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INPUT DATA FOR RADIOLOGICAL ASSESSMENT OF THE DAWN MINING COMPANY URANIUM MILL

March, 1981

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I. Principal operating parameter values to be used in radiological assessment of the Dawn Mining Company uranium mill

Par	Parameter					
A.	General Data					
	America are grade * 11-0-	0.152				
	Average off grade, a cycy	632 0				
	Ore precasting rate MT/d	430.0				
	Dava / an anarational	1450.0				
	bays/yr operacional					
В.	Ore Storage Pile(s)					
	Actual area, acres	13.6				
	Annual average dust loss rate, g/m ² -vr	42.0				
	Dust/ore activity ratio	2.5				
	Reduction factor for chemical spraving and wetting. %	25%				
	Release rate for truck durbing and one bad activities %	1.8×10^{-5}				
	interest the second despand the one part deserver, it					
	Specific Rn-222 flux from ore piles, pCi/m-sec pCi/gm Ra-226	1.0				
с.	Hoppers and Feeders					
	Release rate for ore feeding. 7	0.01				
	Dust/ore activity ratio	2.5				
	Reduction factor for dust control. "	25%				
	Fraction of Rn-222 equilibrium ore content released.	20.0				
D.	Crushing and Grinding					
	Release rate for crushing and grinding, %	. 0.008				
	Efficiency of particulate loss control. %	99.5				
	Eust/ore activity ratio	2.5				
	Fraction of Rn-222 equilibrium ore content released. 5	20.0				
Ξ.	Fine Ore Storage					
	Release rate for fine ore activities. %	0.004				
	Efficiency of particulate loss control. %	80.0				
	Dust/ore activity ratio	2.5				
	Fraction of Rn-222 equilibrium ore content released.	20.0				
Ξ.	Yellowcake Drying					
	Yellowcake production rate, MT year	214.6				
	Release rate of vellowcake product to atmosphere. "	2.26 x 10 ⁻⁵				
	Yellowcake fraction of U-238. "	94.0				
	Yellowcake fraction of Th-230. Z	0.5				
	Yellowcake fraction of Pb-210 %	0.1				
	Vallowake fraction of Ba-276 "	0.1				
	terranduc from the second of	70.0				

Ra-226

Pb-210

Value(s) Parameter G. Yellowcake Packaging 1.62 x 10⁻⁵ Release rate of yellowcake to atmosphere, Z 94.0 Tellowcake fraction of U-238, % 0.5 Yellowcake fraction of Th-230, % 0.1 Tellowcake fraction of Pb-210, % Yellowcake fraction of Ra-226, % 0.1 H. Tailings Impoundment System General Parameters Fraction U to tailings, % 6.0 to 9.2 95.0 Fraction Th to tailings, % 99.3 Fraction Ra and Pb to tailings, % Annual average dust loss rate, g/m²-yr 420.0 2.5 Dust/tails activity ratio Dusting reduction factor for water cover, moisture, 80.03 and chemical agents, % Specific Rn-222 flux from exposed beach, pC1/m²-sec Rn-222 1.0 pCi/ Ra-225 Abandoned . lings impoundment 59.2 Total area acres 1.0-3.0 Area exposed to dusting, % Tailings activities, pCi gm 62.1 U-238 641.2 Th-230 673.6 Ra-226 673.6 Pb-210 Inactive Tailings Impoundment 47.0 Total area, acres Area exposed to dusting, % 0.5 to 10.0 Reduction factor of dust loss by chemical 50.0 stabilization and watting, % Tailings activities, pCi/gm 40.5 U-238 Th-230 641.2 673.6 Ra-226 673.6 Pb-210 New tailings impoundment 27.2 Total area, acres Reduction factor of dust loss by water cover, chemical stabilization, % 0.2 Tailings activities, pCi/gm 26.0 U-238 410.4 Th-230

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SUMMARY OF SOUTCE TERMS

<u>110.</u>	kn 	k v	а 2	kn ² Area	<u><u>v-235</u></u>	<u>Th-230</u>	CI/year Ra-226	
11004567	0.18 0.08 0.08 0.08 0.00 -0.01 0.18	0.00 0.07 0.06 0.00 0.00 0.00 0.00	-3.00 -3.00 12.00 -2.00 20.00 11.00 -3.00	0.0550 0.0000 0.0000 0.0000 0.0000 0.0000 0.0550	2.895-03 1.018-03 6.435-05 1.295-03 1.085-03 7.765-04 5.245-04	2.39E-03 1.012-03 6.43E-05 1.29E-03 5.40E-06 3.88E-06 6.24E-04	2.89E-03 1.01E-03 6.43E-05 1.29E-03 1.08E-06 7.76E-07 6.24E-04	2.1.6.1.1.6.
8 9 10 11 12	-0.39 -0.87 -0.79 -0.91 -1.00	+0.52 -0.40 -0.62 -0.81 -0.86	3.30 1.00 . 1.00 6.00 6.00	0.0072 0.0950 0.0950 0.0410 0.0690	4.695-04 4.042-03 4.042-03 1.125-03 1.882-03	4.85E-03 6.40E-02 6.40E-02 1.77E-02 2.97E-02	5.092-03 6.722-02 6.722-02 1.882-02 3.122-02	5 6 0 H C

INPUT TAILS ACTIVITIES, PCI/G

SET	URANIUM	THORIUM.	RADIUM	LEAD
1	43.2	43.2	43.2	43.2
2	26.0	410.4	431.2	431.2
3	62.1	f'1.2	673.6	673.6
4	40.5	641.2	673.6	673.6

