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DESIGN OF VENTILATION AND SUPPRESSION SYSTEMS FOR FIRES IN ROAD TUNNELS

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ABSTRACT

The interaction between water mist systems and ventilation systems in tunnel fires was modelled by creating a MATLAB program to solve a system of equations generated using comprehensive thermodynamic control volume analyses. The program uses input values for relevant variables in order to estimate the number of jet fans required to prevent smoke backlayering in a tunnel fire. The input variables include tunnel dimensions, water mist system characteristics, and fire size.

A case study was performed on a 6 x 6 x 600 m tunnel in keeping with previous research. It was found that the use of a water mist system had a mixed impact on the number of fans required to prevent backlayering. At heat release rates lower than approximately 15 MW, it was found that the water mist system increased the number of fans required. The main reason for this was determined to be the resistance caused by accelerating the relatively large mass of water introduced into the tunnel. At heat release rates larger than approximately 15 MW, it was found that the water mist system reduced the number of fans required. This was determined to be due to the ability of the water mist system to reduce the heat release rate and to reduce the downstream temperatures.

Several recommendations for future work were made. They mostly concerned collecting more full-scale experimental data in several areas, including: determining the relationship between water mist system characteristics and heat release rate, determining what proportion of water spray evaporates in a tunnel fire, measuring the throttling effect of a fire on longitudinal airflow, and determining the relationship between longitudinal ventilation velocity and water mist systems.

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NOTATION

Α	Area (m ²)	Ŵ	Rate of work (W)
C_d	Drag coefficient (-)	Ζ	Elevation (m)
D_h	Hydraulic diameter (m)		
Ε	Energy (J)		
f	Friction coefficient (-)		
F	Force (N)		
g	Gravity (m/s^2)		
Н	Height (m)	Greek s	ymbols
h	Specific enthalpy (J/kg)	α	Thermal diffusivity (m ² /s)
k	Heat release rate enhancement due to forced ventilation (-)	Xconv	Convective fraction (-)
K _{en}	Entrance loss coefficient (-)	Xrad	Radiative fraction (-)
K _j	Jet fan coefficient (-)	ρ	Density (kg/m^3)
K _{nozzle}	Nozzle coefficient (-)	$\Delta H_{c,eff}$	Effective heat of combustion (J/kg)
K _{traffic}	Traffic density (-)	ψ	Heat release rate enhancement due to enclosure effects (-)
k _{WMS}	Heat release rate reduction due to water mist system operation (-)	ν	Kinematic viscosity (m ² /s)
L	Length (m)		
'n	Mass flow rate (kg/s)		
$\dot{m}_{WMS}^{\prime\prime}$	Water density $(1/min/m^2)$		
$\dot{m}_{WMS}^{\prime\prime\prime}$	Water flux density (l/min/m ³)		
М	Molar mass (kg/mol)		
p	Pressure (Pa)		
Ż	Heat release rate (W)		
\dot{Q}^{*}	Dimensionless heat release rate (-)		
\overline{R}	Ideal gas constant (J/kg/mol)		
Т	Temperature (K)		
V	Velocity (m/s)		
V^*	Dimensionless velocity (-)		
W	Width (m)		

1. INTRODUCTION

1.1. Overview

Road tunnels are an essential part of many transport networks around the world. They are used to move people and cargo from place to place in a timely manner. Societies expect these tunnels to have an acceptable level of safety for users, at a reasonable economic cost. The risk of fire in a road tunnel poses a major safety hazard to people and can cause extensive damage requiring a lot of time and money to repair. As such, it is important that designers adequately address the threat of fire. Industry attempts to do this by formulating and implementing fire safety strategies.

The main goals of a fire safety strategy in a road tunnel are related to life safety and property protection. According to the Tunnel Study Centre (CETU), the main threats to life caused by fire in a road tunnel are exposure to toxic gases and high temperatures [1]. Reduced visibility due to smoke exacerbates these issues, as it can impair self-evacuation and lead to increased exposure times. The main threat to the tunnel itself is high temperatures, which can cause concrete spalling, damage to the road surface, and destruction of services installed in the tunnel. This leads road tunnel designers to try to minimise the temperature and human exposure to toxicity during fires.

There are many possible approaches that can be taken to mitigate the threat posed by fires in tunnels. Given the acceptance and widespread implementation of sprinklers in buildings around the world, it would seem that similar systems could be used effectively in tunnels. However, the World Road Association (PIARC) produced a report in 1999 saying that sprinklers are not useful for saving lives in tunnel fires [2]. This maintained the theme of previous reports by PIARC dating back to at least 1983 [3]. The reasons cited include: the inability to suppress fires inside vehicles, the potential to cause explosions through dispersing petrol or other chemicals, vaporised steam burning tunnel occupants, reduced visibility, and destratification of smoke exposing people to toxic gases. The report discusses longitudinal ventilation systems, where smoke is pushed out one end of the tunnel using jet fans. It also examines transverse and semi-transverse systems, where smoke is extracted through ducts. The report gives more weight to the use of such ventilation systems in fire strategies.

However, in 1999 and the years since, there have been many examples of major fires in road tunnels. Carvel and Marlair provided a summary of fatal or otherwise significant fires in tunnels over the last 100 years or so [4]. One of the most well known is the Mont Blanc tunnel fire in 1999, which resulted in 39 deaths and extensive damage to the tunnel. An event of this scale is clearly unacceptable, and a response from the industry was required.

Designers have since moved towards the use of fixed fire fighting systems (FFFSs). This is probably partly due to the severity of fires such as the Mont Blanc event, where the heat release rate (HRR) was estimated at approximately 190 MW [4]. Full-scale tunnel fire experiments conducted by Ingason measured HRRs in excess of 200 MW [5]. Fires this large are likely to create untenable conditions for people and cause major damage to the tunnel. Deluge sprinkler systems can help prevent fires growing to these sizes, but they use a lot of water. This results in large water storage, pumping, and drainage costs. Water mist systems (WMSs) have shown some ability to limit the HRR and temperatures in full-scale tunnel fire experiments [6][7]. They also use much less water than deluge systems, which can result in substantial monetary savings. Such systems have recently been installed in major urban road tunnels such as the A86 tunnel near Paris and the M30 Madrid ring road [8]. By installing WMSs in tunnels, designers hope to control the HRR of fires and subsequently limit temperatures, smoke production and flame spread.

At present, designers assume a design HRR and then design the ventilation system ignoring the effect of the additional mass of water introduced by WMSs. This additional mass, and the subsequent volume of steam produced, could impact the ability of the ventilation system to control smoke produced by the fire. Longitudinal ventilation systems are often designed to prevent the occurrence of backlayering; a phenomenon where smoke flows upstream of the fire, against the direction of forced ventilation. This gives rise to the question: how does a WMS impact on the ability of a longitudinal ventilation system to prevent backlayering in tunnel fires? Previous work by Looi [9] attempted to answer this question using the software programme Fire Dynamics Simulator [10]. This thesis is a continuation of Looi's, but will attempt to answer the question using a one-dimensional control volume model implemented numerically in MATLAB [11].

1.2. Tunnel Fire Dynamics

Most of the current fire dynamics knowledge is based on open fires or fires in a building compartment. Tunnel fires differ significantly from both open fires and compartment fires. Severe tunnel fires observed around the world have brought tunnel fire dynamics into focus [12].

In open fires, much of the heat released is lost to the surroundings. Contrastingly, in tunnel fires there is a lot more heat feedback due to the enclosed nature of the environment. This feedback can cause objects that would not sustain burning in open conditions to burn vigorously in tunnels. In open fires, oxygen is readily available to facilitate combustion. This may not be the case in tunnel fires. Due to the variable availability of oxygen, fires in tunnels can be either fuel controlled or ventilation controlled. In the case of ventilation-controlled fires, the smoke produced contains high levels of toxic fumes and a substantial amount of unburnt fuel. Furthermore, in tunnel fires there is substantial interaction with the ventilation flow. This can lead to smoke backlayering or a type of resistance to the flow known as throttling [12].

There are several key differences between tunnel fires and compartment fires. In compartment fires, the maximum HRR is typically dictated by the ventilation factor, a function of the openings in the compartment. In tunnels, the maximum HRR depends firstly on whether the tunnel is ventilated naturally or mechanically. In the case of natural ventilation, the tunnel slope, dimensions, wall material, and weather conditions all impact the maximum HRR. With mechanical ventilation, the forced ventilation velocity has a large bearing on the maximum HRR. In compartments, flashover can occur due to the potential build-up of hot gases inside a space. This is not likely to occur in tunnels, as the hot gases can easily escape out the portals and there are substantial heat losses to the walls. In compartment fires, a stratified smoke layer is formed in the upper portion of the compartment. This can be observed at the initial stages of a tunnel fire, however stratification is generally lost at distances far from the fire. Newman described three stratification regions, which can be seen in **Error! Reference source not found.** [13].



Fig. 1. Three stratification regions as described by Newman [13].

The three regions defined by Newman [13] can be described as follows:

- Region I, Fr ≤ 0.9: Obvious stratification. High temperatures at ceiling level, temperatures approximately equal to ambient at ground level. Buoyancy dominated temperature stratification.
- Region II, 0.9 < Fr ≤ 10: No clear stratification, but substantial temperature gradients. Strong interaction between buoyancy forces and the imposed horizontal flow.
- Region III, Fr > 10: Insignificant stratification. Minimal temperature gradients.

The Froude number can be calculated using Eq. (1), as follows [12]:

$$Fr = \frac{u_{avg}^2}{1.5(\Delta T_{avg}/T_{avg})\,gH}\tag{1}$$

where H is the ceiling height, T_{avg} is the average gas temperature (K) over the entire cross-section at a given position, $\Delta T_{avg} = T_{avg} - T_a$ (the average gas temperature rise (K) above ambient over the entire cross-section at a given position), and $u_{avg} = uT_{avg}/T_a$ (m/s) [12].

1.3. Heat Release Rates in Tunnels

The HRR in a tunnel fire is dependent on many factors, including the tunnel construction, the type and number of vehicles in the tunnel, and the ventilation conditions [14]. The HRR is a key parameter in the design of safety systems in tunnels. Typically, designers use tabulated peak HRR values from sources such as PIARC [2] or the National Fire Protection Association (NFPA) [15]. Ingason summarised the results of most full-scale tunnel fire experiments performed in the

past 25 years, which also provides data that could be used for design purposes. This summary found that the peak HRR ranges from 1.5-11 MW for cars, 25-34 MW for buses, and 13-202 MW for heavy goods vehicles (HGVs) [14]. These experiments were carried out under a range of different circumstances, with important factors such as the ventilation conditions varying between tests. It can be seen that HGVs result in the highest HRRs. As such, they are commonly considered to be the highest risk when designing tunnel ventilation systems.

Carvel et al. proposed the following equations to predict the relationship between the HRR in a naturally ventilated tunnel and the HRR in open, ambient conditions [16][17][18]:

$$\dot{Q}_{tunnel} = \psi \dot{Q}_{open} \tag{2}$$

$$\psi = 24 \left(\frac{W_F}{W_T}\right)^3 + 1 \tag{3}$$

where ψ is the enclosure enhancement factor, W_F is the width of the fire object, W_T is the width of the tunnel. This equation is only valid for W_F/W_T values up to 0.5.

Carvel and Beard proposed the following equation quantifying the HRR enhancing effect of forced ventilation in HGV fires [19]:

$$\dot{Q}_{ventilated} = k\dot{Q}_{tunnel} \tag{4}$$

where k is the forced ventilation enhancement factor, which depends on the forced ventilation velocity. A series of curves were proposed to quantify the value of k in the growth phase and for fully involved HGV fires. The value of k was presented for various percentile values and the expected value. Curves were presented for both one-lane and two-lane tunnels. These curves show that forced ventilation can result in substantially higher HRRs than would otherwise be observed. For example, the expected k value is approximately 4 for an HGV fire in the growth phase with a forced ventilation velocity of 3 m/s in a one-lane tunnel.

1.4. Tunnel Ventilation

1.4.1. Ventilation Systems

There are two main types of ventilation systems used in tunnels: longitudinal and transverse. A longitudinal system moves air along the length of the tunnel. A transverse system moves air both along the length of the tunnel and transverse to the tunnel, depending on the type of transverse system [20].

