CONTROL AND MANAGEMENT OF OUTBURST IN AUSTRALIAN UNDERGROUND COAL MINES

Dennis J. Black

ABSTRACT: Outbursts represent a major safety hazard to mine personnel working near the coal face in areas of increased outburst risk. There have been over 878 outburst events recorded in twenty-two Australian underground coal mines and most outbursts have been associated with abnormal geological conditions. Details of Australian outburst incidents and mining experience in conditions where gas content was above current threshold levels is presented and discussed. Mining experience suggests that for gas content below 9.0 m³/t, mining in CO₂ rich seam gas conditions does not pose a greater risk of outburst than mining in CH₄ rich seam gas conditions. Mining experience also suggests that where no abnormal geological structures are present, mining in areas with gas content greater than the current accepted threshold levels can be undertaken with no discernible increase in outburst risk. The current approach to determining gas content threshold limits in Australian mines has been effective in preventing injury from outburst however operational experience suggests the current method is overly conservative and in some cases the threshold limits are low to the point that they provide no significant reduction in outburst risk. Other factors that affect outburst risk, such as gas pressure, coal toughness and stress and geological structures are presently not incorporated into outburst threshold limits adopted in Australian mines. These factors and the development of an Outburst Risk Index applicable to Australian underground coal mining conditions is the subject of ongoing research.

INTRODUCTION

Outburst has been defined as the sudden release of gas and material from the working place that can vary in magnitude and intensity (NSWDMR, 1995). The occurrence of an outburst is preceded by failure of the coal and during an outburst, the failed material is ejected with energy and with gas. The difference between a rockburst and an outburst is the gas that is emitted. The gas contributes in a major way to the expulsion of the coal and is generally thought to be the main contributor to total energy release (Gray and Wood, 2013).

Outbursts of coal and gas have been experienced in underground coal mines in many countries, including Australia (Lama and Bodziony, 1996). Outburst vary in size and intensity, from small bumps equivalent to rib failure without discernible gas release to violent ejections of thousands of tonnes of coal and rock releasing tens of thousands of cubic metres of seam gas. The sudden release of a large volume of seam gas into a mining place following an outburst can create significant potential risks to personnel safety which include: (a) danger of asphyxiation due to oxygen deficiency, (b) poisoning by noxious gases, (c) explosion by inadvertent ignition of the resultant explosive mixtures, (d) injury resulting from the violent ejection of coal and gas, and (e) exposure to dense coal dust.

Gas, geology and stress have been identified as dominant parameters that combine to create outburst conditions and provide the energy required to expel coal from the working face (Black et al., 2009). Outbursts are usually, but not invariably associated with faults, dykes, seam variations and dislocations. In some mines, such as Leichhardt colliery in Queensland some outbursts occurred in areas with no abnormal geological structure or with structures which elsewhere in the same mine had been quite benign (Hanes, 1995).

The general nature of the outburst risk is such that it may be continuously variable, not only between mines but also within an individual colliery’s workings. A single, unchanged approach to the management of the risk is, therefore, inappropriate. A degree of discipline is also warranted to identify, and effectively act upon, changes in the mine operating environment, which may be subtle, which may be associated with the potential for outbursts (NSWDMR, 1995).

AUSTRALIAN OUTBURST HISTORY

The first recorded outburst in Australia occurred at Metropolitan colliery on 30 September 1895 and in the 122 year period to 2017, over 878 outburst events have been recorded in twenty-two Australian underground coal mines. Mining operations in the Bulli seam, located in the southern Sydney basin, have the longest history of outburst in Australia, with fourteen collieries recording over 641 outburst incidents. Ellalong colliery, working the Greta seam in the northern Sydney basin, recorded five relatively small-scale outburst events, and is the only non-Bulli seam mine in New South Wales to record an outburst event. Seven collieries operating in the Bowen basin, Queensland, have recorded over 232 outburst incidents, with the largest number of events recorded during development mining in the Gemini seam at Leichhardt colliery. Table 1 provides a summary list of recorded outburst data compiled from extensive review of published reports and Mines Department records.

The lives of twenty-one men and five horses have been lost in eight outburst incidents in Australian underground coal mines. The largest outburst event recorded in Australia, which claimed the lives of seven men and three horses, occurred at the Collinsville State mine in Queensland on 13 October 1954. In three separate outburst incidents, seven men and two

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horses were killed at Metropolitan colliery. Table 2 provides a summary list of fatal outburst incidents that have occurred in Australian underground coal mines.