		P Size	M/sec	
-	<u>R=-222</u>	Set	Exit Vel	Source Name
-03	0.	3	0.	Ore Pad Activities
-73	1.292+01	2	0.	Hoppers + Feeders
- 25	1.282+01	2	20.	Crushers + Grinders
- 73	1.293+01	2	0.	Fine Cre Storage
-25	0.	1	4.6	Yellowcake Drver
-27	0.	1	11.	Yeilowcake Packaging
-24	7.485+02	3	0.	Ore Pad Windage
- 23	5.093+03	3	0.	Abandoned Tailings
12	2.028+03	3	0.	Inactive Tailings 1
123	2.023+03	3	0.	Inaccive Tailings 2
02	0.91E+01	3	0.	Present Tailings 1
-32	6.57E+01	3	0.	Present Tailings 2

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WAFTICLE SIZES AND FRACTIONAL DISTRIBUTION

 1.0	1.0	5.0	35.0	PDEN
1.000	0	0	0	8.900
0	1.000	0	0	2.400
0	0	0.300	0.700	2.400

	Particulate Source Strength Multipliers						
		by Time Step					
Source No	T Step 1 20 yrs	T Step 2 3 yrs	T Step 3 7 yrs	T Step 4 3 yrs			
1	1.	1.	1.	0.			
2	1.	1.	1.	0.			
3	1.	1.	1.	0.			
4	1.	1.	1.	0.			
5	1.	1.	1.	0.			
6	1.	1.	1.	ο.			
7	0.25	0.25	0.25	0.25			
8	1.	1.	0.333	0.333			
9	0.05	0.01	0.0025	0.0025			
10	0.05	0.01	0.0025	0.0025			
11	0.	0.002	0.002	1.			
12	0.	0.002	0.002	1.			

Radon Source Strength Multipliers by Time Step

Source No.	T Step 1 20 yrs	T Step 2 3 yrs	T Step 3 7 yrs	T Step 4 3 yrs
1	1.	1.	1.	0.
2	1.	1.	1.	0.
3	1.	1.	1.	0.
4	1.	1.	1.	0.
5	1.	1.	1.	0.
6	1.	1.	1.	0. '
7	1.	1.	1.	0.
8	1.	1.	1.	1.
9	0.9	1.	1.	1.
10	0.9	1.	1.	1.
11	0.	0.8	0.8	1.
12	0.	0.8	0.8	1.

II. Source Term Calculations

Source No. 1 - Ore Pad Activities

· Dust Emission from Truck Dumping

= .01 kg/MT ore

· Dust from Front End Loader Operation

The reference gives .025 kg/MT for loading ore into a truck. But clearly most of the dust must be generated when the load is dropped. In the present case, discharge is not into a truck, but into the hopper of the crusher building, which is described as Source No. 2. Therefore, it is reasonable to assume 20% of the above figure is emitted on the Ore Pad. or .005 kg/MT ore.

Dust from Bulldozer Operation

No emission values are reported, so an estimate will be made from the .025 kg/MT value reported above for front end loader operation. The dozer handles only 200 MT/day, compared to 430 MT/day for the loader. Also, the dozer is used only 6 months/ yr, compared to 12 for the loader. Finally, we use a reduction factor of 2 because the material is not dropped. Therefore, we estimate:

.025 $\frac{\text{kg}}{\text{MT}} \times \frac{200}{430} \times \frac{6}{12} \times \frac{1}{2} = .003 \text{ kg/M}$

Total from Machinery on Ore Pad =

.01 + .005 + .003 = .018 kg/MT ore

Ore Processing Rate, MT/d = 430.0

Days/yr operational = 346

Ore Processed, MT/yr = 148,780

Total Dust from Machinery on Ore Pad =

.018 $\frac{\text{kg}}{\text{MT}} \times 148,780 \frac{\text{MT}}{\text{y}} = 2678 \text{ kg/y}$

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Ore activity, pCi/g = 432.0

Dust/ore activity ratio = 2.5

· · Source Value, Ci/y =

= $(432.0)(2.5)(2.678 \times 10^6)(10^{-12} \text{ Ci/pCi}) = 2.89 \times 10^{-3} \frac{\text{Ci}}{\text{Ci}}$

(Ore Pad Windage is not included here. It is presented as Source No. 7. Radon from the ore pad is included with Source No. 7.)

Source No. 2 - Hoppers and Feeders

Front End Loader deposits ore through a grizzly over a hopper which is a part of the dust-controlled crusher building. If the loader were dumping into an open truck, an emission rate of .025 kg/MT ore would be appropriate. Because the material is dumped into a beghour equipped building, a reduction of 50% is justified; also, because 6 months of the operation takes place under very wet, saturated conditions, another 50% reduction should be applied. Thus, dust emissions are:

 $.025 \frac{\text{kg}}{\text{MT}} \times .5 \times .5 = .0063 \text{ kg/MT}$ ore

.. and Source Value = .0063 $\frac{\text{kg}}{\text{MT}} \times 148,780 \frac{\text{MT}}{\text{y}} \times 1080 \frac{\text{pCi}}{\text{g}} \times 10^{-12} \frac{\text{Ci}}{\text{pCi}}$ $\times 10^3 \frac{\text{g}}{\text{kg}} = 1.01 \times 10^{-3} \frac{\text{Ci}}{\text{y}}$