Ventilation systems can also be divided into naturally and mechanically driven systems. Naturally ventilated systems predominantly rely on meteorological conditions and the longitudinal flow generated by moving vehicles (known as the piston effect). The main meteorological effects driving natural ventilation are wind, elevation differences, and temperature differences. However, these effects are generally not reliable enough to base a fire strategy on. As such, a mechanical system is often required [20].

There are three main types of mechanically powered longitudinal systems: injection, jet fans, and push-pull systems. Injection systems use a fan to push air through a nozzle at ceiling level, which induces a flow through the tunnel. Jet fan systems use a series of jet fans along the length of the tunnel, which accelerate the air and create a longitudinal flow in the tunnel. Push-pull systems use reversible fans located in ventilations shafts at ceiling level; fans upstream of the fire are set to supply mode while downstream fans are set to extract mode [20].

There are three main types of mechanically powered transverse systems: fully transverse, semi-transverse – exhaust, and semi-transverse – supply. A fully transverse system uses a full-length supply duct and a full-length exhaust duct. Semi-transverse – exhaust systems employ a full-length exhaust duct but no supply duct. Semi-transverse – supply systems are the opposite; they have a full-length supply duct but no exhaust duct [20].

1.4.2. Throttling Effect

The throttling effect of a fire in a tunnel causes increased resistance to longitudinal flow. This is partially due to the buoyancy of combustion products generating aerodynamic resistance [12]. Lee et al. experimentally measured this phenomenon in 1979. It was found that the flow resistance in the fire zone increased by a factor of 6,

and a factor of 1.5 in the upstream and downstream regions due to the throttling effect [21]. Colella et al. numerically investigated the effect using a combination of 3-dimensional computational fluid dynamics (CFD) and 1-dimensional modelling techniques. It was found that the flow was reduced by up to 30% in the case of a 100 MW fire [22]. Carvel et al. [23] used FDS [10] to simulate the throttling effect. It was found that larger fires required more jet fans to prevent backlayering. These studies consistently detected the presence of the throttling effect, but did not quantify it in a generic way.

Hwang and Chaiken proposed the following formula to quantify the throttling effect [24]:

$$F_{throttling} = \dot{m} \left(V_{upstream} - V_{downstream} \right) \tag{5}$$

where \dot{m} is the mass flow rate in the fire zone CV.

Dutrieue and Jacques used the CFD code FLUENT [25] to simulate the effect of the throttling effect. They proposed the relationship expressed in Eq. (6) [26]. This is an empirical equation, so the units used are important and have been specified below.

$$F_{throttling} = \frac{\dot{Q}^{0.8} V_{upstream}^{1.5}}{D_h^{1.5}} \cdot C \cdot A_{tunnel}$$
(6)

where \dot{Q} is the heat release rate (W), $V_{upstream}$ is the upstream velocity (m/s), D_h is the hydraulic diameter (m), C is an empirical constsant equal to $41.5 \times 10^{-6} \ s^{1.9} k g^{0.2} / m^{2.6}$, and A_{tunnel} is the cross-sectional area of the tunnel (m²).

The models proposed by Hwang and Chaiken [24] and Dutrieue and Jacques [26] were compared by Fleming et al. [27]. It was found that the models gave very similar results when used in a 1000 m long, 40 m² cross-section tunnel with a 20 MW fire.

1.4.3. Tunnel Ventilation Velocity

Jang and Chen proposed governing equations for the flow in tunnels ventilated longitudinally using jet fans. These equations are as follows [28]:

$$\sum_{i=1}^{N} F_i = \rho A_t L \frac{dV}{dt}$$
(7)

where *F* is the force, ρ is air density, A_t is the cross-sectional area of the tunnel, *L* is the length of the tunnel, *V* is the velocity, *t* is the time, and the forces are calculated as follows [28]:

$$F1 = \sum_{j=1}^{J} \frac{\rho}{2} C_{d,j} (V_j - V) |V_j - V| A_{v,j}$$
(8)

$$F2 = -f\frac{\rho}{2}\frac{L}{D_h}A_t V|V|$$
⁽⁹⁾

$$F3 = (p_{in} - p_{out})A_t \tag{10}$$

$$F4 = N_F \rho A_F |V_F| (V_F - V) K_j \tag{11}$$

$$F5 = -K_{en}\frac{\rho}{2}A_t V|V| \tag{12}$$

where $C_{d,j}$ is the drag coefficient of vehicle j, V_j is the velocity of vehicle j, $A_{v,j}$ is the area of vehicle j, f is the tunnel friction coefficient, D_h is the tunnel hydraulic diameter, A_t is the cross-sectional area of the tunnel, p_{in} is the pressure at the inlet portal, p_{out} is the pressure at the outlet portal, N_F is the number of jet fans, A_F is the cross-sectional area of a single jet fan, V_F is the outlet velocity of the fan, K_j is the jet fan coefficient, and K_{en} is the entrance loss coefficient.

In the case of a fire in a tunnel, an additional force to account for the throttling effect could be added based on the equations presented in section 1.4.2.

At steady state, $\frac{dv}{dt} = 0$ and hence Eq. (7) is equal to zero. This means that the sum of the forces acting must also be equal to zero. This force balance can be solved to estimate the number of fans required to achieve steady state at a desired velocity.

Longitudinal ventilation systems are often designed to prevent backlayering. This is achieved by generating an upstream velocity larger than a value known as the critical ventilation velocity (CVV). The CVV is one of the most studied aspects of tunnel fire safety. There are two main types of models that have been used to predict the CVV: critical Froude models and non-dimensional models. According to Ingason et al., a constant critical Froude number does not exist, and consequently these models are not a reasonable way to estimate the CVV [29].

Oka and Atkinson conducted small-scale experiments and subsequently proposed the following non-dimensional model [30]:

$$V_{crit}^{*} = K_{v} \left(\frac{\dot{Q}^{*}}{0.12}\right)^{\frac{1}{3}} for \, \dot{Q}^{*} \le 0.12$$

$$V_{crit}^{*} = K_{v} \quad for \, \dot{Q}^{*} > 0.12$$
(13)

where

$$\dot{Q}^* = \frac{\dot{Q}}{\rho_{\infty} T_{\infty} c_p g^{0.5} H^{2.5}} \tag{14}$$

$$V_{crit}^* = \frac{V_{crit}}{\sqrt{gH}} \tag{15}$$

where V_{crit}^* is the dimensionless CVV, K_v is the experimental constant, \dot{Q}^* is the dimensionless HRR, ρ_{∞} is the ambient density of air, T_{∞} is the ambient temperature of air, c_p is the ambient specific isobaric heat capacity of air, g is gravity, and H is the tunnel height.

Wu and Bakar conducted small-scale experiments and proposed a similar model [31]:

$$V_{crit}^{*} = 0.40 \left(\frac{\dot{Q}^{*}}{0.12}\right)^{\frac{1}{3}} for \, \dot{Q}^{*} \le 0.2$$

$$V_{crit}^{*} = 0.40 \quad for \, \dot{Q}^{*} > 0.2$$
(16)

where

$$\dot{Q}^* = \frac{\dot{Q}}{\rho_{\infty} T_{\infty} c_p g^{0.5} D_h^{2.5}}$$
(17)

$$V_{crit}^* = \frac{V_{crit}}{\sqrt{gD_h}} \tag{18}$$

Where D_h is the hydraulic diameter of the tunnel, calculated as follows:

$$D_h = \frac{4A}{P} \tag{19}$$

where *A* is the cross-sectional area of the tunnel and *P* is the wetted perimeter.

According to Ingason et al. [29], the models proposed by Oka and Atkinson [30] and Wu and Bakar [31] are questionable as both used water spray devices to cool the walls during experiments. This could have increased heat losses to the surroundings and hence impacted the reliability of the results obtained.

Li et al. conducted model-scale tests and compared the results to full-scale data. The following model was proposed [32]:

$$V_{crit}^* = 0.81 \dot{Q}^{*1/3} \quad for \ \dot{Q}^* \le 0.15$$

$$V_{crit}^* = 0.43 \quad for \ \dot{Q}^* > 0.15$$
(20)

where \dot{Q}^* and V_{crit}^* are defined by Eq. (14) and Eq. (15) respectively.

Li et al. [32] also proposed the following equation quantifying the effect of vehicle obstruction on the CVV:

$$\varepsilon = \frac{V_{crit} - V_{crit,ob}}{V_{crit}} = \frac{V_{crit}^* - V_{crit,ob}^*}{V_{crit}^*}$$
(21)

where ε is the reduction ratio and the subscript *ob* indicates vehicle obstruction. It was found that the value of ε was approximately equal to the blockage ratio in the tunnel. Similar results were found by Oka and Atkinson [30], Lee and Tsai [33], and Jomaas et al. [34].

1.5. Fixed Firefighting Systems

1.5.1. Overview

Water based fire suppression systems have been widely implemented in buildings for more than 100 years. Early suppression-fast response (ESFR) systems are commonly used in a variety of contexts. They use large droplets that have sufficient momentum to penetrate flame. These systems are effective when fires can be quickly detected and located. This is suitable for many applications, such as in warehouses. However, in tunnels the fire may occur in a vehicle that is moving, making accurate detection difficult. Longitudinal ventilation flows also hinder detection. As such, ESFR systems are not commonly used in tunnels [35].

Despite the acceptance of sprinkler systems in buildings and other contexts, they have not widely embraced by the European and North American tunnel industries. Both PIARC and the NFPA have long been reluctant to recommend their use. The concerns raised by these organisations include: the potential to cause petrol or other chemical explosions, steam burning occupants, the inability to control fires inside vehicles, smoke stratification is lost, reduced visibility, and maintenance issues. However, a series of major tunnel fires in recent years has caused the industry to respond [35].

Designers have begun to use WMSs in tunnels. These systems use very small water droplets, as opposed to the larger, faster droplets used by ESFR systems. These systems use a fine water mist where 99% of the spray volume is made up of droplets with a mean diameter less than 1000 μ m. These systems use much less water than traditional ESFR systems, saving money, reducing environmental concerns, and limiting water damage to the tunnel during operation. The spray in WMSs is usually generated through atomisation of water by pressurised jet at low, medium or high pressure [35]. These pressures are defined by NFPA 750 as follows [36]:

- Low: operating pressure \leq 12.5 bar
- Medium: operating pressure 12.5-35 bar
- High: operating pressure \geq 35 bar

1.5.2. Dynamics of Suppression by Water Sprays

There are three main mechanisms that WMSs use to suppress flames: heat extraction, displacement, and radiation attenuation. Heat extraction by water sprays is dominated by the evaporation of water. It takes approximately 330 kJ of energy to heat 1 kg of water from 20 °C to 100 °C, and approximately 2300 kJ to evaporate it. When water evaporates, it rapidly expands to about 1700 times its liquid volume. If the vapour is heated up to a typical flame temperature of 800 °C, its volume is likely to be around 5000 times bigger than it was before evaporation. This expansion displaces oxygen, and dilutes vapourised fuel. This can cause the mixture to fall below its lower flammability limit, extinguishing the flame. Water sprays are considered to attenuate radiation in two main ways: shielding by liquid droplets, and attenuation by vapourised water. Liquid droplets reduce radiant transmission by absorbing and scattering radiation. Gaseous steam particles dispersed within the smoke reduce the emission and transmission of radiation. Various researchers have found that the attenuation of radiation by water mists appears to depend weakly on the size of water droplets, but more heavily on the ratio of water mass to air mass. [35].

There are several aspects of water spray fire suppression that are not well understood. Most full-scale WMS testing has been done using open pool fires or uncovered piles of wooden pallets. These fuel loads are not likely to occur in a real tunnel fire, and may not accurately represent real-life fuel loads. Hence the test results must be treated with some caution. The influence of longitudinal ventilation on WMS sprays is also poorly understood [35]. Modelling performed by Rein et al. found that 'emergency' ventilation conditions can cause water mist droplets to be carried hundreds of metres in a tunnel before reaching ground level [37]. This brings the reliability of WMSs into question when used in conjunction with longitudinal ventilation.