Table 1 – Summary of Recorded Outburst Incidents in Australia

<table>
<thead>
<tr>
<th>Mine</th>
<th>State</th>
<th>Seam</th>
<th>Recorded Outburst Events</th>
<th>Recorded Outburst Date Range</th>
<th>Max. Coal Outburst (tonnes)</th>
<th>Max. Gas Outburst (m³)</th>
<th>Gas Type</th>
<th>Associated Geological Structure</th>
<th>Gas Drainage Status</th>
</tr>
</thead>
<tbody>
<tr>
<td>Appin</td>
<td>NSW</td>
<td>Bull</td>
<td>67</td>
<td>May 1966 - Feb 2017</td>
<td>150</td>
<td>5,100</td>
<td>Mostly CH₄</td>
<td>Strike-Slip, Thrust, Normal Faulting, Dykes</td>
<td>No or Inadequate Drainage</td>
</tr>
<tr>
<td>Blayney</td>
<td>NSW</td>
<td>Bull</td>
<td>2</td>
<td>1992</td>
<td>&lt;40</td>
<td>Unknown</td>
<td>Mixed</td>
<td>Fault</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Coal Cliff (closed)</td>
<td>NSW</td>
<td>Bull</td>
<td>2</td>
<td>1961</td>
<td>20</td>
<td>Unknown</td>
<td>CH₄</td>
<td>Dyke</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Cominal (South Bull)</td>
<td>NSW</td>
<td>Bull</td>
<td>9</td>
<td>Oct 1967 - Sep 1983</td>
<td>40</td>
<td>Unknown</td>
<td>CH₄</td>
<td>Shear Fault with Mylonite</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Darke Forest (closed)</td>
<td>NSW</td>
<td>Bull</td>
<td>2</td>
<td>1989</td>
<td>10</td>
<td>Unknown</td>
<td>Mixed</td>
<td>Dyke</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Ellenbrook (Greta)</td>
<td>NSW</td>
<td>Bull</td>
<td>5</td>
<td>1984</td>
<td>&lt;30</td>
<td>Unknown</td>
<td>Mostly CO₂</td>
<td>Bedding plane shear. High Stress</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Kemnay (closed)</td>
<td>NSW</td>
<td>Bull</td>
<td>2</td>
<td>May 1980 - May 1981</td>
<td>100</td>
<td>Unknown</td>
<td>CH₄</td>
<td>Thrust Fault</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Metropolitan</td>
<td>NSW</td>
<td>Bull</td>
<td>169+</td>
<td>Sep 1899 - Jan 2017</td>
<td>250</td>
<td>11,500</td>
<td>Mixed</td>
<td>Thrust Fault &amp; Dykes, Shifting Induced</td>
<td>No or Inadequate Drainage</td>
</tr>
<tr>
<td>North Bull (closed)</td>
<td>NSW</td>
<td>Bull</td>
<td>1</td>
<td>1911</td>
<td>1</td>
<td>Unknown</td>
<td>CH₄</td>
<td>Faults</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Oakdale (closed)</td>
<td>NSW</td>
<td>Bull</td>
<td>1</td>
<td>1916</td>
<td>&lt;10</td>
<td>Unknown</td>
<td>Mixed</td>
<td>Fault</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Bull (closed)</td>
<td>NSW</td>
<td>Bull</td>
<td>3</td>
<td>1972</td>
<td>30</td>
<td>Unknown</td>
<td>CH₄</td>
<td>Faults</td>
<td>No Drainage</td>
</tr>
<tr>
<td>South Bull (Russell Vale)</td>
<td>NSW</td>
<td>Bull</td>
<td>7</td>
<td>Feb 1985 - Apr 1996</td>
<td>120</td>
<td>6,000</td>
<td>Mostly CH₄</td>
<td>Strike-Slip &amp; Thrust Faulting, Dykes</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Tahmoor</td>
<td>NSW</td>
<td>Bull</td>
<td>99</td>
<td>May 1961 - Mar 1992</td>
<td>400</td>
<td>4,500</td>
<td>Mixed</td>
<td>Faults &amp; Dykes</td>
<td>No or Inadequate Drainage</td>
</tr>
<tr>
<td>Tower (Appin)</td>
<td>NSW</td>
<td>Bull</td>
<td>21</td>
<td>Jul 1981 - Dec 2000</td>
<td>80</td>
<td>Unknown</td>
<td>CH₄</td>
<td>Strike-Slip Fault &amp; Dyke</td>
<td>No or Inadequate Drainage</td>
</tr>
<tr>
<td>West Cliff (Appin)</td>
<td>NSW</td>
<td>Bull</td>
<td>260</td>
<td>Dec 1978 - Apr 1999</td>
<td>350</td>
<td>10,000</td>
<td>Mostly CH₄</td>
<td>Thrust Faults &amp; Dykes</td>
<td>No or Inadequate Drainage</td>
</tr>
<tr>
<td>Centrale (closed)</td>
<td>QLD</td>
<td>Goomurra Creek</td>
<td>1</td>
<td>20 Jul 2001</td>
<td>80</td>
<td>1,500</td>
<td>CH₄</td>
<td>Tiptop Strike-Slip Fault</td>
<td>Inadequate Drainage</td>
</tr>
<tr>
<td>Collinsville State</td>
<td>QLD</td>
<td>Bowen</td>
<td>13</td>
<td>Mar 1960 - Mar 1961</td>
<td>500</td>
<td>14,000</td>
<td>CO₂</td>
<td>Strike-Slip, Thrust &amp; Normal Faulting</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Collinsville No 3</td>
<td>QLD</td>
<td>Bowen</td>
<td>2</td>
<td>Mar 1972 - Apr 1972</td>
<td>10</td>
<td>Unknown</td>
<td>CO₂</td>
<td>Strike-Slip &amp; Thrust Faults</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Collinsville No 2</td>
<td>QLD</td>
<td>Bowen</td>
<td>7</td>
<td>Sep 1978 - Nov 1978</td>
<td>35</td>
<td>380</td>
<td>CO₂</td>
<td>Strike-Slip &amp; Thrust Faults</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Leichhardt (closed)</td>
<td>QLD</td>
<td>Gemini</td>
<td>200+</td>
<td>1975 to 1982</td>
<td>350</td>
<td>13,500</td>
<td>CH₄</td>
<td>Thrust Faults High Stress</td>
<td>No Drainage</td>
</tr>
<tr>
<td>Mount No 4 (closed)</td>
<td>QLD</td>
<td>C Seam</td>
<td>3</td>
<td>1980 to 1983</td>
<td>Unknown</td>
<td>Unknown</td>
<td>CH₄</td>
<td>Major Joint Planes</td>
<td>No Drainage</td>
</tr>
<tr>
<td>North Goonyella</td>
<td>QLD</td>
<td>Goonyella Middle</td>
<td>6</td>
<td>Oct 2001 - Jan 2015</td>
<td>150</td>
<td>5,000</td>
<td>CH₄</td>
<td>Shear Zone, Mylonite Infill</td>
<td>Inadequate Drainage</td>
</tr>
</tbody>
</table>

