPORT OF LABORATORY TESTS

	troy oz/ton of	solution
Sample Marked	Gold	Silver
12C 12F	0.072	0.29
13C 13F	0.076	0.22
16C 16F	0.053 0.10	0.19 0.17

Respectfully submitted,

ARIZONA TESTING LABORATORIES

817 West Madison · Phoenix, Arizona 85007

Telephone 254-6181

September 30, 1981 Mr. John Challinor Date: For: 6360 East Rose Circle Drive Lab. No.: 3491 Scottsdale, Arizona 85251

Solutions Sample:

Marked:

Cox Lot 1

9-30-81 Received:

Submitted by: Mr. Challinor

REPORT OF LABORATORY TESTS

Samples	Marked	troy oz/ton	of solution
		Gold	Silver
Cox Lot	1:		
	17-F 17-C	0.072 0.030	
	18-F 18-C	0.079 0.040	0.20 0.095

Respectfully submitted,

BORATORIES ARIZONA TESTING L



GEOCHEMICAL EXHALATION SURVEYS 602-994-9964

JOHN E. CHALLINOR SOIL GAS TECHNOLOGY 6360 E. ROSE CIRCLE DR. SCOTTSDALE, ARIZONA 85251

URANIUM OIL & GAS GEOTHERMAL & METALS

CYANIDE LEACH EVALUATION GOLD ORE SAMPLE LOT No 1

After 122 cycles of leaching, covering a span of approximately ten days, 45 percent of the gold in the fine portion of the sample was recovered as compared to 16 percent for the coarse portion of the sample. Projecting the lines on the recovery graph for another ten days, it is possible the fine ore recovery may reach 60 percent, while the coarse will only reach about 20 percent.

Except for the limiting factors that size of the ore particles have in relation to total recovery, the rest of the leaching parameters are excellent.

Reagent consumption:

The ore is only mildly acid, requiring 3-4 lbs of lime per ton of ore, therefore the use of lime should pose no problem to percolation and is very inexpensive. Cyanide consumption was 0.9 lbs per ton of ore for the 10 day leach period, which is reasonable.

Porosity and percolation:

Water, equal to 20% of the weight of the sample was applied. After two hours, 13% was entrained within the sample and 7% free liquid was recovered. No clay or slimes were apparent in the sample and the water applied to the sample, flushed thru with no detectable restriction. The leach cycle used in the testing was equal to 2½ gallons of leach solution per square foot, per hour, or 200 gallons per day per ton of ore.

Recommendation

Considering the fact that the percent of recovery seems to be related to particle size, the finer the crushing, the higher the recovery, it is important to know how fine the rock can be economically crushed prior to leaching, and what the increase in recovery would be. However, if crushing to much finer than -3/8 inch is feasible, then a form of agglomeration will probably be required to give the leach pile the rigidity to stand and not slump when wetted. A form of 'gunniting the sloping surfaces before wetting would accomplish the same purpose.

> John E. Challinor October 1, 1981.

817 West Madison · Phoenix, Arizona 85007 · Telephone 254-6181

For Mr. John Challinor 6360 East Rose Circle Drive Scottsdale, Arizona 85251

Date September 23, 1981

LAB NO.		OZ. PER TON		PERCENTAGES			
		GOLD	SILVER	COPPER	Iron	Manganese	
3371	Lee Cox Coarse Ore Lee Cox Fine Ore	GOLD 0.28 0.20	SILVER 0.50 0.40	COPPER 0.008 0.009	Iron 3.6 3.5	Manganese 0.11 0.12	

ASSAY CERTIFICATE

Respectfully submitted, ARIZONA TESTING LABOR

ham bert 500 < > 1 A . 0 s . . • 300 0 Ø 0 3 V V 28 holes @ 100' = 2800' = 4.28,000 + moving day 500 × 400 ×100 = 20,000,000 = 1,428.571 Tons (146 ++ .Ton)