Radon in this source:

Estimate 20% of the secular equilibrium content is released:

· · Rn Source Value =

148,780 $\frac{\text{MT}}{\text{y}} \times 10^6 \frac{\text{g}}{\text{MT}} \times 10^{-12} \frac{\text{Ci}}{\text{pCi}} \times .2 \times 432 \frac{\text{pCi}}{\text{g}} = 12.9 \frac{\text{Ci}}{\text{y}}$

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Refs 2

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Source No. 3 - Crushing and Grinding

Dust collection efficiency of a properly function g baghense exceeds 99.5%. The NRC staff downgraded estimated performance for purposes of this source calculation to 99%, to account for downtime and other routine losses of efficiency. However, Dawn's crusher plant baghouse, rated at 12,500 cfm, was replaced in 1980 by a new Fabri-Pulse collector rated at 18,000 cfm. Furthermore, routine procedure is to shut down the crusher operation when baghouse maintenance is required. Under these circumstances, Dawn believes the 99.5% value is justified.

Based on Sears, et al, we estimate .008% of the ore crushed becomes dust to the exhaust air ventilation flow. Actually, this is conservative because Dawn ore may be expected to average more than 6% moisture (upon which Sears' value is based).

The release rate to the atmosphere is then:

= 148,780
$$\frac{\text{MT}}{\text{y}} \times 8 \times 10^{-5} \times (\frac{100 - 99.5}{100}) = .0595 \frac{\text{MT}}{\text{y}}$$

.. and the Source Term Value = .0595 $\frac{MT}{y} \times 1080 \frac{pC1}{g} \times 10^6 \frac{g}{MT} \times 10^{-12} \frac{C1}{pC}$

 $= 6.43 \times 10^{-5} \text{ Ci/v}$

Grinding is done wet, and makes no significant contribution to this source term.

Radon from Source No. 3 is taken as the same value as from Source No. 2, for the same reasons, with the minor difference that the computed value in that case, 12.85, was rounded to 12.9; in order to prevent accumulation of rounding error, it is now taken as 12.8 Ci/y.

Stack exit velocity

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Ventilation exhaust rate = 18,000 cfm.

- Stack cross section = 3.69 ft^2

Nominal exit velocity, m/sec =

= 18,000 $\frac{ft^3}{min} \times \frac{min}{60 \text{ sec}} \times \frac{m}{3.2808 \text{ ft}} \times \frac{1}{3.69 \text{ ft}^2}$

= 24.8 m/sec

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Measured exit velocity = 19.8 - 20.7 m/sec

.. Value selected for MILDOS calculation = 20 m/sec

Source No. 4 - Fine Ore Storage

For totally enclosed conveyor transfer points, uncontrolled emission factor employed by Colorado Air Pollution Control Division, Dept. of Health, is .02 lb/ton (.01 kg/MT) for each transfer point.

Dawn uses 2 transfer points in and 2 out, of dry crushed ore. Therefore, emission factor should be $4 \times .01 = .04 \frac{\text{kg dust}}{\text{MT ore}}$

Efficiency of control of these emissions is estimated to be 80%, probably a conservative value, since all of the transfer points are inside the mill building, in a partitioned-off section of it.

Quantity of Dust Emitted = $.04 \frac{\text{kg}}{\text{MT}} \times 148,780 \frac{\text{MT}}{\text{y}} \times .2$

.. Source value, Ci/y = 1080 $\frac{pCi}{g} \times 1.190 \times 10^{6} \frac{g}{y} \times 10^{-12} \frac{Ci}{pCi}$ = 1.29 x 10⁻³

Again, the radon source term is estimated to be 12.9 Ci/y for reasons given for above unit operations.

Source No. 5 = Yellowcake Drying

(Tellowcake Packaging is treated separately, as Source No. 6)

Recovery rate of $U_30_8 = 94.0\%$

Y.C. Production Rate =

148,780 MT x .153% x 94.0% = 214.0 MT

where 0.153% is the average ore grade.

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Operating Time = 298
$$\frac{days}{yr} \approx 24 \frac{h}{day} = 7152 \frac{h}{yr}$$
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Hourly Production Rate =

 $=\frac{214.0 \text{ MT/y}}{7152 \text{ h/y}} = .030 \text{ MT/h}$

Particulate emission rate, measured = $6.79 \times 10^{-4} \text{ kg/b}$

Dryer release fraction = $= \frac{6.79 \times 10^{-4} \text{ kg/h}}{.030 \times 10^{3} \text{ kg/h}} = 2.26 \times 10^{-5}$

Y.C. annual release from dryer = = 214.0 $\frac{\text{MT}}{\text{y}} \ge 2.26 \ge 10^{-5}$ = 4.84 $\ge 10^{-3} \frac{\text{MT}}{\text{y}}$

U-238 particulate source term = = 4.84 x 10⁻³ $\frac{MT}{v}$ x 10⁶ $\frac{2}{MT}$ x 3.33 x 10⁻⁷ $\frac{Ci}{e U-238}$ x .85 x .79 .. = 1.08 x 10⁻³ Ci/y

> where .85 is the fractional U content of U30g and .79 is the reported Y.C. purity.