1.7. Modelling of Tunnel Fires

There are several types of models used to predict fire behaviour, including CFD, onedimensional, and control volume (CV) modelling. Each type of model has strengths and weaknesses that must be understood by designers making decisions based on the output of such simulations. Fires in tunnels result in complex three-dimensional flow phenomena. Tunnel geometry, ventilation conditions, and other factors interact and result in a range of behaviours. CFD models find a solution to the Navier-Stokes equations using appropriate boundary conditions. The complex physical interactions that occur in tunnel fires can be simultaneously modelled in order to quantify their influence on the overall behaviour. However, the physical behaviour of such problems is not fully understood, and some mathematical assumptions need to be made to obtain solutions. It is important for users to understand these limitations in order to reduce uncertainties to acceptable levels. CFD models are much cheaper to perform than full-scale experiments, and can approximate tunnel fire behaviour with acceptable accuracy when used properly. However, they can be computationally expensive [38].

One-dimensional models divide the tunnel geometry into a series of nodes and branches. It is assumed that the properties are constant over the height at any cross-section in the tunnel. One of the main limitations of one-dimensional models is when there are vertical temperature or velocity gradients. One-dimensional models are relatively simple to use, and are computationally inexpensive [39].

Control volume models work by dividing the tunnel geometry into a series of CVs and then analysing them. The properties throughout each CV are assumed to be constant. Conservation equations are applied in order to predict the properties in each CV. One of the main limitations of CV models occurs when the flow behaviour departs from the conceptual basis on which the model was created. CV models have the advantage of being computationally inexpensive and relatively easy to use [40].

1.8. Problem Statement and Objectives

The problem has been introduced above, and can be summarised by the following problem statement:

Tunnel designers have recently moved towards fire strategies using water mist systems to control the heat release rates in tunnel fires. However, when designing longitudinal ventilation systems to prevent backlayering, they do not currently account for the additional mass and steam introduced to the tunnel by the water mist system. This thesis aims to quantify the interaction between longitudinal ventilation systems and water mist system operation in tunnel fires.

2. METHODOLOGY

2.1. Overview

A MATLAB [11] programme was created to determine the number of jet fans required to prevent smoke backlayering in the event of a tunnel fire. The program was designed so that it could provide results for different tunnel geometries, water mist systems, design fires, and other relevant inputs defined by the user. The MATLAB files created can be found in Appendix A.

Once all of the input parameters have been defined, the program divides the tunnel into four sections: inlet, upstream, fire zone, and downstream (Fig. 2). The downstream section is further divided into multiple CVs in order to increase the accuracy of the solution. Conditions outside of the tunnel are given assumed ambient values by the user. The schematic was created by the author using draw.io [41].



Fig. 2. Generic problem domain. Created using draw.io [41].

It can be seen in the diagram that the fire zone and downstream CVs are grey. This is to denote the presence of smoke in these sections. Strictly speaking, to avoid backlayering there should be no smoke upstream of the fire source. However in this study it is assumed that each CV has constant properties throughout. As such the entire fire zone CV is assumed to be filled with smoke, even though a small amount may be upstream of the fire source.

In each CV, the values of four key properties are calculated: static pressure, gas temperature, gas density, and gas velocity. These values are used to calculate the forces acting on the flow in each CV. It is assumed that the system is at steady state, and the properties are constant throughout each CV. The steady state assumption is used to simplify the problem. The constant properties assumption was introduced in section 1.7. Once the key properties and forces acting on the flow in each CV have

been calculated, a force balance is performed. The calculation procedure is iterated until two conditions are met: the forces acting in the tunnel are balanced, and the upstream velocity is greater than the CVV. The algorithm used to solve the problem is briefly introduced below, with each step explained in more depth later in this chapter.

- 1. The user defines key parameters of the problem including tunnel dimensions and the HRR in open, still, ambient conditions
- 2. The CVV is calculated
- 3. The initial velocity at the inlet portal is set as equal to the CVV, the initial number of jet fans is set to one, and initial values for other values are defined
- 4. The HRR is adjusted to account for enclosure effects, forced ventilation, and suppression by the WMS
- 5. The key properties and the forces acting are calculated in each CV
- 6. A force balance is performed on the tunnel
- 7. If the force balance is negative, another fan is added and the algorithm returns to step 4
- 8. If the force balance is positive, the ventilation velocity is increased and the algorithm returns to step 4
- 9. If the force balance is sufficiently close to zero, the solution is deemed to have converged

Fig. 3 shows the algorithm in the form of a flow chart. The MATLAB script file "MasterScript" was written to carry out this algorithm and can be found in Appendix A1.



Fig. 3. Calculation process. Created using [41].

The following sections aim to explain and justify each part of the algorithm.

2.2. Algorithm

2.2.1. Problem Initialisation

The problem is initialised in steps 1-3 defined above. These steps are contained within the MATLAB script file "initialiseProblem", which can be found in Appendix A2.

In step 1, the user specifies a range of relevant parameters. The user specifies these values in the MATLAB script file "Parameters", which can be found in Appendix A3.

Example input values are tabulated in Appendix B. Appropriate values and references are provided for parameters that are likely to be independent of the tunnel, such as the drag coefficient of an average car.

In step 2, the CVV is calculated according to **(20)** proposed by Li et al. [32], repeated below for convenience. It is assumed that the CVV is not reduced due to the effect of blockage as described by Li et al. [32]. This is a conservative assumption, as a larger CVV requires more thrust and hence more fans.

$$V_{crit}^{*} = 0.81 \dot{Q}^{*1/3} \quad for \ \dot{Q}^{*} \le 0.15$$

$$V_{crit}^{*} = 0.43 \quad for \ \dot{Q}^{*} > 0.15$$
(20)

The dimensionless CVV is assumed to be equal to its maximum theoretical value of 0.43. This is again a conservative assumption. As such, it is implicitly assumed that the dimensionless HRR is larger than 0.15. It is reasonable to assume that the dimensionless HRR is this big, as full-scale experiments have demonstrated that very high HRRs can occur in tunnel fires involving HGVs [5]. For example, consider a large road tunnel with a height of 6 m. A HRR of 15 MW is large enough for the dimensionless HRR to exceed 0.15. It has not been demonstrated that WMSs can reliably control HRRs in tunnel fires to 15 MW. As such it was deemed reasonable to conservatively assume that the dimensionless HRR exceeds 0.15. The MATLAB script file "Critical_ventilation_velocity" is used to perform the calculations for step 2 (Appendix A4).

In step 3, the upstream velocity is set as equal to the CVV, and the number of fans is set to an initial value of one. Other parameters are given initial values so that the first iteration of the algorithm can be completed. The solution is independent of the initial values given to these other parameters. These operations are contained in the "initialiseProblem" script file.

2.2.2. Heat Release Rate Adjustment

The HRR previously defined for open, still, ambient conditions is adjusted to account for enclosure effects, forced ventilation, and suppression by the water mist system. The MATLAB script file used to perform these calculations is called "HeatReleaseRateAdjustment" and can be found in Appendix A5. These calculations also require the use of "interpolateCustom", which can be found in Appendix A6.

The increase in HRR due to enclosure effects was modelled according to Eq. 2 and Eq. 3 proposed by Carvel et al. [16][17][18], repeated below. These equations are only valid if the width of the fire source is less than or equal to half the width of the tunnel.

$$\dot{Q}_{tunnel} = \psi \dot{Q}_{open} \tag{2}$$

$$\psi = 24 \left(\frac{W_F}{W_T}\right)^3 + 1 \tag{3}$$

The increase in HRR due to forced ventilation velocity was modelled using the curves proposed by Carvel and Beard [19]. The HRR is increased by a scale factor k according to Eq. 4, repeated below.

$$\dot{Q}_{ventilated} = k \dot{Q}_{tunnel} \tag{4}$$

Carvel proposed sets of k value curves for both the growth phase and for fully involved tunnel fires. It was deemed that the curves for the growth phase were inappropriate as they are inherently temporally dependent, while the model assumes a steady state. A value taken from a growth phase curve represents how much the forced ventilation has increased the growth rate up to a given time in a fire, while a value taken from a fully involved curve represents how much the forced ventilation increases the HRR at its peak. For example, if a k value of 4 is taken from a growth phase curve, this implies that the HRR at a given time is four times larger than it would have been at the same time had there been no forced ventilation. Contrastingly, if a k value of 4 is taken from a fully involved curve, this implies that the HRR reached a peak value four times larger than it would have had there been no forced ventilation. On this basis, it was deemed more appropriate to use the fully involved curves.

The curves Carvel proposed were presented with several percentile distributions: 10%, 30%, 50%, expected, 70%, and 90%. For example, a value of 90% estimates an enhancement factor that will be equal to or greater than the actual enhancement

factor in 90% of fires. The user specifies which distribution to use. Different curves were presented for one-lane and two-lane tunnels. The user also specifies whether the tunnel being analysed is one-lane or two-lane. The k values were extracted from the curves presented by Carvel and Beard [19] using WebPlotDigitizer [42]. The tabulated k values can be found in Appendix C.

The decrease in HRR due to the operation of a WMS was modelled using Eq. (22), as follows:

$$\dot{Q}_{adjusted} = k_{WMS} \dot{Q}_{ventilated} \tag{22}$$

where $\dot{Q}_{adjusted}$ is the final adjusted HRR and k_{WMS} is how much the HRR is reduced due to WMS operation.

This simple model was chosen because the effect of a WMS on the HRR in a tunnel fire is not adequately understood. The actual relationship between WMS operation and HRR reduction may depend on many other factors, including the mass flow density of the WMS, droplet size, and HRR at the time of activation. It is claimed that the operation of a WMS can reduce the HRR to approximately 20-70% of its potential maximum HRR, according to the publicly available test reports [6][43]. The effect of this factor is studied extensively in the results (section 3).

The final result of these operations is $\dot{Q}_{adjusted}$, which is used for the rest of the calculations in a given iteration of the algorithm. Alternatively, the user can bypass this part of the algorithm and directly specify a value for $\dot{Q}_{adjusted}$.

2.2.3. Control Volume Analyses

The goal of this section is to present the calculation of the static pressure, temperature, density, and velocity in each CV and the forces acting on the flow.

In the inlet section, three assumed values and a form of Bernoulli's equation are used to determine the four key properties and the forces acting. The temperature and density are assumed to be equal to the ambient values. The temperature is assumed as equal to ambient because there is no external heat source or sink acting on the CV. The density is assumed as equal to ambient using the incompressible flow assumption, as the Mach number in this context will always be well below 0.3. The velocity in the inlet is initially assumed to be equal to the CVV, and is updated later in the algorithm as required. The mass flow rate of air in the tunnel is calculated at the inlet, as follows:

$$\dot{m}_{air} = \rho_{inlet} A_{tunnel} V_{inlet} = \rho_{\infty} A_{tunnel} V_{inlet}$$
(23)

where \dot{m}_{air} is the mass flow rate of air, ρ_{inlet} is the density of air at the inlet, A_{tunnel} is the cross sectional area of the tunnel, V_{inlet} is the inlet velocity, and ρ_{∞} is the ambient air density.

A form of Bernoulli's equation is then used in order to calculate the static pressure at the inlet portal, as follows:

$$p_{inlet,portal} = p_{\infty} - \frac{\rho_{inlet} V_{inlet}^2}{2}$$
(24)

where $p_{inlet,portal}$ is the pressure at the inlet portal, and p_{∞} is the ambient pressure.

This equation gives the pressure at the inlet portal. However, it does not account for pressure losses along the length of the inlet section. Hence the pressure in the inlet CV is calculated as follows:

$$p_{inlet} = p_{inlet,portal} + \Delta p_{losses,inlet} \tag{25}$$

where

$$\Delta p_{losses,inlet} = \frac{F1_{inlet} + F2_{inlet} + F5}{A_{tunnel}}$$
(26)

$$F1_{inlet} = \frac{-\rho_{inlet}}{2} C_{d,car} V_{inlet}^2 A_{car} N_{cars,inlet}$$
⁽²⁷⁾

$$N_{cars,inlet} = L_{inlet} K_{traffic}$$
⁽²⁸⁾

(- - ****

$$F2_{inlet} = -f \frac{\rho_{inlet}}{2} \frac{L_{inlet}}{D_h} A_{tunnel} V_{inlet}^2$$
⁽²⁹⁾

$$F5 = -K_{en} \frac{\rho_{inlet}}{2} A_{tunnel} V_{inlet}^2$$
(30)

where $N_{cars,inlet}$ is the number of cars in the inlet section, $K_{traffic}$ is the car density, and all other parameters are as introduced in section 1.4.3.