Table 2 – Summary of Fatal Outburst Incidents in Australia

<table>
<thead>
<tr>
<th>Mine</th>
<th>State</th>
<th>Seam</th>
<th>Date</th>
<th>Loss of Life</th>
<th>Outburst Size (tonnes)</th>
<th>Gas Volume</th>
<th>Gas Type</th>
</tr>
</thead>
<tbody>
<tr>
<td>Metropolitan</td>
<td>NSW</td>
<td>Bull</td>
<td>10 Jun 1896</td>
<td>3 men, 1 horse</td>
<td>Unknown</td>
<td>220</td>
<td>Mixed CH₄, CO₂</td>
</tr>
<tr>
<td>Metropolitan</td>
<td>NSW</td>
<td>Bull</td>
<td>27 Jun 1925</td>
<td>2 men, 1 horse</td>
<td>220</td>
<td>Mixed CH₄, CO₂</td>
<td></td>
</tr>
<tr>
<td>Collinsville State</td>
<td>QLD</td>
<td>Bowen</td>
<td>13 Oct 1954</td>
<td>7 men, 3 horses</td>
<td>500</td>
<td>14,000 m³ CH₄</td>
<td></td>
</tr>
<tr>
<td>Metropolitan</td>
<td>NSW</td>
<td>Bull</td>
<td>02 Dec 1954</td>
<td>2 men</td>
<td>90</td>
<td>CO₂</td>
<td></td>
</tr>
<tr>
<td>Leichhardt</td>
<td>QLD</td>
<td>Gemini</td>
<td>01 Dec 1976</td>
<td>2 men</td>
<td>350</td>
<td>13,500 m³ CH₄ + CO₂</td>
<td></td>
</tr>
<tr>
<td>Tahmoor</td>
<td>NSW</td>
<td>Bull</td>
<td>24 Jun 1985</td>
<td>1 man</td>
<td>400</td>
<td>4,500 m³ CH₄ + CO₂</td>
<td></td>
</tr>
<tr>
<td>South Bull</td>
<td>NSW</td>
<td>Bull</td>
<td>24 Jul 1991</td>
<td>3 men</td>
<td>300</td>
<td>6,000 m³ CH₂ + CO₂</td>
<td></td>
</tr>
<tr>
<td>West Cliff</td>
<td>NSW</td>
<td>Bull</td>
<td>25 Jan 1994</td>
<td>1 man</td>
<td>350</td>
<td>CO₂</td>
<td></td>
</tr>
</tbody>
</table>

Virtually all outburst events in the Bulli seam have been associated with geological structures and have occurred in areas where no substantial gas drainage has been undertaken (Lama, 1995). The highest number of outburst events have occurred at West Cliff, Metropolitan, Appin and Tahmoor collieries which all work the Bulli seam and Leichhardt colliery which worked the Gemini seam. Brief summaries of the outburst histories at these five collieries are provided.
West Cliff Colliery

West Cliff colliery commenced coal production in October 1976 and longwall mining was introduced in 1982 (Eade, 2002). The first recorded outburst occurred at West Cliff on 20 December 1976 and two hundred and fifty four (254) outburst events were recorded at West Cliff prior to the mine ceasing production in 2015-2016. The size of outbursts varied from four to over 300 tonnes, with the majority being related to zones of strike-slip faulting. Outbursts at West Cliff typically occurred in association with pulverised coal in shear zones running through the coal. The shear zones were also regions of high gas pressure, which when intersected, resulted in displacement of pulverised coal into the excavations (Marshall et al., 1980). Walsh (1999) reported that of the approximately 250 outbursts recorded at West Cliff, 70% occurred on strike-slip faults, 4% on dykes and faults, 1% on thrusts, 3% on normal faults; and 19% on bedding slips.

The largest outburst at West Cliff, reported to have displaced 320 tonnes of coal, occurred at the northwest end of a normal fault where the gas drainage holes had not penetrated. There was a major joint zone 3-4 m wide in the roof associated with this outburst site and a mylonite band some 30-mm thick. The gas composition had been predominantly methane. In the north-eastern part of the mine, outburst events had occurred in areas with high gas content (>16 m³/t) and high concentrations of carbon dioxide (>95% CO2) (Harvey, 1994).

Marshall et al. (1980) reported the time taken for an outburst to manifest at West Cliff varied from several seconds to almost a minute. In the opinion of Harvey (1994), mining operations at West Cliff had been possible through the use of gas drainage and specified outburst mining procedures.

On 03 April 1998, West Cliff became the first mine in Australia to record an outburst on a retreating longwall face. Two outbursts of approximately 17 tonnes were identified on the longwall face (LW23) at Chocks #45 and #54 during the flit run of the cutting cycle (Walsh, 1999). The outbursts, comprising some fines but mostly blocky coal (Piper, 1998) were identified as cones extending into the face (>1.0 metre) at the top of the 2.5 metre Bulli seam. Two hours after the outbursts, gas continued to be liberated from the cavities and could be heard as an audible hiss and a visible haze (Walsh, 1999). No abnormal structures, such as strike-slip or thrust faults were noted at the outburst sites, however a bedding plane slip was present in the seam, approximately 100 mm below the roof. There was also no prominent cleat noted at the outburst sites (Walsh 1999).

Metropolitan Colliery

Mining operations at Metropolitan colliery commenced in 1888 with the Bulli seam being mined by hand, with some single round shotfiring (Ward, 1980). Metropolitan was the first colliery in Australia to record an outburst, which occurred on 30 September 1895. The mine has since recorded over 169 outburst events.

The highest incidence of outburst occurred during the mining of the 2 South District between 1961 and 1968, where over 100 outburst events were reported to have been induced by shotfiring. The largest reported outburst ejected 250 tonnes of coal and an unknown quantity of predominantly CO2 (Lama, 1995). On 03 April 1998, West Cliff became the second mine in Australia to record an outburst on a retreating longwall face. Two outbursts of approximately 17 tonnes were identified on the longwall face (LW23) at Chocks #45 and #54 during the flit run of the cutting cycle (Walsh, 1999). The outbursts, comprising some fines but mostly blocky coal (Piper, 1998) were identified as cones extending into the face (>1.0 metre) at the top of the 2.5 metre Bulli seam. Two hours after the outbursts, gas continued to be liberated from the cavities and could be heard as an audible hiss and a visible haze (Walsh, 1999). No abnormal structures, such as strike-slip or thrust faults were noted at the outburst sites, however a bedding plane slip was present in the seam, approximately 100 mm below the roof. There was also no prominent cleat noted at the outburst sites (Walsh 1999).