- .. Th-230 source term = 0.5% of U-238 source term = $1.08 \times 10^{-3} \times .005 = 5.40 \times 10^{-6} \text{ Ci/y}$
- 5 .. Ra-226 source term = 0.1% of U-238 source term $= 1.08 \times 10^{-3} \times .001 = 1.08 \times 10^{-6}$. Ci/y
- •• Pb-210 source term = Same as Ra-226 = 1.08 x 10⁻⁶ Ci/y

Stack Height = 20 meters

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.. Stack Exit Velocity = 15 fps = 4.6 m/sec

•• Radon source term = 0

Source No. 6 - Yellowcake Packaging

Hourly Y.C. Production Rata = .030 MT/h

Particulate emission rate, measured =

 $= 4.85 \times 10^{-4} \text{ kg/h}$

Packaging release fraction =

- $= \frac{4.85 \times 10^{-4} \text{ kg/h}}{.030 \times 10^{3} \text{ kg/h}} = 1.62 \times 10^{-5}$
- Y.C. Annual release from packaging stack = = 214.0 $\frac{\text{MT}}{\text{y}} \ge 1.62 \ge 10^{-5} = 3.47 \ge 10^{-3} \frac{\text{MT}}{\text{y}}$

·· U-238 particulate source term =

= $3.47 \times 10^{-3} \frac{\text{MT}}{\text{y}} \times 10^{6} \frac{\text{g}}{\text{MT}} \times 3.33 \times 10^{-7} \frac{\text{Ci}}{\text{g} \text{ U}-238} \times .85 \times .79$

= $7.76 \times 10^{-4} \text{ Ci/y}$

where .85 is the fractional U content of U_3O_8 and .79 is the reported Y.C. purity,

- Th-230 source term = 0.5% of U-238 value = 7.76 x 10⁻⁴ x .005 = 3.88 x 10⁻⁶ Ci/y
- •• Ra-226 source term = 0.1% of U-238 value, = 7.76 x 10⁻⁴ x .001 = 7.76 x 10⁻⁷ Ci/y
- Pb-210 source term = same as Ra-226 = 7.76 x 10⁻⁷ Ci/y Stack Height = 11 meters

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. Stack Exit Velocity = 36.3 fps = 11 meters/sec

·· Radon source term = 0

Source No. 7 - Ore Pad Windage

Dust emission rate, uncontrolled	=	42.0 g/m ² /yr	2 (10% of
Pad Area	-	55,000 m ²	tailings dust loss
Ore Activity	=	432.0 pCi/g	rate)
Dust/Ore activity ratio	=	2.5	

Control factor due to snow cover and precipitation, 6 months, is 100%

Control factor due to wetting down the pad, during the other 6 months, is 50%

Average annual control = 75%, or .. Average annual control factor (Particulate Source Strength Multiplier, all Timesteps) = 0.25

.. Source value (without control)

$$42.0 \frac{g}{m^2 - y} \ge 55,000 \text{ m}^2 \ge 432 \frac{\text{pCi}}{g} \ge 10^{-12} \frac{\text{Ci}}{\text{pCi}} \ge 2.5$$
$$= 2.49 \ge 10^{-3} \text{ Ci/yr}$$

Radon release from ore stored on pad:

Release rate = 1.0
$$\frac{pCi/m^2 - sec Rn - 222}{pCi/gm Ra - 226}$$

·· Radon source term =

 $1.0 \times 432 \frac{pCi}{g} \times 3.156 \times 10^7 \frac{sec}{yr} \times 10^{-12} \frac{Ci}{pCi} \times 55,000 \text{ m}^2 =$

= 748 Ci/yr

Source No. 8 - Abandoned Tailings Area

MILDOS limits tailings areas, for calculation purposes, to 0.1 km². The actual size of the abandoned tailings area at Dawn is 59.2 acres = 0.24 km². However, this area is presently covered by a layer of woodchips nominally 2 ft thick. A conservative estimate of the effectiveness of this cover, at present, is 97%, so that

Area exposed to dusting = 3% of .24 km² = .0072 km²

Since this is less than 0.1 km², there is no need to subdivide it into multiple sources.

When, in 1985, the permanent thick cover is in place, a very conservative estimate of its effectiveness is taken to be 99%, so that only 1% is exposed. We will retain the .0072 km² area in the calculation, and account for the threefold future decrease in area exposed to dusting by means of the part culate source strength multiplier, which varies with timestep.

We will assign the timesteps as follows:

Timestep	1	Past	Twenty 1	lears		
Timestep	2	1981	- 1984 .	Three	Years	(To Final Cover on Source No. 8)
Timestep	3	1985	- 1991	Seven	Years	(To End of Operation)
Timestep	4	1992	- 1994	Three	Years	(Drying Time for Sources Nos. 11, 12)

For Source No. 8, the values of particulate source strength multipliers are:

Timestep	1	1.000	
	2	1.000	
	3	3.333 x 1	0-1
	4	3.333 x 1	0-1

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44	à. à.	-	133	61 m	1.1	-	123.	Plum'	8

U -238	62.1
Th-230	641.2
Ra-226	673.6
Pb-210	673.6

Fraction U to tailings = 9.2% Fraction Th to tailings = 95.0% Fraction Ra and Pb to tailings = 99.8%

Annual average dust loss rate (from uncontrolled tailings)

Dusting reduction factor for water cover, moisture, chemical agents = 80.0%

Dust/Tails activity ratio = 2

Source Terms (particulates), based on present area exposed to dust erosion, before dust control factors are applied:

 $U-238 = (420 \frac{g}{m^2-y})(62.1 \frac{pCi}{g})(2.5)(.0072 \text{ km}^2)(10^{-6}) = 4.69 \times 10^{-4} \frac{Ci}{yr}$

where 2.5 is the Dust/Ore Activity Ratio and 10⁻⁶ is a conversion factor = $\frac{Ci}{pCi} \times \frac{m^2}{km^2}$

 $Th-230 = (420)(641.2)(2.5)(.0072)(10^{-6}) = 4.85 \times 10^{-3} Ci/y$ Ra-226 = (420)(673.6)(2.5)(.0072)(10^{-6}) = 5.09 \times 10^{-3} Ci/y

$$Pb-210 = 5.09 \times 10^{-3} Ci/y$$
 (Same as Ra-226)

Coordinates x, y, z of Source No. 8 were determined by dividing it into three sub-areas of simple geometric shape (rectangles, triangles), locating the center (centroid) of each, and calculating the coordinates by

$$\overline{x} = \frac{\overline{z} \times_i A_i}{\overline{z} A_i} \qquad \overline{y} = \frac{\overline{z} y_i A_i}{\overline{z} A_i} \qquad \overline{z} = \frac{\overline{z} z_i A_i}{\overline{z} A_i}$$

Coordinate values are:

$$x = -0.39 \text{ km}$$

$$y = -0.52 \text{ km}$$

$$z = +3.3 \text{ km}$$
Specific Radon flux = $1.0 \frac{\text{pCi/m}^2 - \sec \text{Rn} - 222}{\text{pCi/g Ra} - 226}$
Area for Rn calculation = $59.2 \text{ acres} = 2.396 \times 10^5 \text{ m}^2$
Radon Source Term =
$$= (1.0) (673.6 \frac{\text{pCi}}{\text{g}}) (2.396 \times 10^5 \text{ m}^2) (3.156 \times 10^7 \frac{\sec}{\text{y}}) (10^{-12} \text{ Ci/pCi})$$

$$= 5.09 \times 10^3 \frac{\text{Ci}}{\text{y}}$$

Source No. 9 - Inactive Tailings I

Preparation of a new subgrade tailings disposal area is now underway, and it is contemplated that within a few months, it will be in use. The disposal area previously in use (designated as "Present Pile Area" in the August 1980 Radiological Assessment performed in the course of obtaining a license amendment authorizing the new disposal area) will be allowed to dry, and as it does, the exposed surfaces will be covered <u>initially</u> with a onefoot layer of earth, or by a chemical stabilization technique, with the objective of controlling dust emission. The permanent cover will be designed in accordance with the NRC Mill Licensing Requirements (FR October 3, 1980, 65521-65538), Appendix A, and its counterpart in the regulations of the State of Washington, or whatever revisions thereto may be in effect.

Since the disposal area used recently may no longer be in use at the time the Dawn license renewal application is reviewed, we have changed its name, in the interest of clarity, from Present Tailings Area to <u>Inactive Tailings Area</u>. Similarly, we have changed the designation of what, in the August 1980 assessment was called Future Pit Area to New Tailings Area. 14

Since the Inactive Tailings Area exceeds 0.1 km², we have arbitrarily subdivided it for MILDOS calculation purposes into two parts, locating a curter and representative elevation for each. As with Source No. 8, the calculation of coordinates is based on ratios of $\Sigma \propto_i A_i/\Sigma A_i$, where the \propto_i (or γ_i or z_i) are taken from simple geometric shapes into which the overall area has been divided.

For Source No. 9 (Inactive Tailings I) Area = 0.095 km²

 $x = -0.87 \text{ km}^2$ $y = -0.40 \text{ km}^2$ $z = \pm 1.0 \text{ m}$

Efficiency of Dust Control:

During the transition, when this site ceases to receive tailings, is allowed to dry, and a chemical stabilizer or a one-foot cover layer of earth is applied, we as ume there will be a small zone which is no longer wet enough to prevent dusting, and to which the cover has not yet been applied. This zone is estimated to be 10% of the area.

During the interim period 1982-1984, when the transition to use of the new tailings area has been made, and the stabilizer or one-foot earth cover is in place, we estimate only 2% of the area will be exposed to dusting. When, about 1985, the final thick cover is in place, we assume 0.5% as a maximum, very conservative estimate of the area exposed to dusting.

Summarizing:

Timestep	1	1981 (Tra	ansition	=	10% e	xposed
Timestep	2	1982-1984	(Interim)	=	2% ex	posed
Timestep	3	1985-1991		=	0.5%	exposed
Timestep	4	1992-1994		=	0.5%	exposed

Following the practice of NRC in the August 1980 Radiological Assert, we retain the full area of .095 km² for calculation of the source term table, and apply the above control values by means of particulate source strength multipliers. These are:

For	Timestep	1	.100
	Timestep	2	.020
	Timestep	3	.005
	Tinestep	4	.005

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These particulate source strength multipliers take into account only the reductions due to areas no longer exposed. In addition, it is appropriate to state the degree of reduction due to wetting. Conservatively, this is taken as 50%, since snow cover and saturation due to precipitation are expected 6 months of the year. The overall control values (particulate source strength multipliers) are then

Cimestep	1	.050
Cimestep	2	.010
imestep	3	.0025
Timestep	4	.0025

The fraction of U to tailings has decreased, compared to the Abandoned Tailings areas, because of increased U recovery. Based on 1980 production data, it is:

Fraction U to tailings = 6.0%Thus U-238 activity = $62.1 \times 6.0/9.2 = 40.5 \frac{\text{pCi}}{2}$