This method results in the calculated CV pressure being slightly too low. It essentially calculates the pressure at the end of the CV then assumes that is the value throughout the CV. It would be more accurate to assume the pressure in the CV is equal to the average of the start and end values. However, calculating the average value becomes difficult in the later CVs and has negligible impact on the solution obtained. As such, this assumption was made in all of the CVs considered.

The code used to perform the inlet calculations can be found in the MATLAB script file "InletCV" (Appendix A7).

In the upstream section, it is again assumed that the density is equal to ambient density, as the flow velocity in this context will always result in a Mach number well below 0.3. The temperature is again assumed to be equal to the ambient temperature, as there is no external heat source or sink. The inlet and upstream densities are both equal to the ambient density, and by extension the upstream velocity is equal to the inlet velocity due to the mass balance requirement. The pressure in this CV was calculated as follows:

$$p_{upstream} = p_{inlet} + \Delta p_{fans} + \Delta p_{losses, upstream}$$
(31)

where

$$\Delta p_{fans} = \frac{F4}{A_{tunnel}} \tag{32}$$

$$\Delta p_{losses,upstream} = \frac{F1_{upstream} + F2_{upstream}}{A_{tunnel}}$$
(33)

where $F1_{upstream}$ and $F2_{upstream}$ are calculated the same way as in the inlet CV, and F4 is calculated as follows:

$$F4 = N_{fans}\rho_{upstream}A_{fan}V_{fan}(V_{fan} - V_{upstream})K_j$$
(34)

The code used to perform the above calculations can be found in the MATLAB script file "UpstreamCV" (Appendix A8).

In the fire zone section, the mass flow rates of pyrolysis gases is calculated as follows:

$$\dot{m}_{pyrolysisGases} = \frac{\dot{Q}_{adjusted}}{\Delta H_{c,eff}}$$
(35)

where $\Delta H_{c,eff}$ is the effective heat of combustion of the fuel in the tunnel.

The mass flow rates of water, steam and water that does not evaporate are calculated as follows:

$$\dot{m}_{water} = \dot{m}_{WMS}^{\prime\prime\prime} N_{zones} L_{zone} W_{tunnel} H_{tunnel}$$
(36)

$$\dot{m}_{steam} = \dot{m}_{steam, radiation} + \dot{m}_{steam, convection}$$
(37)

$$\dot{m}_{water,lost} = (1 - x_{evap})\dot{m}_{remaining}$$
(38)

where

$$\dot{m}_{steam,radiation} = \frac{\chi_{rad} \dot{Q}_{adjusted}}{h_{steam,373} - h_{water,\infty}}$$
(39)

$$\dot{m}_{steam,convection} = x_{evap} \dot{m}_{remaining} \tag{40}$$

$$\dot{m}_{remaining} = \dot{m}_{water} - \dot{m}_{steam, radiation} \tag{41}$$

where $\dot{m}_{WMS}^{\prime\prime\prime}$ is the WMS water flux density, N_{zones} is the number of WMS zones, L_{zone} is the length of one WMS zone, $\dot{m}_{steam,radiation}$ is the mass flow rate of steam created by the evaporation of water due to radiation, χ_{rad} is the radiative fraction (taken as 0.35 [12]), $h_{steam,373}$ is the specific enthalpy of steam at 373 K, $h_{water,\infty}$ is the specific enthalpy of ambient water, $\dot{m}_{remaining}$ is the remaining amount of water, $\dot{m}_{steam,convection}$ is the mass flow rate of steam created by the evaporation of water due to convection, x_{evap} is the proportion of remaining water evaporated due to convection, and $\dot{m}_{water,lost}$ is the mass flow rate of water that does not evaporate and falls to the ground.

It is assumed that all of the radiation energy released by the fire is used to evaporate water. This assumption is made for several reasons. Firstly, water mist has been experimentally observed to strongly attenuate and absorb radiation [35]. Secondly, making this assumption results in higher downstream forces, which is a conservative result in the context of this work. Thirdly, it allows the heat losses model to meet its required boundary condition. The downstream forces acting on the flow are dependent on the value of the term $\rho_{downstream}V_{downstream}^2$. By increasing the mass of steam in the air, the value of this term also increases. Increasing the mass of steam in the air also increases the specific ideal gas constant of the mixture. This lowers the density, which causes an increased velocity and hence increases the value of the $\rho_{downstream}V_{downstream}^2$ term. The heat losses model used has a boundary condition requiring that the heat transfer by convection to the tunnel surface is equal to the heat transfer by conduction at the surface. For this boundary condition to be valid, the heat transfer by radiation must be negligible. Assuming that all of the radiant heat was attenuated by water mist allows this boundary condition to be met. The presence of water in smoke reduces the emission and transmission of radiation, thus reducing the radiant heat transfer between smoke and the tunnel walls [35].

The amount of water evaporated due to the convective heat release is assumed to depend on a user specified value for x_{evap} . This is a crude assumption, the implications of which are discussed further in section 2.3.1. This value is set to zero by default. It is assumed that the temperature of liquid water suspended in the flow is equal to the temperature in the fire zone. This can lead to a problem for fire zone temperatures in excess of 100 °C, as it creates a large energy barrier. Setting x_{evap} to zero essentially means that the system has to provide enough energy to evaporate all of the water if the temperature is to exceed 100 °C. An attempt was made in section 2.3.1 to calibrate the value of x_{evap} using experimental results.

The mass flow rate of gas leaving the fire zone CV is then calculated as follows:

$$\dot{m}_{firezone} = \dot{m}_{upstream} + \dot{m}_{fuel} + \dot{m}_{steam} \tag{42}$$

An energy balance is solved iteratively in order to determine the four key properties and the forces acting on the CV. The energy balance solved was based on the following steady state form described by Moran and Shapiro [44]:

$$\dot{E}_{CV} = 0 = \dot{Q}_{CV} - \dot{W}_{CV} + \dot{m}_{in} \left(h_{in} + \frac{V_{in}^2}{2} + g z_{in} \right) - \dot{m}_{out} \left(h_{out} + \frac{V_{out}^2}{2} + g z_{out} \right)$$
(43)

where \dot{Q}_{CV} accounts for heat transferred across the CV boundary, \dot{W}_{CV} accounts for work transferred across the CV boundary, \dot{m} is the mass flow rate across the CV boundary, h is the specific enthalpy, V is the velocity, and z is the average elevation. The general energy balance equation above was altered into the following form in order to analyse the fire zone CV:

$$\dot{E}_{CV} = 0 = \dot{Q}_{CV} - \dot{W}_{CV} + \dot{m}_{upstream} \left(h_{upstream} + \frac{V_{upstream}^2}{2} + g z_{upstream} \right) + \dot{m}_{water} \left(h_{water} + \frac{V_{water}^2}{2} + g z_{water} \right) - \dot{m}_{firezone} \left(h_{firezone} + \frac{V_{firezone}^2}{2} + g z_{firezone} \right) - \dot{m}_{water,lost} \left(h_{water,lost} + \frac{V_{firezone}^2}{2} + g z_{water,lost} \right)$$
(44)

The value of \dot{Q}_{CV} is calculated as follows:

$$\dot{Q}_{CV} = \dot{Q}_{adjusted} - \dot{Q}_{losses} \tag{45}$$

where $\dot{Q}_{adjusted}$ is the value calculated earlier in the algorithm and \dot{Q}_{losses} is calculated as follows:

$$\dot{Q}_{losses} = hA_{surface,CV} \left(T_{firezone} - T_{surface} \right) \tag{46}$$

$$h = \frac{Nu \cdot k_{air} (T_{firezone})}{D_h} \tag{47}$$

$$Nu = 0.125f \cdot Re \cdot Pr_{air}^{\frac{1}{3}}(T_{firezone}) \qquad Re > 10,000$$

$$Nu = 3 \qquad Re \le 10,000$$
(48)

$$Re = \frac{V_{firezone}D_h}{v_{air}(T_{firezone})}$$
(49)

where *h* is the convective heat transfer coefficient of the flow, *Nu* is the flow Nussult number, and k_{air} , Pr_{air} , and v_{air} are the thermal conductivity, Prandtl number, and kinematic viscosity of air respectively. The values for k_{air} , Pr_{air} , and v_{air} are functions of the fire zone temperature, taken from tabulated data by [45] that can be found in Appendix D. The equations for *Nu* are taken from [46] for internal forced convection. The variable $A_{surface,CV}$ is the surface area of the CV. The value of $T_{surface}$ is calculated using a model suggested by Drysdale [47] for semi-infinite masses with the following boundary condition: the heat transferred to the surface by convection is equal to the heat transferred through the surface by conduction, after starting at ambient conditions. The equation used is as follows:

$$T_{surface} = T_{\infty} + \left(T_{firezone} - T_{\infty}\right) \left[1 - \exp\left(\frac{\alpha t}{(k/h)^2}\right) \cdot \operatorname{erfc}\left(\frac{(\alpha t)^{0.5}}{k/h}\right)\right]$$
(50)

where α and k are the thermal diffusivity and thermal conductivity of the material conducting heat respectively. The values of α and k are taken as $5 \times 10^{-7} \frac{m^2}{s}$ and $1 \frac{W}{mK}$ respectively, assuming the tunnel walls are made of concrete and using data from Drysdale [47]. The value of t was increased from zero in arbitrary steps of 60 s, until the surface temperature was changing by less than an arbitrary amount of 1% per iteration. These conditions were deemed to be close enough to steady state. The code used to calculate \dot{Q}_{losses} is called "heat_losses" and can be found in Appendix 10.

The value of \dot{W}_{CV} is calculated as follows:

$$\dot{W}_{CV} = V_{firezone} \left(F \mathbf{1}_{firezone} + F \mathbf{2}_{firezone} + F \mathbf{6} \right) \tag{51}$$

The values of $h_{upstream}$ and $V_{upstream}$ were calculated previously. The value of $z_{upstream}$ is equal to half the tunnel height, as this is the average elevation. The value of h_{water} is taken from tabulated data by the National Institute of Standards and Technology (NIST) [48] and is assumed to be at ambient conditions of 20 °C and 101,325 Pa. The value of V_{water} is based on properties of the WMS, and is calculated as follows:

$$V_{water} = \frac{K_{nozzle} p_{water}}{\rho_{water} A_{nozzle}}$$
(52)

The value of z_{water} is equal to the height of the tunnel as it is assumed that the WMS is located at ceiling level. The value of $h_{firezone}$ is calculated as follows:

$$h_{firezone} = h_{firezone,air} \frac{\dot{m}_{air} + \dot{m}_{pyrolysisGases}}{\dot{m}_{downstream}} + h_{firezone,steam} \frac{\dot{m}_{steam}}{\dot{m}_{downstream}}$$
(53)

The values of $h_{firezone,air}$ and $h_{firezone,steam}$ are found using functions called "spec_ent_air" and "spec_ent_steam", which look up tabulated values for the specific enthalpy based on an input temperature. Where required these functions will linearly interpolate to find more accurate values. The temperature of the air is equal to the fire zone temperature. The temperature of the steam is taken as the larger of the fire zone temperature or 100 °C. The codes can be found in Appendix 11 and Appendix 12. The specific enthalpy values of air and steam are taken from Moran and Shapiro [44] and NIST [48] respectively and can be found in Appendix E and F. The mass flow rate of pyrolysis gases is included with the air term, as it is difficult to estimate the specific enthalpy of pyrolysis gases. The value of $V_{firezone}$ is calculated as follows:

$$V_{firezone} = \frac{\dot{m}_{firezone}}{\rho_{firezone} A_{tunnel}}$$
(54)

where $\rho_{firezone}$ is calculated using the ideal gas law, as follows:

$$\rho_{firezone} = \frac{p_{firezone}}{R_{mix}T_{firezone}}$$
(55)

The ideal gas constant of the mixture is calculated based on the mass fractions of steam and air in the mixture. The influence of pyrolysis gases on this value is ignored, as it is difficult to estimate the molar mass and the mass fraction is low. The value of R_{mix} is calculated as follows:

$$R_{mix} = R_{air} \frac{m_{air}}{m_{air} + m_{steam}} + R_{steam} \frac{m_{steam}}{m_{air} + m_{steam}}$$
(56)

The value of $p_{firezone}$ is calculated as follows:

$$p_{firezone} = p_{upstream} + \Delta p_{firezone} \tag{57}$$

$$\Delta p_{firezone} = \frac{F1_{firezone} + F2_{firezone} + F6}{A_{tunnel}}$$
(58)

where F6 is the force due to the throttling effect of fire.