Appin Colliery

Appin commenced operation in 1962 and longwall mining was introduced in 1969 (Eade, 2002). The first recorded outburst occurred in May 1966, ejecting 50 tonnes of coal and an unknown quantity of CH4. The outburst occurred in a zone of joints that were evident in the immediate roof (Harvey, 2002). A total of 67 outburst have been recorded at Appin. Twenty outbursts events were recorded in the 27 years to 1994, the largest occurred in July 1969 when development mining intersected a strike-slip fault with mylonite displacing 100 tonnes of coal and an unknown volume of CH4 (Lama, 1991).

In the years following 1994, 47 seven outbursts have been recorded at Appin, the largest reported in May 2009 while operating a remote controlled continuous miner to develop through a known thrust fault zone that had been difficult to drill and drain gas below the outburst threshold. The outburst displaced 150 tonnes of coal and released 1,140 m³ of predominantly CH4.

The largest recorded outburst to have been induced while shotfiring through a dyke associated with strike-slip faulting occurred in January 2013 and released an estimated 5,100 m³ of predominantly methane. The mass of coal displaced by the outburst, in addition to the planned shotfiring excavation, was not reported.

Outburst events at Appin typically occur in areas where prominent geological features have been intersected. Such features include faults, particularly strike-slip and thrust faults, adjacent to dykes and associated cindered coal.

Harvey (2002) reported five small outbursts, four being less than eight tonnes and one of up to 20 tonnes, had occurred in areas where 'no prominent geological structure' had been identified.
Gas content at Appin has been measured at levels exceeding 16 m³/t and an extensive gas drainage system is used to prevent or minimise the risk of outbursts and manage gas liberated during mining. Composition of the gas is predominantly CH₄, however high CO₂ has been recorded adjacent to faults and dykes (Harvey, 2002).

**Tahmoor Colliery**

Mine development commenced at Tahmoor in 1978 and longwall mining was introduced in 1986 (Newman, 2005). Ninety-nine (99) outbursts have been recorded at Tahmoor colliery in the years following the first recorded outburst in 1981. The mine identified the developing outburst problem, with events progressing in significance from slumps and pressure bumps to large outbursts occurring on geological structures, particularly dykes and strike-slip faults (Stone, 1991, Newman, 2005 and Wynne and Case, 1995). Wynne and Case (1995) reported that all outburst events at Tahmoor had occurred during the cutting phase of the development mining cycle. Table 3 lists the structural association of outbursts recorded at Tahmoor (Stone, 1991).

<table>
<thead>
<tr>
<th>Geological Structure</th>
<th>No. of Outbursts</th>
<th>Violent Outburst</th>
<th>Size of Outburst</th>
</tr>
</thead>
<tbody>
<tr>
<td>Across Dyke</td>
<td>3</td>
<td>3</td>
<td>5 - 400 t</td>
</tr>
<tr>
<td>Strike-Slip Fault / Dyke</td>
<td>27</td>
<td>17</td>
<td>6 - 120 t</td>
</tr>
<tr>
<td>Strike-Slip Fault</td>
<td>49</td>
<td>16</td>
<td>5 - 60 t</td>
</tr>
<tr>
<td>Reverse Fault</td>
<td>4</td>
<td>1</td>
<td>5 - 40 t</td>
</tr>
</tbody>
</table>

Following the fatal outburst incident that occurred in the 204 Panel on 24 June 1985, Tahmoor worked to modify the Joy 12 continuous miner to provide increased protection for the miner driver in outburst conditions. A completely enclosed cabin was built which was protected by one-inch thick bulletproof glass (Figure 1). Inside the cabin, the operator, who wore an air mask, communicated through radio control with the shuttle car driver and the crew at a fresh air base. The enclosed cabin was ventilated with two sources of fresh air supply, along with two additional sources of emergency air supply. Air pressure in the cabin was maintained to stop potential gas seepage into the miner's cabin. The outburst miner was replaced in 1992 with the introduction of the remote controlled Alpine Bolter Miner (ABM20).

Wynne (2002) listed critical events and stages in the evolution of outburst management at Tahmoor, which include:

- 1981 – first recorded outburst;
- 1985 – continuous miner driver killed by outburst whilst cutting dyke;
- 1985 – encapsulated continuous miner introduced for cutting outburst structures;
- 1982 to 1992 – averaging 10 outbursts per year crossing structures;
- 1992 – introduced ABM20 continuous miner capable of being remotely operated;
- 1992 – commenced pre-drainage of coal around structures;
- 1992 – draft Outburst Management Plan;
- 1992 – remote mining through fault, last recorded outburst;
- 1992 to 1997 – ongoing refinement of drilling techniques;
- 1994 – Outburst Management Plan formalised;
- 1999 to 2001 – shotfiring (grunching) through “tight” coal zones.
Wynne (2000) reported there had not been an outburst event at Tahmoor Colliery since 1992, due to effective pre-drainage and the Outburst Management Plan.

In 2003, Tahmoor introduced increased gas content threshold levels for outburst control, which included among other management controls, (a) increased drilling for pre-mining gas content reduction and compliance core sampling, and (b) limited rate mining.

**Leichhardt Colliery**

Leichhardt colliery was the most outburst prone underground coal mine in Australia, with more than two hundred outbursts reported between 1973 and 1978. Leichhardt colliery commenced operations in the Gemini seam in 1973 and, following the fatal outburst in December 1978, the mine was placed on ‘care and maintenance’ and operated as an experimental mine for four years before closing in December 1982.

The average thickness of the Gemini seam was 6.0 m and the mine operated at a depth of 350 to 410 m. Within the mine, shallow dipping reverse faults of minor displacement and associated slickensides were common. Seam gas was predominantly CH4 (95% CH4 / 5% CO2). Characteristics of the Gemini seam included high desorption rate, high gas pressures, low permeability and the gas content was around 15 m³/t (Truong et al., 1983 and Hanes, 2001).

Mining was done by continuous miners, shotfiring and an Alpine Roadheader. Drilling large diameter boreholes in advance of mining was used for a period to reduce stress and gas emissions and was considered to improve the mining conditions. The practice was later abandoned due to a belief that large diameter boreholes were no more effective than smaller diameter holes and operational problems were associated with drilling large diameter boreholes. Other preventive techniques such as shotfiring, and delayed action shotfiring were also practised which generally reduced the frequency of outburst occurrences (Truong et al., 1983).