Other parameters remain unchanged:

Fraction Th to tailings = 95.0%
Fraction Ra and Pb to tailings = 99.8%
Annual average dust loss rate (from uncontrolled
 tailings) = 420.0 g/m²-yr

Source Terms (Particulates), before dust control factors are applied:

$$U-238 = (420 \frac{g}{M^2 - y}) (40.5 \frac{pCi}{g}) (2.5) (.095 \times 10^6 m^2) (10^{-12} \frac{Ci}{pCi}) = 4.04 \times 10^{-3} \frac{Ci}{y}$$

....

$$1n-230 = (420)(541.2)(2.3)(.095 \times 10)(10) = 0.40 \times 10$$
 01/y

··
$$Ra-226 = (420)(673.6)(2.5)(.095 \times 10^{\circ})(10^{-12}) = 6.72 \times 10^{-2} Ci/y$$

Specific Radon flux = 1.0 $\frac{pCi/m^2 - sec Rn - 222}{pCi/g Ra - 226}$

Refs

2

2

· · Radon Source Term =

=
$$(1.0)(673.6)(.095 \times 10^6 m^2)(3.156 \times 10^7 \frac{\text{sec}}{\text{y}})(10^{-12} \frac{\text{Ci}}{\text{pCi}})$$

= $2.02 \times 10^3 \text{Ci/y}$

This has identical parameters to Source No. 9, except for the coordinates, which are:

x = -0.79 km y = -0.62 kmz = -0.62 km

Source No. 11 - New Tailings I

The New Tailings area exceeds the MILDOS limit of 0.1 km², and is therefore subdivided for calculation purpose into two parts, Sources No. 11 and No. 12. For No. 11:

> Area = 0.041 km^2 x = -0.91 kmy = -0.81 kmz = 6 m (1739' Average elevation of area)

10-pg.5

Efficiency of Dust Control:

0

From the start of use, in 1981, and through its operational period, the new tailings facility provides for complete cover of tailings by water, i.e., the pool will extend over the entire area. The capacity of the facility at the present milling rate is sufficient to accommodate approximately thirteen years'

4-pg.81

production of tailings, which is in excess of the indicated available ore. Therefore, no exposed tailings beach will exist during the life of the mill. Making some provision for occasional spills, we assume <u>1% of the area is exposed</u> to dusting, but since spills will be cleaned up, this should be reduced by 80%, providing:

Net overall control factor (particulate source strength multiplier) = .01 x .20 = 2×10^{-3}

This applies to Timesteps 2 and 3. (During Timestep 1, prior to use of the New Tailings Area, the multiplier is of course zero.)

Furing the Drying Period, Timestep 4, the area will be fully exposed, so that the multiplier = 1.00

Tailings are projected to have the following activities, lower than those of tailings deposited earlier, because of declining ore grade:

U-238	26.0	pCi/gm
Th-230	410.4	
Ra-226	431.2	
Pb-210	431.2	

The U-238 value above reflects a decrease in the fraction of tranium reporting to the tailings pond, based on 1980 production data, from a former value of 9.2% to the current value of 6.0%. Thus:

Fraction	U to tailings	=	6.0%
Fraction	Th to tailings	=	95.0%
Fraction	Ra and Pb to tailings	=	99.8%

Source terms (particulates) based on full area, <u>before</u> source strength multipliers for % exposed area and % dust control are applied:

 $U-238 = (420 \frac{g}{m^2-y})(26.0 \frac{pCi}{g})(2.5)(.041 \text{ km}^2)(10^{-6}) = 1.12 \times 10^{-3} \text{ ci/y}$ $Th-230 = (420 \frac{g}{m^2-y})(410.4 \frac{pCi}{g})(2.5)(.041 \text{ km}^2)(10^{-6}) = 1.77 \times 10^{-2} \text{ Ci/y}$

19 Refs

• Ra-226 =
$$(420 \frac{g}{m^2 - y})(431.2 \frac{pCi}{g})(2.5)(.041 \text{ km}^2)(10^{-6}) = 1.86 \times 10^{-2} \text{ Ci/y}$$

.. Pb-210 = 1.86 x 10⁻² Ci/y (Same as Ra-226)

Radon Source Term:

Since there will be no exposed beach, it is inappropriate to use the specific Rn-222 flux of $1.0 \frac{pCi/m^2-sec}{pCi/g} \frac{Rn-222}{Ra-226}$,

used earlier for the tailings areas not convered by water. Tanner used 2.2 x 10^{-6} as the value of D/P for mud with 85% moisture, 5-pg.G-13 which leads to the specific radon flux calculation:

$$J_{\infty} = [Ra] \le o (\lambda D/P)^{\frac{14}{2}} \times 10^{4}$$

$$= [Ra] (0.2) (1.6) [(2.1 \times 10^{-6}) (2.2 \times 10^{-6})]^{\frac{14}{2}} \times 10^{4}$$

$$= [Ra] (6.88 \times 10^{-3})$$

We will round this to 7 x $10^{-3} \frac{pCi/m^2-sec}{pCi/g} \frac{Rn-222}{Ra-226}$,

from which:

ž

•• Radon Source Term = $(7 \times 10^{-3})(431.2)(4.1 \times 10^{5} m^{2})(3.156 \times 10^{7} \frac{\text{sec}}{\text{vr}})$

$$(10^{-12} \frac{\text{Ci}}{\text{pCi}}) = 3.91 \times 10^{1} \text{ Ci/y}$$

Source No. 12 - New Tailings 2

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Discussion of the remainder of the new tailings area, designated here New Tailings 2, is identical to New Tailings 1 (Source No. 11), except that:

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Area = .069 km<sup>2</sup>
x = -1.00 km
y = -0.86 km
z = 6 m
```

Since the area differs, the particulate and radon source terms are recomputed to reflect this change, giving:

Particulate Source Terms:

U-238 = 1.88×10^{-3} Ci/y Th-230 = 2.97×10^{-2} Ci/y Ra-226 = 3.12×10^{-2} Ci/y Pb-210 = 3.12×10^{-2} Ci/y

Radon Source Term = 6.57 x 10¹ Ci/y

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- Greenwald, N. and Serrano, L., "Radionuclide Release Ratus, Dawn Mining Company, Aug. 4-8, 1980," Magma Copper Company Department of Environmental Affairs.
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III. Receptor Coordinates

A. <u>Nearest Receptors in the</u> 16 Compass Directions

The 16 compass directions are listed below in order of frequency with which the wind blows <u>toward</u> (not from) that direction. Average wind speed is also listed, and where several sectors have the same wind direction frequency, they are listed in order of decreasing wind speed.

Next, the nearest receptor in the stated direction is identified, followed by his coordinates. Where the nearest receptor in a given direction is located outside the immediate mill vicinity (described under Table B), coordinates are omitted, but the approximate distance of the receptor from the mill is given.

	Direction	Frequency	Speed mph	Nearest Recentor	x km	y km	z km	Distance km
	1					1		1
1	NE	17	11.7	R-17	.52	.69	-2.	.86
2	NNE	15	11.5	R-129				2.2
3	SW	14	5.9	R-154				2.9
4	· WSW	8	6.3	R-161				· 3.1
5	N	7	8.3	R-28	11	.68	12.	.69
6	ENE	6	10.4	R-16	.82	.48	-2.	.95
7	NNAV	6	7.8	R-27	14	.64	12.	.66
8	NW	6	7.1	R-19	70	. 56	-5.	.90
9	SSW	4	5.8	R-107				4.1
10	WNW	3	5.7	R-31	1.36	.32	-31.	1.40
11	W	3	5.5	R-165				2.50
12	E	2	7.7	R-5	.89	07	5.	.89
13	ESE	2	7.5	R-3	.64	.31	2.	.71
14	S	2	5.5	R-109				2.3
15	SSE	1	6.1	R-1	2	62	3.	.75
16	SE	1	5.9	R-2	.44	45	2.	. 62

B. All Residences in Immediate Mill Vicinity

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Sec. B

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The immediate vicinity is that shown on a Dawn Mining Company large-scale map (1" = 300'), which extends to 1.25 km north of the mill, 2.1 km to the east, 2.4 km to the south, and 2.4 km to the west of the yellowcake dryer stack.

Residence		Distance	x	7	z
No.	Direction	(km)	<u>(km)</u>	<u>(km)</u>	(m)
R-1	SSE	.75	.42	62	3.
R-2	SE	.62	.44	45	2.
R-3	ESE	.71	.64	.31	2.
R-4	ESE	.88	.86	19	5.
R-5	E	.89	.89	07	5.
R-6	E	.97	.96	.14	1.
R-7	Ξ	1.08	1.06	.18	5.
R-8	Ε	1.59	1.58	.17	9.
R-9	Е	1.70	1.66	.35	8.
R-10	ENE	1.74	1.69	.41	7.
R-11	ENE	2.14	2.00	.75	6.
R-12	E	1.33	1.31	.24	2.
R-13	ENE	1.32	1.25	.43	3.
R-14	Ε	1.18	1.16	.24	2.
R-15 ·	ENE	1.04	·.97	.38	· 1.
R-16	ENE	.95	82	.48	-2.
R-17	NE	.86	.52	.69	-2.
R-18	NNW	.85	43	.73	-2.
R-19	NW	.90	70	.56	-5.
R-20	NNW	1.05	58	.88	-2.
R-21	NNW	.98	44	.88	-2.
R-22	NNW	.99	31	.94	-3.
R-23	NNW	.82	32	.75	-4.
R-24	N	1.17	.01	1.17	-1.
R-25	N	1.17	.10	1.17	1.
R-26	NNW	1.27	34	1.22	11.
R-27	NNW	.66	14	.64	12.
R-28	N	.69	11	.68	12.
R-29	NNW	.68	14	.67	12.
R-30	NW	1.39	-1.02	.94	-6.

Residence		· Distance	1 X	v	z
	Diraction	(km)	(km)	(km)	(m)
R-31	NUCA	1.40	1.36	.32	-31.
R-32	NW	1.13	93	.65	-12.
R-33	ENE	1.74	1.56	.77	4.
R-34	222	1.47	86	1.19	11.
R-35	ENE	1.75	1.55	.81	4.
R-36	Ξ	1.73	1.70	.34	9.
C. <u>Commerce</u> Other S in Imme	ial Recepto tations of diate Mill	ors and Interest Vicinity			
C-1 55050	PCP	78	1 75	- 22	4