The value of $F1_{firezone}$ is calculated in the same way as for previous CVs, except that values for $C_{d,HGV}$ and A_{HGV} are used instead of the corresponding values for cars, and the value of N_{HGV} is set to one. The value of $F2_{firezone}$ is calculated in the same way as for previous CVs. The value of F6 is calculated as follows:

$$F6 = F6_{air} + F6_{steam} + F6_{water,lost}$$
⁽⁵⁹⁾

$$F6_{air} = \dot{m}_{upstream} \left(V_{upstream} - V_{firezone} \right) \tag{60}$$

$$F6_{steam} = \dot{m}_{steam} \left(0 - V_{firezone} \right) \tag{61}$$

$$F6_{water,lost} = \dot{m}_{water,lost} \left(0 - V_{firezone} \right)$$
(62)

The throttling model proposed by Hwang and Chaiken [24] was used rather than that proposed by Dutrieue and Jacques [26]. This decision was made because the model by [26] does not account for the gas cooling effect of the WMS or the effect of the added water mass; it is only dependent on the HRR, hydraulic diameter, and upstream ventilation velocity. In contrast, the model by [24] does depend on both the mass flow rate and the gas temperature. Adding water increases the mass flow rate and hence increases the throttling effect. Decreasing the temperature results in a reduced throttling effect.

The value of $z_{firezone}$ is taken as half the tunnel height. The value of $h_{water,lost}$ is taken from tabulated data by NIST [48], assuming the temperature is the lower of the fire zone temperature or 100 °C. It is assumed that $V_{water,lost}$ is equal to the velocity in the rest of the CV before it hits the ground. The value of $z_{water,lost}$ is zero.

The energy balance is then considered as follows:

$$\dot{E}_{cV} = 0 = \dot{Q}_{adjusted} - \dot{Q}_{losses} + \dot{m}_{upstream} \left(h_{upstream} + \frac{V_{upstream}^2}{2} + g z_{upstream} \right) + \dot{m}_{water} \left(h_{water} + \frac{V_{water}^2}{2} + g z_{water} \right) - \dot{m}_{firezone} \left(h_{firezone} + \frac{V_{firezone}^2}{2} + g z_{firezone} \right) - \dot{m}_{water,lost} \left(h_{water,lost} + \frac{V_{firezone}^2}{2} \right)$$
(63)

Of these terms, only \dot{Q}_{losses} , $h_{firezone}$, $V_{firezone}$, and $h_{water,lost}$ are dependent on the four key variables in the fire zone CV; the rest of the variables are constant. As such the energy balance can then be considered in the following form:

$$-\dot{Q}_{losses} + \dot{m}_{firezone} \left(h_{firezone} + \frac{V_{firezone}^2}{2} \right) - \dot{m}_{water,lost} \left(h_{water,lost} + \frac{V_{firezone}^2}{2} \right) = C \quad (64)$$

The temperature is then varied and the calculation process iterated until the energy balance is satisfied. At this point, converged values for the four key properties and the forces acting on the CV are obtained. The code used to perform these calculations is called "FirezoneCV" and can be found in Appendix A9.

The analysis performed in the downstream CVs is very similar to that in the fire zone. Again the elevation terms are cancelled out, and the mass flow rates are equal. Using these assumptions, the energy balance in the first CV takes the following form:

$$\dot{E}_{CV,n} = 0 = \dot{Q}_{CV,n} - \dot{W}_{CV} + \dot{m} \left(h_{CV,n-1} + \frac{V_{CV,n-1}^2}{2} - h_{CV,n} - \frac{V_{CV,n}^2}{2} \right)$$
(65)

The value of \dot{Q}_{CV} is simply equal to the value of \dot{Q}_{losses} in the CV under consideration, which is calculated the same way as in the fire zone. The subscript n denotes which CV is being considered. In the first downstream CV the value of n is 1 and the subscript n - 1 refers to the fire zone CV. Again the energy balance equation is solved by varying the temperature and iterating the calculation process until convergence is achieved. The code used to perform these calculations is called "DownstreamCVs" and can be found in Appendix A13.

2.2.4. Force Balance

In this section all of the forces considered are summed, in order to determine if there is a shortfall or an excess of thrust from the jet fans. If there is a shortfall, another fan is added. If there is an excess, the inlet velocity is increased until the forces are balanced. An arbitrary value of ± 1 N is used as the threshold for acceptable convergence. The code used to perform the calculations in this section is a MATLAB script file called "ForceBalance" and can be found in Appendix A14.

2.3. Model Validation

2.3.1 Overview

The model is made up of several components that are combined with the aim of simulating complex behaviour. Due to the specific nature of the problem and the high cost of full-scale tunnel fire tests, there is no experimental data that is fully applicable for validating the model. As such, the model's reliability is validated by comparing applicable data with the results obtained using different components of the model. The model was validated using experimental data for the downstream temperatures in tunnel fires with suppression, and simulation results for the number of fans required to prevent backlayering in an unsuppressed tunnel.

2.3.2. Downstream Temperatures

The model was validated for the case when there is a FFFS in the tunnel. Results from the model were compared to those from two full-scale experiments conducted by Marioff [6]. In the first experiment, a stable HRR of approximately 20 MW was measured while the WMS was operating (Fig. 4).



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Fig. 4. Heat release rate during 'standard severity' full-scale experiment by Marioff [6].
The temperatures 25-30 m downstream of the fire can be seen in Fig. 5. The green line is the temperature measured 1.5 m above the ground, the blue at 3 m, and the red at ceiling level. It can be seen that the temperature is approximately 45-60 °C throughout the cross section. This shows that the 1D model assumption is appropriate when a WMS limits the HRR to a steady state; it is reasonable to assume constant properties throughout the height of the cross section in this case.



Fig. 5. Temperatures at different heights during 'standard severity' full-scale experiment conducted by Marioff [6].

The tunnel used for these experiments was 5.2 m high, 9.5 m wide, and 600 m long. The HRR was assumed to be a constant 20 MW for validating the model. The radiative fraction was set to 0.35. The forced ventilation velocity was 2-3 m/s. A value of 2.5 m/s was used in the model. The location of the fire was not specified, so it was assumed to be in the middle of the tunnel. The specifications of the WMS used in these experiments were not explicitly reported. However, these experiments were used as the basis for specifying the WMS used in the Madrid M30 road tunnel. The WMS installed in the M30 tunnel is designed to provide an average water flux density of 0.533 1/min/m³ over three zones with length of 24 m each [8]. Based on this, it was assumed that the WMS used in testing had the same average water flux density, zone length, and number of zones.

The model was slightly modified in order to accommodate these input values. The script file "HeatReleaseRateAdjustment" was deactivated; a constant HRR of 20 MW was used instead. The inlet velocity was set to a constant 2.5 m/s; it was not altered in the script "ForceBalance". Using these inputs, the model gave a temperature of 64 °C in a CV 25-30 m downstream of the fire. This is close to the range measured experimentally. One reason for the slight overprediction is the assumption that x_{evap} is equal to zero. If a value of 5% is assumed, the temperature predicted by the model drops to 52 °C.

In the second experiment, the measured HRR is shown in Fig. 6. It can be seen that the WMS did not limit the HRR to a steady state in this case.



Fig. 6. Heat release rate during 'high severity' full-scale experiment conducted by Marioff [6].

The experimental temperatures 25-30 m downstream of the fire are shown in Fig. 7. Again the green, blue, and red lines are measured temperatures at respective heights of 1.5 m, 3 m, and ceiling height.



Fig. 7. Temperatures at different heights during 'high severity' full-scale experiment conducted by Marioff [6].

It can be seen that the temperature is not constant over the height of the tunnel. This suggests that the 1D model assumption is flawed when the WMS does not control the HRR of the fire. However, this data can still be of some use for comparison to the results of the model.

This was done by assuming a steady state HRR of 65 MW, approximating the period from 12-18 min in Fig. 6. The temperatures at the ceiling, 3 m, and 1.5 m were estimated using the peak values in Fig. 7. These values occur at approximately 18 min, corresponding to the time when the HRR began to decrease. These were taken as 200 °C, 175 °C, and 70°C respectively. The temperature at the ground level was assumed to be 50 °C. Based on these assumptions, the temperature profile in the tunnel was assumed to have the following form (Fig. 8).



Fig. 8. Assumed peak temperature profile during 'high severity' full-scale experiment conducted by Marioff [6].

The model provides a single estimated temperature, and as such cannot be directly compared to the temperature profile presented above. A method was devised to compare the model's results with this temperature profile.

An average experimental temperature was estimated by calculating the area beneath the temperature curve in Fig. 8 and dividing it by the height of the tunnel. This gave an estimated average temperature of 132 °C. The calculation is presented below:

$$T_{air} = \frac{1}{H} \int_0^H T_{air} \, dH = 132 \, {}^{\text{o}}C$$

The model estimated a temperature slightly below 100 °C at this location. This is substantially lower than the estimated experimental temperature calculated above. This is a limitation of the model; it has a problem estimating downstream temperatures in excess of 100 °C. This is because of an assumption previously made. It was assumed that any water not evaporated by radiation has the same temperature as the air in the fire zone. This creates a large energy barrier, where the system has to provide enough heat to evaporate all of the remaining water if the temperature is to

exceed 100 °C. If it is assumed that the remaining water reaches a temperature of 100 °C but none evaporates, the model estimates a temperature of 176 °C at this location. This is substantially higher than the experimental result. This suggests that the model can bound the solution, which is a useful result. It was found that 18% of the remaining water would have to evaporate for the model to give a temperature of 132 °C. With more experimental data, the value of x_{evap} could be calibrated to more accurately predict downstream temperatures. Based on these results, it was concluded that the model is valid for predicting gas temperatures up to 100 °C with the evaporation proportion set to 0%. The general trends predicted by the model were also deemed to have been validated, although caution should be exercised when considering temperatures above 100 °C.

2.3.3. Backlayer Prevention

The final result of the model is to calculate how many fans are required to prevent backlayering for a given set of input values. Consequently the model's ability to accurately do this must be verified. The most convenient way to do this is by comparing the results of this model with those predicted by other models. The work of Carvel et al. [23] presented the number of fans required to prevent backlayering under different circumstances. The results of Carvel et al. [23] will be used to verify that this model can provide accurate results.

Carvel et al. [23] used a tunnel 8 m high, 6.5 m wide, and 100 m long for their simulations. There was no suppression by a WMS or otherwise. The centre of the fire was positioned 32 m from the upstream portal and 68 m from the downstream portal. Seven fans were used. The fans were modelled as 0.5 x 0.5 m squares with zero thickness, producing an outlet velocity of 35 m/s. The upstream ventilation velocity was set to the critical value of 3.4 m/s for simulations where the HRR was greater than the critical dimensionless HRR of 0.15, which in this case is all fires with a HRR of 30 MW or more. The HRR was varied from 10-90 MW in steps of 10 MW. The fire source was generally 3 m wide x 4 m long x 1 m high. Simulations were conducted with the fans located at 25 m and 50 m from the fire, and with varying fire areas. Table 1 compares the results of the model with those found by Carvel et al. for fires with HRRs 30 MW or more. Simulation 1 shows the results with the fans located 25 m from the fire, and standard fire area. Simulation 2 shows the results with the fans 50 m from the fire and using the standard fire area. Simulation 3 shows the

results with the fans 50 m from the fire and varying the fire area. Where the table says N/A, it is because the seven fans used by Carvel et al. were not sufficient to prevent backlayering. The model has the ability to predict how many fans are required in these more extreme cases.