Mining induced cleavage was typical in the coal at Leichhardt. It curved around the face forming large sheets of coal which easily spalled or at times, burst. Outbursts had not occurred in the western workings of the mine due to changes in the gas and/or structural regime and the coal which was free from bursts lacked the mining induced cleavage (Moore and Hanes, 1980 and Hanes, 2006). In the eastern part of the mine, outbursts occurred frequently (daily) from the rib (Moore and Hanes, 1980). The outbursts were typically small and occurred as the violent buckling of a few tonnes of the cleated coal into the opening. At times, the miner driver could “turn on” an outburst for visitors (Moore and Hanes, 1980).

Typically, outburst prone coal was intensely cleaved around the mine opening. Outbursts were partly controlled by stress and by the cleats. Drives near parallel to the maximum principal stress were free from outbursts and other mining strain, whereas drives nearly perpendicular to the principal stress were highly strained and outburst prone. The rib which first intersected the cleat was the focus of most bursts, which projected perpendicular to the cleat (Hanes, 2001). Outburst cavities in the Gemini Seam were typically oriented such that their axes were perpendicular to the face cleat direction and the bursts occurred from the ribs or face generally on the side which first encountered the cleat (Moore and Hanes, 1980). Some outbursts occurred with their axes perpendicular to prominent induced cleavage and many bursts occurred from coal roof with their axes perpendicular to the bedding planes (Hanes, 1979).

The orientation of the fatal outburst that occurred on 01 December 1978 was apparently controlled by cleat orientation. The axis of the burst cavity was approximately perpendicular to the dominant cleat direction over most of its length. Also, the axis of the burst cavity was nearly parallel to the mean strike of slickenside planes in the burst cavity walls (Hanes, 1979).

Tight ribs preceded most outbursts. Pick marks were obvious for the full height of the seam. The face at the fatal 1978 outburst had very hard coal ribs which “rang” when hit with a hammer (Hanes, 2006).

Measured gas pressure gradients showed that gas pressure was the controller of outbursts. When the pressure gradient in the face was high, outbursts occurred (Hanes, 2001). Gas flow measurements showed that on drilling, gas did not flow from drainage holes until the holes had stood for 2 to 3 months (Hanes, 2001). Figure 2 shows the results of pressure measurements in the coal ahead of the working face in outburst and benign conditions reported by Hanes (1995).

![Figure 2. Gas pressure and content gradients recorded in the Gemini seam, Leichhardt colliery (Hanes, 1995)](image-url)
INTRODUCTION OF OUTBURST THRESHOLD LIMITS IN BULLI SEAM MINES

The earliest attempts to develop safe threshold values for mining the Bulli seam were based upon measurement of gas emission rate from freshly cut coal (Lama, 1995). A version of the French, Belgian and Polish gas emission meter was introduced to Australian mines by Hargraves, where a 4.0-gram coal sample of -14 to +25 mesh fraction was collected and gas emission measured over a 2 to 6-minute period (Lama, 1995). Indices were developed which showed that if the gas emission was greater than 1.5 cc/g for CH₄ and 1.2 cc/g for CO₂, then the face was liable to outburst (Lama, 1995). The method required indices to be developed for each site, to suit local conditions, yet based on work in French mines the index value CO₂ areas was dropped to 1.0 cc/g (Lama, 1995). There were several problems with this method which affected the accuracy and repeatability of the measurement, which included moisture, variability of coal ply and depth of drill hole from which the sample was sourced. These issues, combined with the introduction of high performance roadway development systems to support the introduction of longwall mining rendered the method, which required frequent gas emission measurement of coal samples collected from 2-3 metres holes drilled ahead of the advancing working faces, unsuitable as it adversely affected productivity and the results were considered unreliable (Lama, 1995).

Early attempts at drilling larger diameter boreholes, up to 300 mm diameter, ahead of the working face as a means of reducing stress also aided in draining gas and reducing gas emissions thereby reducing outburst risk (Lama, 1995).

Lama (1995) reported gas drainage investigations at Tahmoor showed that when an area had been drained to gas levels between 9.0 - 10.7 m³/t, with CO₂ percentage 40 - 45%, there were no violent outburst events, even when structures such as dykes were present in the area. Lama also reported an outburst with the emission of almost 3000 m³ of gas occurred in an area where gas content was between 11 - 12 m³/t, with gas pressure of 1700 kPa. When gas levels were dropped to 6.0 m³/t, no outbursts occurred. Work at Metropolitan colliery found the Bulli seam had been mined without outburst in areas where the gas emission value was below 0.6, with desorbable gas content of 4.0 m³/t in 90% CO₂ (Lama, 1995).

It should be noted that gas content values reported by Lama were desorbable gas content (Q₁ + Q₂) as the gas content test method did not routinely measure the Q₃ residual gas content component. The test method used by Lama to determine the desorbed gas content involved measuring gas desorbed from coal samples over a maximum 48 hour period, or the time when at least one negative gas emission value was observed as a result of minor pressure and temperature changes causing resorption of gas from the surroundings into the coal sample (Lama, 1995).

Lama (1995) provided a brief description of work to measure the residual gas content of a selection of Bulli coal samples which, with some corrections, average Q₃ values of 2.01 m³/t for CH₄ and 2.4 m³/t for CO₂. The approach taken by Lama was to add the average residual gas content values (Q₃) to the measured desorbed gas content (Q₁ + Q₂) values to report total gas content (Q₁ + Q₂ + Q₃). Results of gas emissions from slow desorption testing of coal samples by Black (2011) raises concerns for the accuracy of Lama’s approach to determining total gas content, as indicated by gas emission measurements of two coal samples, shown in Figure 3, which show gas desorption occurs for a substantially longer period than 48 hours with Q₃ greater than 2.0 m³/t being measured from coal core after slow desorption testing for a period of 600 days.