C-2 Store	201	1.68	1.65	.33	8.
C-2 Store	5	1 33	1 31	25	2.
C-1 Store	ENT	1 01	1.51	49	1
C-4 SLOED	LAL NTI	1 27	.00		-4
Factory)	71.04	1.2/	05	.,,,	
Air Samplg.Stat. 1	SN	1.58	-1.20	-1.03	-6.
Air Samply Stat. 2	NE	0.86	19 N	Same as R-17	
Stat.2, Direction,	NE	0.36	.22	.29	-2.
Nearer		1 26	1 02	1 00	-2
Stat.2, Direction,	NE	1.35	.02	1.09	-2.
Grazing Area 1	W	2.03	-2:03	11	-37.
		1.44	1.5	1 10	
Grazing Area 2	NNW	1.20	45	1.17	11.
Grazing Area 3	N	.80	.10	./8	-10.
Grazing Area,due S	S	2.00	1 0.0	-2.00	-30.
D. Specifi	ic Towns and	d Cities			
Spokane	SE	41.88	33.3	-25.39	140.
Springdele	NNE	16.46	4.36	15.87	145.
Deer Park	E	27.72	26.98	6.35	0.
Hunters	NW	36.07	-27.78	23.01	0.
Wellpinit	W	11.99	-11.80	-2.10	0.
Davenport	SW	36.79	-24.10	-27.80	0.
Rearden	S	26.98	-0.3	-26.98	.0.

C-1 C-2 C-3 C-4

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E. Mill and Tailings Pond Property Boundaries

-

	x ([m])	y (km)	z (m)
Mill, N Bound -	0.	.39	-24.
Mill, NE Bo e.g	.36	.41	-18.
Mill, E Boundary	.40	0.	1.
Mill, SE Boundary	.40	37	0.
Mill, S Boundary	0.	-1.04	6.
Mill, W Boundary	-1.22	0.	-6.
Tailings, N Boundary	64	25	-2.
Tailings, NW Boundary	-1.22	25	-6.
Tailings, W Boundary	-1.22	59	-8.
Tailings, S Boundary	82	-1.00	-4.
Tailings, SE Boundary	32	91	0.

	and Radia	l Increme	nt, to a	Distance	of Eight	-		
	lilometer	s from th	e Mill					
	0-1. km	1-2 km	2-3 'm	3-4 km	4-5 km	5-6 km	6-7 km	7-8 km
S	• 2	3	• 7	2	0	0	0	•
NNE	0	2	12	0	3	15	16	18
NE	5	4	7	6	6	6	14	11
ENE	3	17	2	0	0	0	0	0
Ξ	6	13	0	4	11	0	0	0
ESE	6	0	0	2	8	0	0	0
SE	5	0	0	5	14	11	5	16
SSE	3	0	0	0	13	9	0	21
S	0	0	5	0	0	0	9	29
SSW	0	0	0	0	12	0	8	11
SW	0	0	1	0	3	0	0	9
WSW	0	0	0	14	3	0	0	. 0
W	0	0	15	15	3	0	18	42
1017	0	6	13	5	0	0	0	0
NW	4	9	6	0	0	0	0	0
NNW	20	16	13	3	0	6	3	0

IV. Population Distribution by Directional Sector

				7 Annua	1 Cons	umption	1 Locally Grown	
			% of Time	of Lo	scally i	Crown:	Feed for	I of Local Feed
So.	Age	Sex	at Residence	Meat	Milk	Veg.	Fowl Fork Seef Other	Which is Pasture Grass
1.1.1			1					
E=1=1	26	2	22				Non trailings of	
-2	3	1. S. 1.	50		0	0	DOE APPLICADIE	51.14.1
-3		13	5-1 C					
R-2-1	-8	M	00					
-2	25		70					
-3	8		80	0	Q.	0	0.04×	16.48.4
	3		20					
-3			12 (
R-3-1	21		55					
-2	20		100	100	100	100	- 100 20	20
- 3			80					
-4	12	M	30 /					
R-4-1	01			50	0	0	- 50 -	
2	10		10			75		
R-3-1	11		100	0	0	12	E set s	17.0.
2-0-1	33	M						
+2	35	2	43		1.00			
	14	M	00	40	100	23	50	
	11	3	80					
-5	0	1	80 /					
8-10-1	47	M	22	20		50	nor known	
		3			0	30	NO'C KOTO'AN	
-3	11							
R-17-1	30	M	09					
	÷	5	2				- 0 -	0
		2	**	• • • •	Ŷ			
	3		12					
-2		1	2 5					
X+13+1	21		60					
	25	21	20	. 0	0	0	n.a.	n.a.
			00					
	13	5	76 5					
R-19-1	20		13					
	43		70	- 0	0	10	n.a.	n.a.
- 3	17		75	8 12 P. C.				
	1/		13	0	0		5.3.	n.a.
K+21-1	-0		60 1		~			
X-11-1	23		60	0	0	0	0.4.	n
	17	5	00	~	· · ·	, i i i i i i i i i i i i i i i i i i i		
			50					
8-13-1	26		50	1.1.1				
			20		0	50	50	50
	é		90					
	5		50					
0. 27-1	20		90	1.1.1.1.1.1.1				
a-1/-1		E	90					
	5		90	0	0	10	n.a.	ü.a.
	1		90					
2-18-1	50	ų	90				10	
1-10-1	44		75	10	0	25	10	
8-29-1	48	5	50					
	60	w.	90	- 0	0	25	n.a.	n.a.
-1	12	М	80					
			and the second					

V. Characteristics of All Individuals Residing Nithin One Kilometer of the Mill

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C.

Notes: R-2 is the nearest residence, at 0.52 km SE. R-17 is the nearest residence in the prevailing win direction, at 0.86 km NE.