HRR (MW)	Model	Simulation 1	Simulation 2	Simulation 3
30	4	3	3	3
40	4	4	4	4
50	5	4	5	4
60	5	5	5	5
70	6	5	5	5
80	6	N/A	6	6
90	6	N/A	7	N/A

Table 1. Number of fans required to prevent backlayering. Results of the modeland FDS simulations by Carvel et al. [23]

It can be seen that the model provides similar results to those obtained by Carvel et al. [23]. Based on this, it was deemed that the model provides a reasonable level of accuracy when predicting the number of fans required to prevent backlayering.

3. RESULTS AND DISCUSSION

3.1. Overview

A case study was performed on a 6 m wide x 6 m high x 600 m long tunnel in keeping with the work done by Looi [9]. Initially, the tunnel was considered without a fire and without WMS operation. This was to demonstrate the effect of aerodynamic resistance, friction resistance, portal pressure difference and inlet flow separation resistance. The tunnel was then considered for a range of unsuppressed fires. This was to demonstrate the throttling effect and the impact of increased downstream temperatures on the other forces. The tunnel was then considered with WMS operation. The influence of the WMS is analysed and discussed.

3.2. Case Study

3.2.1. Ambient Conditions

The model was used to analyse the tunnel without any fire in the tunnel and without a WMS in operation. This was done to provide baseline values of the different forces acting in the tunnel, for comparison with the corresponding values in the cases with fire in the tunnel.

The relevant inputs for this simulation are summarised in Table 2. The number of lanes was set to two in order to determine the number of cars in the tunnel. Drag coefficients for an HGV and an average car were taken based on values from Colella [49]. According to Colella, traffic density during morning rush hour ranges from 8-23 vehicles/lane/km [49]. Based on this, a value of 15 vehicles/lane/km was chosen. The fan diameter and outlet velocity used were specified by Pollrich [50], and are the same as those used by Looi [9]. Jang and Chen specified appropriate values of 0.9 and 0.6 for the pressure rise coefficient and the entry loss coefficient respectively [28].

Parameter	Value (unit)	Ref.
Tunnel height	6 (m)	-
Tunnel width	6 (m)	-
Tunnel length	600 (m)	-
Number of lanes	2 (-)	-
HGV drag coefficient	0.4 (-)	[49]
HGV height	3 (m)	-
HGV width	3 (m)	-
Car drag coefficient	0.3 (-)	[49]
Car height	1.5 (m)	-
Car width	2 (m)	-
Traffic density	15 (vehicles/lane/km)	[49]
Friction factor	0.02 (-)	[28]
Fan diameter	0.84 (m)	[50]
Fan outlet velocity	26.2 (m/s)	[50]
Pressure rise coefficient	0.9 (-)	[28]
Entry loss coefficient	0.6 (-)	[28]

Table 2. Input parameters for ambient simulation.

It was found that three fans were needed to induce a forced ventilation flow with velocity larger than the CVV. The size of each force modelled is shown in Table 3.

Cause of force	Force (N)
Aerodynamic resistance of cars (F1)	-140
Wall friction (F2)	-529
Portal pressure difference (F3)	-254
Jet fan thrust (F4)	1077
Inlet flow separation (F5)	-153
Throttling effect (F6)	0

Table 3. Simulated forces in ambient conditions.

Table 3 shows that wall friction was the dominant mode of flow resistance in the tunnel. The aerodynamic resistance of cars, portal pressure difference, and inlet flow separation all caused sizeable drag forces that the fans needed to overcome. As expected, the throttling force was zero.

3.2.2. Unsuppressed Fires

The model was used to analyse the tunnel with a range of HRRs. The script file "HeatReleaseRateAdjustment" was bypassed, and HRRs ranging from 2-100 MW in steps of 2 MW were directly specified. Other inputs are shown in Table 4. The other inputs were the same as those specified in Table 2.

Parameter	Value (unit)	Ref.
Inlet CV length	100 (m)	-
Upstream CV length	185 (m)	-
Fire zone CV length	30 (m)	-
Total downstream length	285 (m)	-
Number of downstream CVs	30	-
Radiative fraction	0.35 (-)	[43]

Table 4. Inputs parameters for unsuppressed fire simulations.

The number of fans required to prevent backlayering was plotted as a function of the HRR (Fig. 9).



Fig. 9. Number of fans required to prevent backlayering as a function of HRR.

It can be seen that increased fire size results in a larger number of fans to prevent backlayering. This makes sense, as F1, F2 and F6 all increase with increased downstream temperatures. F1 and F2 increase because they are both of the form $F = A + B\rho_{downstream}V_{downstream}^2$, and increased downstream temperatures increase the value of the term $\rho_{downstream}V_{downstream}^2$. This is because of the ideal gas law and the mass balance requirement. The pressure in the tunnel is approximately equal to ambient pressure, so an increased temperature results in a decreased density. The mass flow rate of air is equal to $A_{tunnel}\rho_{downstream}V_{downstream}$ and is constant, so a decreased density increases the downstream velocity, the $\rho_{downstream}V_{downstream}^2$ term and the values of F1 and F2. The sum of F1 and F2 at each HRR considered can be seen in Fig. 10.



Fig. 10. Sum of aerodynamic car drag and wall friction, as a function of HRR.

F6 is dependent on the mass flow rate and the change in velocity across the fire zone CV. The mass flow rate is essentially constant for all of the HRRs considered. Larger fires result in larger mass flow rates due to the production of additional pyrolysis gases, however this additional mass is negligible to the throttling force predicted. Larger fires result in larger throttling forces because they increase the downstream temperatures. As described above, this leads to a decreased density, an increased velocity, and hence an increased throttling force.

The forces for selected HRRs are shown in Table 5.

	Force (N)		
Cause of force	0 MW	50 MW	100 MW
Aerodynamic resistance of cars (F1)	-140	-201	-312
Wall friction (F2)	-529	-713	-1080
Portal pressure difference (F3)	-254	-236	-277
Jet fan thrust (F4)	1077	1805	2854
Inlet flow separation (F5)	-153	-142	-166
Throttling effect (F6)	0	-506	-1028

Table 5. Simulated forces for different HRRs.

It can generally be seen that as the HRR increases, the magnitude of each force increases. There are some exceptions to this trend. It can be seen that the values of F3 and F5 for the 50 MW simulation are lower than those for the 0 MW simulation. This is due to the discrete number of fans. The model finds the minimum number of fans required to prevent backlayering, but this results in a surplus of thrust. The model increases the inlet velocity until increased resistance balances the excess thrust. The decrease observed in F3 and F5 between the 0 MW and 50 MW simulations is because the 0 MW simulation had a larger excess of thrust. This resulted in a new inlet velocity higher than that calculated for the 50 MW simulation. This effect can be observed in Fig. 11.



Fig. 11. Force due to inlet flow separation as a function of the HRR.

It can be seen that the magnitude of F5 decreases steadily until another fan is added, at which point there is a sudden increase in magnitude. The peak values at the top of the graph are all very similar; this is because they occur when the force balance is approaching zero. The reason for the small differences observed in these values is the convergence criterion set. The solution convergence was deemed acceptable for an absolute force balance of less than 10 N. The peaks at the bottom of the graph are decreasing in magnitude; this is because the solution becomes more sensitive to velocity as the HRR increases. This is again because the downstream forces depend on the $\rho_{downstream}V_{downstream}^2$ term. The velocity is increased because the density is decreased; the density is decreased because the temperature is increased at an essentially constant pressure. The temperature increases approximately linearly with increased HRR; this is because the specific heat capacity of air is essentially constant over a wide range of temperatures. This linearly increase in temperature with increasing HRR is shown by Fig. 12, which shows the temperature calculated by the model at the middle of the downstream section.



Fig. 12. Temperature at the middle of the downstream section for different HRRs.

The small drops in temperature at HRRs of approximately 30 MW, 50 MW, 70 MW, and 100 MW are due to the addition of a fan at these points. This results in an increased inlet velocity, and hence an increased mass flow rate of air. A higher mass flow rate of air with the same amount of energy results in lower temperatures. The results presented in this section appear to demonstrate that higher downstream temperatures require more fans to prevent backlayering. This trend is clearly demonstrated in Fig. 13.



Fig. 13. Number of fans required to prevent backlayering, as a function of temperature at the centre of the downstream section.

Fig. 13 shows that the number of fans increases as the downstream temperature increases. It also shows the downstream temperature decreasing immediately after a fan is added. For example, the temperature increases from approximately 380 °C to 520 °C while the number of fans is set to 7, but decreases to about 500 °C as soon as another fan is added. The main reason for this is again the increased mass flow rate of air reducing the average gas temperature. It can be concluded from the results presented in this section that the downstream temperature has a large influence on the predicted number of jet fans necessary to prevent backlayering.

Using the model for unsuppressed fires has limitations that must be taken into account. The equations used to model heat losses through the tunnel walls are not suited to unsuppressed fires, as they assume the radiation heat transfer to the tunnel walls is negligible. This is not accurate for unsuppressed tunnel fires; a large amount of radiant heat will be transferred to the walls. Realistically, this would result in lower temperatures downstream than were calculated using this model. Higher temperatures result in higher resisting forces, which suggests that the model may over predict the number of fans required to prevent backlayering. The use of a onedimensional model also has its limitations when considering unsuppressed fires. Due to the buoyancy of hot gases, the temperature at the top of the tunnel is likely to be much higher than at the bottom; the tunnel will probably have a high level of smoke stratification in the case of an unsuppressed fire. This is a problem for the model, as it assumes that the temperature is the same over the entire height of the tunnel at a given longitudinal location. At large distances from the fire, the stratification may be limited and the model may become more applicable. Despite these limitations, the model still identifies the trends likely to occur. These trends are useful as long as the limitations of the model are understood and taken into account.

3.2.3. Suppressed Fires

The model was then used to analyse the tunnel with a range of fires and a WMS operating. The script file "HeatReleaseRateAdjustment" was partially bypassed for the analysis. The enhancement of HRR due to enclosure effects and forced ventilation were both bypassed. The HRR reducing effect of the WMS was considered. The value of k_{WMS} was varied in steps of 25% from 20-70% in line with the values reported in literature. This allowed comparison for the same input HRRs. The parameters specified in Table 3 and Table 4 were again used in this analysis. In addition to these parameters, the length of a WMS zone was set to 30 m, three zones were used, and the water flux density was varied from 0.3-0.7 1/min/m³ (1.8-4.2 1/min/m²) in steps of 0.2 1/min/m³ (0.6 1/min/m²). The input parameters are summarised in Table 6.

Parameter	Value (unit)	Ref.
HRR reduction factor	20-70% (-)	[6][43]
Water flux density	$0.3-0.7 (1/\min/m^3)$	-
WMS zone length	30 (m)	-
Number of WMS zones	3	_

 Table 6. Input parameters for suppressed fire simulations.

Results are presented for cases where the downstream gas temperatures are below 100 °C, as the validation section showed that the model has limited reliability beyond

this temperature. The proportion of remaining water evaporating was set to 0%, again due to the results of the validation section. The model performed well when the gas temperatures were below 100 °C and the proportion of water evaporating was 0%, so presenting the results of similar cases was deemed the most reasonable approach.

The estimated number of fans required to prevent backlayering can be seen in Fig. 14-19. The HRR on the x-axis is the 'unadjusted' HRR. This corresponds to the HRR used in section 3.2.2. This HRR is then multiplied by the appropriate k_{WMS} value in order to find the number of fans required to prevent backlayering at this 'unadjusted' HRR. The results were divided into six graphs for clarity. The lines were slightly vertically shifted so that they can be seen more easily. In each graph, the value of either k_{WMS} or the water flux density (denoted WFD) was kept constant while the other was varied.



Fig. 14. Number of fans required with an HRR reduction factor of 0.7.



Fig. 15. Number of fans required with an HRR reduction factor of 0.45.



Fig. 16. Number of fans required with an HRR reduction factor of 0.2.