Based on the results of gas content measurements and a review of gas content threshold values used by other countries, such as Poland, Russia, Germany, Bulgaria and China, Lama proposed gas content threshold values based on desorbable gas content (1991) and total gas content (1995), having added the average Q₃ test results to the desorbable gas content threshold values. The gas content threshold values proposed by Lama, for both desorbable and total gas content, are presented in Table 4.
In 1992 the NSW Chief Inspector of Coal Mines (CICM), concerned about the increasing number of outburst incidents, indicated by data presented in Figure 4, reported in Annual Reports of the NSW Department of Mineral Resources, and the apparent lack of effective management of outburst risk, formed specialist work groups to identify the regional characteristics of outbursts and develop the most appropriate means of protecting mine workers (Harvey, 1994). The working group identified the need for management plans, and all mines operating in the Bulli seam were requested to prepare outburst management plans to specify how they would manage outburst risk (NSWDMR, 1992, Harvey, 1994). The objective of the NSW Department of Mineral Resources (DMR) was that all Southern Coalfield mines would be operating under auditable outburst management plans by 30 June 1993. However, the plans submitted to the DMR were generally regarded as inadequate and were returned to the mines for further development. During 1992 and 1993, the CICM was considering the introduction of gas content threshold limit values as a means of reducing outburst risk. During this process, gas content TLVs as low as 6.0 m\textsuperscript{3}/t in 100% CH\textsubscript{4} conditions and 3.0 m\textsuperscript{3}/t in 100% CO\textsubscript{2} were considered.

Table 4. Lama’s recommended gas threshold values for safe mining of Bulli seam (total gas content) (Lama, 1991 and 1995)

<table>
<thead>
<tr>
<th>Proposed Outburst Threshold Limits [ seam Status]</th>
<th>Desorbable Gas Content (m\textsuperscript{3}/t) (Lama, 1991)</th>
<th>Total Gas Content (m\textsuperscript{3}/t) (Lama, 1995)</th>
</tr>
</thead>
<tbody>
<tr>
<td>Level 1 TLV: Presence of Structure</td>
<td>8.0</td>
<td>8.0 + 2.0 = 10.0</td>
</tr>
<tr>
<td>Level 2 TLV: No Structures Present</td>
<td>10.0</td>
<td>10.0 + 2.0 = 12.0</td>
</tr>
</tbody>
</table>

In 1992 the NSW Chief Inspector of Coal Mines (CICM), concerned about the increasing number of outburst incidents, indicated by data presented in Figure 4, reported in Annual Reports of the NSW Department of Mineral Resources, and the apparent lack of effective management of outburst risk, formed specialist work groups to identify the regional characteristics of outbursts and develop the most appropriate means of protecting mine workers (Harvey, 1994). The working group identified the need for management plans, and all mines operating in the Bulli seam were requested to prepare outburst management plans to specify how they would manage outburst risk (NSWDMR, 1992, Harvey, 1994). The objective of the NSW Department of Mineral Resources (DMR) was that all Southern Coalfield mines would be operating under auditable outburst management plans by 30 June 1993. However, the plans submitted to the DMR were generally regarded as inadequate and were returned to the mines for further development. During 1992 and 1993, the CICM was considering the introduction of gas content threshold limit values as a means of reducing outburst risk. During this process, gas content TLVs as low as 6.0 m\textsuperscript{3}/t in 100% CH\textsubscript{4} conditions and 3.0 m\textsuperscript{3}/t in 100% CO\textsubscript{2} were considered.

Following the fatal outburst at West Cliff colliery on 25th January 1994, the DMR issued a notice to all mines operating in the Bulli seam pursuant to Section 63 of the Coal Mines Regulation Act 1982, detailing actions to be implemented to prevent further outburst related fatalities. Arguably the most significant of these actions was the stipulation of limits on seam gas content prior to mining, known as outburst threshold limit values (TLV). Figure 5 shows the Bulli seam TLV prescribed in the Section 63 notification (Clarke, 1994). The TLV varied linearly based on gas composition, with the presence of CO\textsubscript{2} seam gas considered a significantly higher outburst risk than CH\textsubscript{4} seam gas. The Level 1 TLV for ‘normal’ mining was 9.0 m\textsuperscript{3}/t in 100% CH\textsubscript{4} conditions and 5.0 m\textsuperscript{3}/t in 100% CO\textsubscript{2} conditions. If gas content was not reduced below the Level 1 TLV, mining was only permissible under outburst mining procedures. The Level 2 TLV for ‘outburst’ mining was 12.0 m\textsuperscript{3}/t in 100% CH\textsubscript{4} conditions and 10.0 m\textsuperscript{3}/t in 100% CO\textsubscript{2} conditions. If gas content was not reduced below the Level 2 TLV, mining was only permissible using remotely operated equipment, with all personnel remaining clear of the outburst risk zone.

The introduction of TLV resulted in a significant increase in the intensity of drilling and gas drainage to identify geological structures and reduce gas content below threshold limits. Mine operators developed comprehensive outburst management plans which included standard drilling patterns and routine management controls to deal with the issue of gas content reduction. However, these TLVs preceded the introduction of intensive inseam gas drainage drilling and the capability of directional drilling technology to aid in locating geological structures and other outburst risk zones.
Lama (1995) presented gas content data collected over a three (3) year period from sites where headings had been mined through structures with and without outbursts. With reference to the gas content and outburst data, presented in Figure 6, Lama proposed the two threshold limits lines; solid TLV line for areas with structures present and dotted TLV line for areas without structures. Lama stated the proposed values were safe and the safety factor of 19% (1.1 m³/t) was greater than the error associated with gas content measurement. Points lying between the two TLV lines show small outbursts occurring on structures. Details of the size of the outburst events, stated by Lama to be "too small to cause any major damage or endanger life", are also presented in Figure 6. The gas content TLVs proposed by Lama for safe mining of the Bulli seam, based on the presence or absence of geological structures, and based on the work presented in Figure 6, have been summarised in Figure 7. The proposed Level 1 TLV of 6.4 m³/t for CO₂ and 9.4 m³/t for CH₄ was considered safe under all circumstances, i.e. when mining near geological structures with a development advance rate up to 50 m/d. Lama also suggested that if the rate of development advance was reduced to 10-12 m/d, the Level 1 TLV could be safely increased by 20%. The Level 2 TLV of 10.0 m³/t for CO₂ and 12.0 m³/t for CH₄ was proposed for development mining in areas where no geological structures were present within 5.0 m of the excavation.