Milk Creek Placer 550 Y 600 X 15' = 4,950,000 cm ft = 183,333 cm yds 600 X 27 9166.6 200/183, 3.33 @ 200 y 1s/d= 916 days supply gravel. Lambert -600 . ٠ 8 8 ø 300 . 0 V 20 holest 100 = 2000' 660× 300 × 100' = 18,000,000 Guff = 1,285,714 Time 26 hold - 2600 600× HOO × 100 - 24,000,000 " - 1, 714,286 " 33 holes = 3300' = 33000

11:30 Am 4-26-82 Dong: Dick Lombert - Phoned from Detroit 313-644-8973 Claims are located in Sections 2, 3, 10 \$ 11, T45, R5W 92 Claims Length 13,570' Width 5960' - Goes over mountain m N. + 5 of June Did not know location of mine. We Can probably get it of Ho Topo Sheet. Need Citrus Valley East (1/2 min) RSW 6 5 4 3 2 1 T 7 8 9 10 11 12 4

Lowbert

1800' - 7-8 days of drilling - I day hove in & 1 day out = 8-9 days Geol/Eng - 5-6 dury + exp. in Field 18000 22 days office - Report & Recommendations 2800 7-8 day @ 350 = \$ 2450 - 2800 640 200 80/day Subaritaria = 560 - 640 21,640 24, 124e = 900 mi e.40 = 360 500 Maring Rig 22,140 Contregency 15% 3321 25,461 (Report - 2 day @ 350 700 Shit Sampler - 1/2 day - 3525 \$4025+ Eng-- Geolo [7-8 duy @ 350 = = 2450-2800) A552118 - 100 @ = 750-1000 Milage - 900 @ 40 = 360 2000' drilling @ 1000 = 20,000 50 to 200' holes Morring Nig 200 mi @ 2.75 - 550

Sections, 2, 3 10 + 11 745 R5W, 11:30 Lambert - Detroit 4-26-82 hugth d Witth 5960' 13,570 92 clany 313-644-8973 Did hot Know bocs ties of humo 8-5-W. 5 K 3 2 1 T 4 > 8 9 10 11 17 5 350 350 175 25 1400 350 325 2.75 40 900 35000 18 4 3600 30 200 M 100 2800 2450 3675-

1 55 19 20 94 2RT - 400 kule So mili day 6B-hoc 400 miles

W 26 april 1982 Geological Program Progozal N for Goldbog Mining claune Sect 2TH Sohow, Marcopa Couty lengona Bolemonal Servicesi Geological & Engineering 7-8 days Field time @ 350/day 2450 - 2800 4-5 days Office - data Repute, Mappy Sample prep & delacy 1400 - 1750 Expenses 7-8 days @ \$65/day 455 - 520 850-900 miles @ . 40/min 340-360 1005 Cenargs 750-1000 Sub totals 5395 6430 Subcontract Drelling 1900 to 2000 feet - 50 to 200 day @ 10/1 19,000 - 20,000 molelization 750 - 800 25145 27230 Contengency 16% 3771 4087 28816 29953 31314.

Lambert Base Plan - Spacing & huber of holes Subject to Change Holes 50 to 200' 100 ð ø Depth & Spacing lasy change based on Results white drilling 400 (300 × 100 - 12,000,000 = 857,143 Tous 1. 18tholes = 2000" = 20,000 + moving chig. Note 2 holes + 200' deep 2000' = 20,000 / 2000' = 20,000 / (No Geal/Eng services or expenses added) 2, 400 × 400 ×100 = 16,000,000 = 1,142,857 Tays 23 holes = 2300' = 23000 1 more thy.

100 cloum hunte beelogiet Thur lay DAVID ROSE 313-332-3131 Sell all a Part 2 veins 1/2 02 - . 75 03 water well 500' 12t 300' Material - 35 from Main Vein - OXIDE 955-4972 Ebene Richard Kambert 1648 fockridge Bloomfield Ill 48013 Ux failent as the 5215 No 24th ST \$204 85016



DEAR RESIDENT:



PLEASE CALL FOR FURTHER INFORMA-TION.

INSPECTOR PHONE #

ADDITIONAL INFORMATION ON REVERSE