Several observations can be made from looking at these graphs. Fig. 14-16 show that increasing the water flux density increases the number of fans required to prevent backlayering up to heat release rates of approximately 15 MW. In all three of the graphs, the lines representing WMS operation never increase beyond four fans, whereas the blue line representing an unsuppressed fire reaches five fans at approximately 29 MW. It was estimated that the blue line 'overtook' the other three lines at an average value of approximately 15 MW. The purple line is always at four fans, while the other lines only reach this level at increased HRRs. The purple line represents the highest water flow rate. The larger mass of water introduced into the tunnel was determined to be the reason that more fans were required. This increased mass resulted in an increased thrust requirement to accelerate it to the velocity of the flow in the fire zone. The throttling force in the model represented this behaviour. The throttling equation depends on the mass flow rate in the fire zone and the change in velocity of this mass flow rate. The average longitudinal velocity of the water was assumed to be zero upon entry into the tunnel, and the final velocity was assumed to be equal to the calculated fire zone flow velocity. This inertial resistance dominates the model at lower HRRs, before being surpassed by heat-induced resistances at higher HRRs.

At higher HRRs, the mechanisms previously described begin to dominate. Increased temperatures lead to decreased densities, increased velocities, and increased downstream forces. This is why the blue line increases to five fans while the others do not: the downstream temperatures in unsuppressed fires are larger than those in suppressed fires. If the HRR were to be increased further beyond the 30 MW shown in the graphs, the blue line would start to diverge further from the others due to these temperature-induced forces.



Fig. 17. Number of fans required with a WFD of 0.3 l/min/m³.



Fig. 18. Number of fans required with a WFD of 0.5 l/min/m³.



Fig. 19. Number of fans required with a WFD of 0.7 l/min/m³.

Fig. 17-19 show that decreasing the value of k_{WMS} reduces the number of fans required to prevent backlayering. The purple line represents the lowest value of k_{WMS} and is the last to increase to four fans, apart from in the case with a water flux density of 0.7 l/min/m³. In this case, the inertia of the high mass of water causes the line to sit at four fans even at low HRRs. The orange line represents the case with the highest k_{WMS} value, so it makes sense that this line is consistently the first one to reach four fans. The higher k_{WMS} value results in higher temperatures, so the temperature driven forces begin to dominate earlier for this simulation than for the purple line. Fig. 17 shows the purple line only reaching four fans at a HRR close to 30 MW. This also suggests that

4. CONCLUSIONS

A one-dimensional control volume model was implemented in MATLAB in order to analyse the relationship between the operation of a water mist system and the number of jet fans required to prevent smoke backlayering in a tunnel fire. It was found that a water mist system has a mixed effect on the number of fans required. At heat release rates lower than approximately 15 MW, it was found that the water mist system increased the number of fans required. This was deemed to be due to resistance generated in accelerating the relatively large mass of water to the speed of the fire zone flow. For heat release rates larger than 15 MW, it was found that a water mist system reduced the number of fans required. It was found that two main mechanisms were responsible for this reduced number of fans: heat release rate reduction and downstream gas cooling. The forces acting against the jet fans were found to be heavily influenced by the downstream temperatures generated by the fire. Hence when a water mist system is activated, the magnitudes of these forces are likely to be reduced.

The model could be improved in many ways. The relationship between water mist system operation and heat release rate is not well understood, full-scale experiments should be conducted in order to quantify this relationship. The proportion of water evaporated during water mist system operation is unknown, but full-scale data would be useful for estimating the gas temperatures downstream of a fire. The throttling effect of fire should also be investigated at full-scale. The interaction between water mist and longitudinal ventilation is also poorly understood at present, and may prove to be a major limitation of water mist systems if tested at full-scale. Unfortunately, it is very costly to carry out experiments at full-scale, so it may be some time before these key relationships are understood in more depth.

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APPENDICES

Appendix A: MATLAB Codes Appendix A1: MasterScript

% Master script

clear clc

 $\$ Call script to define variables and initialise problem initialiseProblem

```
while stopIterations == 0
```

HeatReleaseRateAdjustment InletCV UpstreamCV FirezoneCV DownstreamCVs ForceBalance

end

Appendix A2: InitialiseProblem

```
% Initialise problem
% Call script to define problem parameters
Parameters
% Calculate critical ventilation velocity for a large fire
% Assumes dimensionless HRR is above 0.43 --> maximum critical
velocity
Critical ventilation velocity;
% inital upstream velocity is specified as equal to critical
velocity
% some other variables are given initial values for first loop
% all of these values will change in the loop
V upstream = V crit;
V inlet = V crit;
rho inlet = rho ambient;
N fans = 1; % the while loop adds fans as necessary
% While loop will keep adding fans as long as stopAddingFans =
0
% This value is changed to 1 as soon as the force balance
becomes positive
stopAddingFans = 0;
% while loop will continue to iterate until convergence is
achieved
% at convergence this value becomes equal to 1
stopIterations = 0;
% Set initial forces for first iteration
F1 fire zone low = 0;
F2 fire zone low = 0;
F6 fire zone low = 0;
F1 fire zone high = 0;
F2 fire zone high = 0;
F6 fire zone high = 0;
F1 fire zone middle = 0;
F2 fire zone middle = 0;
F6 fire zone middle = 0;
% Create arrays to store downstream data
T downstream = zeros(N_CV + 1, 1);
rho downstream = zeros(N CV + 1, 1);
p downstream = zeros(N CV + 1, 1);
V downstream = zeros(N_CV + 1, 1);
F 1 downstream = zeros(N_CV + 1, 1);
F 2 downstream = zeros(N CV + 1, 1);
E_downstream = zeros(N_CV + 1, 1);
```

```
% Parameters
% specify the size of a fire in open ambient conditions
% this will be increased due to being in a tunnel
% will also be increased due to windspeed
% will also be decreased by the effect of water mist system
% see HeatReleaseRateAdjustment script
Q = 10 * 10^{6}; %(W)
H tunnel = 5.2; % (m) tunnel height
W tunnel = 9.5; % (m) tunnel width
A tunnel = H tunnel * W tunnel; % (m<sup>2</sup>) tunnel area
L tunnel = 600; % (m) tunnel length
N lanes = 2; % one or two lane tunnel
% Specify fire enhancement factor percentile
% The fire enhancement factor has percentile distributions
% For example, k 90% estimates an enhancement factor that will
be equal to
% or greater than the enhancement factor in 90% of fires
% If you want k 90%, select fire enhancement percentile = 6
% 1 = expected, 2 = 10%, 3 = 30%, 4 = 50%, 5 = 70%, 6 = 90%
fire enhancement percentile = 1;
L fire zone = 24; %(m) length of one WMS section
N zones = 3; % number of WMS zones
densityWMS = 0.5333; % (1/min.m<sup>3</sup>) density of water mist
system spray
% Assume water mist system limits HRR to a ratio of its
theoretical max
k WMS = 0.5;
% Proportion of remaining water evaporated
% Remaining water is that not evaporated by radiation
x evap = 0;
L inlet = 50; % (m) length of inlet section - distance from
inlet portal to fans
L upstream = 176; %(m) length of tunnel upstream of fire
location
L downstream = 250; %(m) length of tunnel downstream of fire
zone CV
N CV = 20; % number of control volumes the downstream section
is split into
rad fraction = 0.45; % Fraction of HRR converted into
radiation
conv fraction = 1 - rad fraction; % Fraction of HRR converted
```

```
into convection
```

```
% Assumed value for effective heat of combustion of fuel in
tunnel
H c eff fuel = 20 * 10^{6}; (J/kg)
%F1
% Drag coefficients of heavy goods vehicle and some average
car
C \, d \, HGV = 0.4;
C d car = 0.3;
% HGV dimensions
W HGV = 3; %(m)
H HGV = 3.5;
A_HGV = W_HGV * H_HGV; %(m^2)
% Car dimensions
W \text{ car} = 2; \%(m)
H car = 1.5;
A_car = W_car * H_car; %(m^2)
% Traffic conditions
K traffic = 0; % (cars per metre of tunnel) assumes 10 m per
car, 4 \text{ m car} + 6 \text{ m gap}
%F2
% Friction factor
f = 0.02;
%F4
% Fan data
D fan = 0.84; % diameter (m)
A fan = pi * D fan^2 / 4; %(m^2)
V_fan = 26.2; % output velocity (m/s)
K j = 0.9;
%F5
% Entry loss coefficient due to inlet flow separation
K_{en} = 0.6;
%Ambient properties
p ambient = 101325; %(Pa)
T ambient = 293; %(K)
M air = 29; % (g/mol)
R idealGas = 8.314; % (J/mol.K)
R air = R idealGas / M air * 1000; %(J/kg.K)
rho ambient = p ambient / T ambient / R air; %(kg/m^3)
c p = 1000; %(J/kgK)
h water = spec ent steam(T ambient); %(J/kg) Specific enthalpy
of ambient liquid water
rho water = 1000; %(kg/m^3)
```

```
M_water = 18; % (g/mol)
R_water = R_idealGas / M_water * 1000; % (J/kg.K)
% WMS nozzle properties
% Just used to find water velocity as it enters the tunnel
% Unimportant parameters, negligible influence on the solution
K_nozzle = 4.3; %(1/min/bar^0.5)
p_nozzle = 40; %(bar)
m_nozzle = K_nozzle * p_nozzle^0.5 / 60; %(kg/s)
D_nozzle = 0.005; %(m)
V_nozzle = m_nozzle / rho_water / (pi * D_nozzle^2 / 4); %
water velocity
```

Appendix A4: Critical_ventilation_velocity

```
%Critical ventilation velocity
%dimensionless heat release rate and critical ventilation
velocity
%Q_star = Q / rho_ambient / c_p / T_ambient / 9.81^0.5 /
H_tunnel^2.5;
% assume Q_star > 0.15
Q_star = 0.2;
if Q_star > 0.15
        V_crit_star = 0.43;
else
        V_crit_star = 0.81 * Q_star^(1/3);
end
V_crit = V_crit_star * (9.81 * H_tunnel)^0.5;
```

Appendix A5: HeatReleaseRateAdjustment

```
% Heat Release Rate Adjustment
% Find enhancement factor due to being in an enclosure
if W_HGV / W_tunnel <= 0.5
    psi = 24 * (W HGV / W tunnel)^3 + 1;
else
    'Ratio W HGV / W tunnel is too big'
    'Tunnel enhancement factor psi is inapplicable'
end
Q natural ventilation = psi * Q;
% Find ventilation velocity enhancement factor
if N lanes == 1
    load('oneLane k factors.mat');
    if fire_enhancement_percentile == 1
        V values = oneLanekfactors(:,1);
        enhancement values = oneLanekfactors(:,2);
    elseif fire enhancement percentile == 2
        V values = oneLanekfactors(:,3);
        enhancement values = oneLanekfactors(:,4);
    elseif fire enhancement percentile == 3
        V values = oneLanekfactors(:,5);
        enhancement values = oneLanekfactors(:,6);
    elseif fire enhancement percentile == 4
        V values = oneLanekfactors(:,7);
        enhancement_values = oneLanekfactors(:,8);
    elseif fire enhancement percentile == 5
        V values = oneLanekfactors(:,9);
        enhancement values = oneLanekfactors(:,10);
    elseif fire enhancement percentile == 6
        V_values = oneLanekfactors(:,11);
        enhancement values = oneLanekfactors(:,12);
    end
```

```
elseif N lanes == 2
    load('twoLane k factors.mat');
    if fire enhancement percentile == 1
        V values = twoLanekfactors(:,1);
        enhancement values = twoLanekfactors(:,2);
    elseif fire enhancement percentile == 2
        V values = twoLanekfactors(:,3);
        enhancement values = twoLanekfactors(:,4);
    elseif fire enhancement percentile == 3
        V values = twoLanekfactors(:,5);
        enhancement values = twoLanekfactors(:,6);
    elseif fire enhancement percentile == 4
        V values = twoLanekfactors(:,7);
        enhancement values = twoLanekfactors(:,8);
    elseif fire enhancement percentile == 5
        V_values = twoLanekfactors(:,9);
        enhancement values = twoLanekfactors(:,10);
    elseif fire enhancement percentile == 6
        V values = twoLanekfactors(:,11);
        enhancement_values = twoLanekfactors(:,12);
    end
end
k factor = interpolateCustom(V values, enhancement values,
V upstream);
```

Q_ventilated = Q_natural_ventilation * k_factor;

```
% Updated heat release rate
Q_adjusted = Q_ventilated * k_WMS;
```