Details of the outburst incidents referred to by Lama (Figure 6) have not yet been located for review and verification of prevailing conditions. While details of the timing and proximity of the gas content sample locations relative to the outburst reference points A to I were not reported by Lama, the description of "Material thrown out = nil" suggests the incidents may have been gas blowers rather than outbursts. Also, given the gas content testing method used by Lama involved measurement of desorbed gas content and the addition of average residual gas content (Q₃) values to determine 'total gas content', it is suggested potential error may be present in the reported gas content values.
The gas content TLV prescribed by the DMR (Figure 5), which are lower than the TLVs proposed by Lama (Table 4 and Figure 7), indicates an additional level of conservatism and increased ‘factor of safety’ was applied by the DMR. Although the prescribed TLV was conservative (Black et al., 2009), the lower gas content threshold did achieve the objective of the DMR, which was to eliminate fatal outburst incidents in the Bulli seam. The removal of gas by gas drainage and the reduction of gas content to safe levels were uncritically accepted by the mining industry (Lama, 1995). Favourable conditions present in the mines at that time enabled the seam gas content to be reduced below TLV relatively easily and without delay to mine operations and coal production.

**IMPACT OF SEAM GAS COMPOSITION ON OUTBURST RISK**

The outburst TLVs adopted in Australian mines reflect the view that mining in coal with high concentrations of CO2 represents a significantly greater risk of outburst in comparison to mining in coal with high concentration of CH4.

Commonly reported views of researchers suggest (a) CO2 is more outburst prone than CH4 (Lama, 1995 and Williams, 2000), (b) CO2 outbursts are more violent than CH4 outbursts (Hargraves, 1980, Hanes, 2001, Lama and Saghafi, 2002), and (c) CO2 reduces the strength of coal (Wu, et al., 2010).

Wynne (2002) questioned the difference in threshold limit for CO2 and CH4, posing the question “is an outburst more likely in a CO2-rich seam than in a CH4-rich seam?”.

Results of gas content and gas composition data collected from core samples near recorded outburst events in Australian underground coal seams, presented in Figure 8, highlight the absence of outburst events below approximately 9.0 m³/t, specifically in conditions where CO2 is the dominant seam gas.

The basis for the commonly held view that outbursts associated with carbon dioxide are more violent, more difficult to control and more dangerous appears to be due to the greater sorption capacity of coal for carbon dioxide (Hargraves, 1980, Hanes, 2001, Lama and Saghafi, 2002). Laboratory based experimentation of outburst propensity using briquettes formed from pulverised coal in small-scale outburst simulation apparatus, such as those described by Skoczylas (2012), Wang *et al.* (2015) and Zhao *et al.* (2016), do not discuss the fact that when comparing the burst response of coal samples saturated with CO2 and CH4 at the same pressure, the effective gas content of the CO2 test sample will be approximately twice that of the CH4 test sample, due to the inherent sorption characteristics of coal. Consequently, for a given coal seam
gas pressure, coal samples will contain a larger volumes of carbon dioxide and emission problems therefore appear more acute (Beamish and O’Donnell, 1992).

The sorption capacity of Bulli seam coal for both CO2 and CH4, as reported by Black (2011) and presented in Figure 9, highlight the increased sorption capacity of CO2 in comparison to CH4. With reference to the isotherms for CO2 and CH4 presented in Figure 9, and considering the measurement of gas content is the principal measure of outburst risk; for a given gas content value the isotherms indicate the gas pressure of the CH4 rich sample will be substantially greater than the CO2 rich sample and therefore CH4 rich coal potentially contains gas at higher pressure than the CO2 rich coal and therefore potentially represents a greater outburst risk.

Figure 9: Example Bulli seam isotherm curves for methane and carbon dioxide sorption (after Black, 2011)

With respect to the impact of CO2 on coal strength, the effects of sorption induced swelling and potential weakening of coal samples observed in laboratory testing does not typically relate to in situ conditions experienced in Australian coal seams containing CO2. Australian coal seams containing high concentrations of CO2, such as the Hoskissons seam mined at Narrabri, the Greta seam mined at Austar and the Bulli seam mined at Metropolitan, Tahmoor and Appin do not routinely experience weakened coal conditions.

OUTBURST THRESHOLD LIMITS APPLICABLE TO NON-BULLI SEAM MINES

Outburst thresholds in non-Bulli seam mines are established based on the GeoGAS Desorption Rate Index (DRI) and acceptance of the outburst mechanism where the desorption rate of gas is directly used as an indicator of outburst proneness (GeoGAS, 2007). The background and relevance of using DRI as the basis for determining outburst TLV has been reviewed and discussed in Black (2018).

In the DRI approach, outburst proneness is regarded by GeoGAS as being directly related to the desorption rate of the coal. Bowen Basin coals (Goonyella Middle and German Creek seams) have higher DRI compared to the Bulli seam and, accordingly, the gas content thresholds are lower. For CH4, the Bulli seam gas content threshold is 9.5 m³/t at a DRI of 900. For the same DRI, the Goonyella Middle seam has a gas content of 7.0 m³/t and the German Creek seam (Middlemount/Tieri) a gas content of 7.7 m³/t (GeoGas, 2007).

The DRI900 method was proposed, based on a review of the Bulli seam threshold for ‘normal mining’, which is effectively equivalent to a TLV for structured coal, and no work was done to establish a method to determine TLV for non-structured coal.

Kidybinski (1980) in Lama (1995) recognised that factors other than gas pressure and gas content play an important role in promoting instantaneous outbursts of coal and gas in coal seams. Outbursts often occur in coal weakened by local geological distortion. Strength variations within the coal, due to tectonic and sedimentary conditions are often greater than strength variations due to variations in gas pressure and desorption phenomena. Kidybinski further suggested that coal strength, coal weakness, may have a greater effect on local outburst hazard than gas pressure and desorption characteristics. These additional, and potentially more significant outburst risk factors are not considered in the GeoGAS DRI900 approach to determination of outburst TLVs in Australian coal seams.

MINING EXPERIENCE ABOVE NORMAL OUTBURST GAS CONTENT THRESHOLD LIMITS

In the years following the 1994 introduction of the Bulli seam TLVs there has been significant advances in directional drilling technology and the standard of management plans used at most mines to identify, assess and control outburst risk. Also, many mines have progressively moved into areas of increased gas content and reduced permeability, where it is becoming increasingly difficult to drain gas below the ‘normal mining’ TLV and mine operators are questioning the appropriateness of the outburst TLVs (Black and Aziz, 2008 and Black et al., 2009).

Several Australian underground coal mines have completed formal reviews of their outburst management plans which led to increasing TLV supported by additional management controls, such as increased drilling density and increased gas content testing. Increased TLV, presented in Figure 10, were approved for Tahmoor colliery in 2003 and West Cliff colliery in 2005. In the years following the changes to the TLV, both collieries operated without an outburst incident. Metropolitan colliery has also reviewed and introduced additional TLVs to allow controlled mining in areas where gas content remains above the 1994 ‘normal mining’ TLV.
At Tahmoor colliery, in addition to the Level 1 TLV, below which no restrictions are placed on mining, introduced two additional TLV levels. The Level 2 TLV applies to structured coal and where the measured gas content is greater than Level 1 and less than Level 2, in addition to more intensive drilling and coring, the rate of development advance is restricted to 12 m/day. The Level 3 TLV applies to coal free of geological structures. Where the measured gas content is greater than Level 1 and less than Level 3, in addition to increased drilling and gas content testing, the rate of development advance is restricted to 25 m/day in each heading and cut-through to a maximum of 75 m in any 24 hour period. In areas where gas content remains above the defined TLV, normal mining is prohibited and grunching is the only approved development mining method.

At West Cliff colliery, in addition to the Level 1 TLV, one additional TLV was introduced. While no restrictions were placed on the rate of development advance, where the measured gas content was between the Level 1 and Level 2 TLV increased drilling, structure identification and gas content testing was required. Where the gas content remained above the Level 2 TLV, normal mining was prohibited and an alternative mining method, such as remote control or grunching, was required.

Figure 11 shows gas test results from areas of the Bulli seam where gas content was above the ‘normal mining’ outburst threshold limit that were mined by non-standard methods without inducing an outburst. The figure shows gas data from areas mined by (a) fully remote controlled continuous miner operation, (b) grunching using conventional shotfiring, and (c) limited rate mining where limits are placed on the maximum hourly and daily rate of advance of conventional continuous mine development operations. In the 15 year period that Tahmoor colliery has employed limited rate mining through structured and non-structured coal, with gas content up to 12.0 m³/t (CH₄) and 10.0 m³/t (CO₂), an outburst has not occurred (Borg, 2014). Tahmoor has also mined over 3,000 metres of roadways, by grunching due to inability of conventional pre-drainage to reduce gas content of the coal seam below the original TLV, without inducing an outburst in ‘tight’ coal with gas content up to 14 m³/t (Wynne, 1999, 2000, 2011).

Blanch (2017) raised concerns in relation to the use of limited rate mining, suggesting there was increased risk of an outburst event occurring if mining were to be undertaken in areas where the gas content remained above the 1994 ‘normal mining’ TLV. Tahmoor’s 15 years’ experience mining through such areas, along with similar experience at West Cliff, Metropolitan and other Australian collieries, does suggest (a) the use of limited rate mining is an effective control to mine in areas of increased gas content, or (b) the 1994 TLV are very conservative and the gas content levels in those areas did not present an increased outburst risk, particularly in areas where no geological structurers are present. Further
investigations are planned to assess seam gas pressure in advance of development working faces and the impact of mining rate on the seam gas pressure profile.

Figure 12 presents gas test results from locations where outburst have occurred in areas of the known outburst risk in the Bulli seam that were mined using remotely operated continuous miner and gruncheing methods. Like the historical outburst events data presented in Figure 8, the recently acquired outburst data presented in Figure 12 highlights the absence of outburst events below approximately 9.0 m³/t, and does not support the view that coal rich in CO₂ is at greater risk of outburst. Further investigations into the impact of gas composition on outburst risk are planned.

CONCLUSIONS

Outbursts represent a major safety hazard to personnel working near the coal face in areas of increased outburst risk. The current approach to TLV have been effective in preventing injury from outburst however increasing evidence, based on operational experience, suggest the current method is overly conservative in some conditions, to the point of adversely impacting mine productivity without delivering significant incremental increase in safety.

While abnormal geological conditions have been linked to all fatal outburst incidents and at least 98% of the non-fatal outburst incidents recorded in Australia, the presence or absence of geological structure is typically not reflected in outburst TLVs in Australian mines. Gas content is recognised as having the most significant impact on outburst risk and gas drainage to reduce gas content to safe levels plays a significant role in control and reduction of outburst risk in Australian underground coal mines.

Investigations into Australian outburst history and mining experience in areas where gas content was above the 1994 Bulli seam outburst threshold limits has provided no evidence that outburst had occurred at gas content levels below approximately 9.0 m³/t, independent of gas composition and geological conditions. Using limited rate mining methods, mining has been carried out, without outbursts, in areas where gas content TLV has been equivalent to 12.0 m³/t (CH₄) and 10.0 m³/t (CO₂).

While China, Russia and other European countries assess outburst risk through measurement of gas emission rate from fresh cut coal samples, using outburst indices such as $\Delta P$, $\Delta P_{0-60}$, $\Delta P$ Express and KI Index, Australia is the only country that uses a measure of gas emission from crushed coal during the Q3 gas content testing as the basis for establishing outburst TLVs.

Coal mining practice in Australia requires rapid mining rates to sustain high productivity retreating longwalls which in turn rely on effective systems to identify areas of increased outburst risk and effective treatments to reduce the outburst risk in advance of mining operations.

Control and management of outburst risk, including measures to predict and reduce outburst risk in advance of planned mining, must be effective and continue to support the high safety and production targets of the Australian underground coal mines.

Further work will continue in association with the University of Wollongong to (a) investigate the impact of gas composition, coal toughness and gas pressure on outburst risk, and (b) develop a multi-factor Outburst Risk Index appropriate for assessing outburst risk in Australian mining conditions.

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