Appendix A6: interpolateCustom

```
function k_factor = interpolateCustom(V_values,
enhancement_values, V_upstream)
loopBreaker = 0;
for index = 1 : length(V_values)
    if V_values(index, 1) > V_upstream && loopBreaker == 0
        k_factor = (V_upstream - V_values(index - 1, 1)) /
(V_values(index, 1) - V_values(index - 1, 1)) *
(enhancement_values(index, 1) - enhancement_values(index - 1, 1))
) + enhancement_values(index - 1, 1);
        loopBreaker = 1;
    end
```

end

Appendix A7: InletCV

```
% Inlet CV
T_inlet = T_ambient;
rho_inlet = rho_ambient;
m_air = rho_inlet * A_tunnel * V_inlet;
N_cars_inlet = K_traffic * L_inlet;
F1_inlet = -0.5 * rho_inlet * C_d_car * V_inlet^2 * A_car *
N_cars_inlet;
D_h = (2 * H_tunnel * W_tunnel)/(H_tunnel + W_tunnel);
F2_inlet = -0.5 * f * rho_inlet * L_inlet / D_h * A_tunnel *
V_inlet^2;
F5 = -0.5 * K_en * rho_inlet * A_tunnel * V_inlet^2;
p_inlet_portal = p_ambient - 0.5 * rho_inlet * V_inlet^2;
p_losses_inlet = (F1_inlet + F2_inlet + F5) / A_tunnel;
p_inlet = p_inlet_portal + p_losses_inlet;
```
```
Appendix A8: UpstreamCV
```

```
% Upstream CV
V_upstream = V_inlet;
T_upstream = T_ambient;
rho_upstream = rho_ambient;
N_cars_upstream = K_traffic * L_upstream;
F1_upstream = -0.5 * rho_upstream * C_d_car * V_upstream^2 *
A_car * N_cars_upstream;
F2_upstream = -0.5 * f * rho_upstream * L_upstream / D_h *
A_tunnel * V_upstream^2;
F4 = N_fans * rho_upstream * A_fan * V_fan * (V_fan -
V_upstream) * K_j;
p_fans = F4 / A_tunnel;
p_losses_upstream = (F1_upstream + F2_upstream) / A_tunnel;
p_upstream = p_inlet + p_fans + p_losses_upstream;
```

```
% Fire zone CV
A surface fire zone = 2 * L fire zone * (H tunnel + W tunnel);
%(m^2)
h upstream = spec ent air(T upstream);
m water = densityWMS * N zones * L fire zone * W tunnel *
H tunnel / 60;
E water = m water * (h water + V nozzle<sup>2</sup>/2 + 9.81 *
H tunnel);
E_upstream = m_air * (h_upstream + V_upstream<sup>2</sup>/2 + 9.81 *
H tunnel/2);
eb ConvergenceError = 0.0001;
eb Error = 1;
T fire zone low = 273;
T fire zone high = 2250;
m_fuel = Q_adjusted / H_c_eff_fuel;
m_steam_rad = rad_fraction * Q_adjusted / (spec_ent_steam(373))
- h water);
m water lost = m water - m steam rad;
m steam conv = x evap * m water lost;
m water lost = (1 - x evap) * m water lost;
m steam = m steam rad + m steam conv;
if m steam >= m water
    m_steam = m_water;
end
m downstream = m air + m fuel + m steam;
R mix = m air / (m air + m steam) * R air + m steam / (m air +
m steam) * R water;
while eb Error > eb ConvergenceError
    % Low
    if T fire zone low < 373
        T steam low = 373;
    elseif T fire zone low >= 373
```

```
T steam low = T fire zone low;
    end
    T water lost low = T fire zone low;
    p fire zone low = p upstream + (F1 fire zone low +
F2 fire zone low + F6 fire zone low) / A tunnel;
    rho_fire_zone_low = p_fire_zone_low / R_mix /
T fire zone low;
    V fire zone low = m downstream / rho fire zone low /
A tunnel;
    h fire zone low air = spec ent air(T fire zone low);
    h fire zone low steam = spec ent steam(T steam low);
    h fire zone low = (m air + m fuel) / m downstream *
h fire zone low air + m steam / m downstream *
h fire zone low steam;
    h water_lost_low = min(spec_ent_steam(T_water_lost_low),
419140);
    F1 fire zone low = rho fire zone low/2 * C d HGV * (-
V fire zone low) * V fire zone low * A HGV;
    F2 fire zone low = -1 * f * rho fire zone low / 2 *
L_fire_zone / D_h * A_tunnel * V_fire_zone_low *
abs(V fire zone low);
    F6 fire zone low air = m air * (V upstream -
V fire zone low);
    F6 fire zone low steam = m steam * (0 - V fire zone low);
    F6 fire zone low water = m water lost * (0 -
V fire zone low);
    F6 fire zone low = F6 fire zone low air +
F6 fire zone low steam + F6 fire zone low water;
    Q loss fire zone low = heat losses(T fire zone low,
V_fire_zone_low, D_h, f, A_surface_fire_zone);
    W fire zone = V fire zone low * (F1 fire zone low +
F2 fire zone low + F6 fire zone low);
    E_fire_zone_low = m_downstream * (h_fire_zone_low +
V_fire_zone_low^{2/2} + 9.81 * H tunnel/2);
    E water lost low = m water lost * (h water lost low +
V fire zone low<sup>2</sup>/2);
    bal low = Q adjusted - Q loss fire zone low - W fire zone
+ E upstream + E water - E fire zone low - E water lost low;
    % High
    if T fire zone high < 373
        T steam high = 373;
```

```
elseif T fire zone high >= 373
        T steam high = T fire zone high;
    end
    T water lost high = T fire zone high;
    p fire zone high = p upstream + (F1 fire zone high +
F2 fire zone high + F6 fire zone_high) / A_tunnel;
    rho fire zone high = p fire zone high / R mix /
T fire zone high;
    V fire zone high = m downstream / rho fire zone high /
A tunnel;
    h fire zone high air = spec ent air(T fire zone high);
    h fire zone high steam = spec ent steam(T steam high);
    h fire zone high = (m air + m fuel) / m downstream *
h fire zone high air + m steam / m downstream *
h fire zone high steam;
    h water lost high = min(spec ent steam(T water lost high),
419140);
    F1_fire_zone_high = rho_fire_zone_high/2 * C_d_HGV * (-
V_fire_zone_high) * V_fire_zone_high * A_HGV;
    F2 fire zone high = -1 * f * rho fire zone high / 2 *
L fire zone / D h * A tunnel * V fire zone high *
abs(V_fire_zone_high);
    F6 fire zone high air = m air * (V upstream -
V fire zone high);
    F6 fire zone high steam = m steam * (0 -
V fire zone high);
    F6 fire zone high water = m water lost * (0 -
V fire zone high);
    F6 fire zone high = F6 fire zone high air +
F6_fire_zone_high_steam + F6_fire zone high water;
    Q loss fire zone high = heat losses(T fire zone high,
V_fire_zone_high, D_h, f, A_surface_fire_zone);
    W fire zone = V fire zone high * (F1 fire zone high +
F2 fire zone high + F6 fire zone high);
    E fire zone high = m downstream * (h fire zone high +
V fire zone high^{2/2} + 9.81 * H tunnel/2);
    E water lost high = m water lost * (h water lost high +
V fire zone high<sup>2</sup>/2);
    bal_high = Q_adjusted - Q_loss_fire_zone_high -
W fire zone + E upstream + E water - E fire zone high -
E water lost high;
```

%Middle

T_fire_zone_middle = (T_fire_zone_high + T_fire_zone_low) / 2; if T fire zone middle < 373 T steam middle = 373; elseif T fire zone middle >= 373 T steam middle = T fire zone middle; end T water lost middle = T fire zone middle; p fire zone middle = p upstream + (F1 fire zone middle + F2 fire zone middle + F6 fire zone middle) / A tunnel; rho_fire_zone_middle = p_fire_zone_middle / R_mix / T fire zone middle; V fire zone middle = m downstream / rho fire zone middle / A tunnel; h fire zone middle air = spec ent air(T fire zone middle); h fire zone middle steam = spec ent steam(T steam middle); h_fire_zone_middle = (m_air + m_fuel) / m_downstream * h_fire_zone_middle_air + m_steam / m downstream * h fire zone middle steam; h water lost middle = min(spec_ent_steam(T_water_lost_middle), 419140); F1 fire zone middle = rho fire zone middle/2 * C d HGV * (-V fire zone middle) * V fire zone middle * A HGV; F2 fire zone middle = -1 * f * rho fire zone middle / 2 * L fire zone / D h * A tunnel * V fire zone middle * abs(V fire zone middle); F6 fire zone middle air = m air * (V upstream -V fire zone middle); F6 fire zone middle steam = m steam * (0 -V fire zone middle); F6_fire_zone_middle_water = m_water_lost * (0 -V fire zone middle); F6 fire zone middle = F6 fire zone middle air + F6 fire zone middle steam + F6 fire zone middle water; Q loss fire zone middle = heat losses(T fire zone middle, V fire zone middle, D h, f, A surface fire zone); W fire zone = V fire zone middle * (F1 fire zone middle + F2 fire zone middle + F6 fire zone middle); E fire zone middle = m downstream * (h fire zone middle + V fire zone middle²/2 + 9.81 * H tunnel/2); E water lost middle = m water lost * (h water lost middle + V fire zone middle²/2);

```
bal_middle = Q_adjusted - Q_loss_fire_zone_middle -
W_fire_zone + E_upstream + E_water - E_fire_zone_middle -
E_water_lost_middle;

if bal_low * bal_middle < 0
    T_fire_zone_high = T_fire_zone_middle;
elseif bal_high * bal_middle < 0
    T_fire_zone_low = T_fire_zone_middle;
else
    'temperature convergence error_fire_zone'
    break
end</pre>
```

```
eb_Error = abs((T_fire_zone_high - T_fire_zone_low) /
T_fire_zone_low);
```

```
% Values of temp, density and velocity in fire zone
T fire zone = T fire zone middle; % (K)
p fire zone = p fire zone middle;
rho fire zone = rho fire zone middle;
V_fire_zone = V_fire_zone_middle;
F1 fire zone = F1 fire zone middle;
F2_fire_zone = F2 fire zone middle;
F6 = F6 fire zone middle;
% Put the above values into arrays
% These arrays will give the distribution of each parameter
throughout the
% downstream section of the tunnel
T downstream(1, 1) = T fire zone;
rho downstream(1, 1) = rho fire zone;
V downstream(1, 1) = V fire zone;
p_downstream(1, 1) = p_fire_zone;
F_1_downstream(1, 1) = F1_fire_zone;
F 2 downstream(1, 1) = F2 fire zone;
E \text{ downstream}(1, 1) = E \text{ fire zone middle};
Q losses(1, 1) = Q loss fire zone middle;
```

Appendix A10: heat_losses

```
% Heat losses function
function Q losses = heat losses(T gas, V gas, D h, f,
A_surface_CV)
    % Load air data for kinematic viscosity, Pr, thermal
conductivity and
    % thermal diffusivity
    load('flow data air.mat');
    % Interpolate for kinematic viscosity of air
    T values = flowdataair(:,1);
    visc kin values = flowdataair(:,2);
    visc kin = interp1(T values, visc kin values, T gas);
    % Interpolate for Prandtl number of air
    Pr values = flowdataair(:,3);
    Pr = interp1(T_values, Pr_values, T_gas);
    % Interpolate for thermal conductivity of air
    k values = flowdataair(:,4);
    k_air = interp1(T_values, k_values, T_gas);
    % Find Reynolds number
    Re = V gas * D h / visc kin;
    % If statement to find which turbulence regime to use
    % Find Nusselt number accordingly
    if Re > 10000 % turbulent
        Nu = 0.125 * f * Re * Pr^{(1/3)};
    elseif Re <= 10000 % laminar</pre>
        Nu = 3;
    end
    % Find convective coefficient
    h = Nu * k_air / D_h;
    % Set convergence parameters for surface temperature
    surfaceError = 1;
    surfaceConvergence = 0.01;
    % Set initial time value and time step
    t surface = 0;
    delta t surface = 60; % seconds
```

```
% Set ambient tunnel wall surface temperature
    T surface ambient = 293; % K
    % Define concrete properties
    alpha conc = 0.0000005; % m2/s
    k conc = 1; % W/mK
    while surfaceError > surfaceConvergence
        % Calculate surface temperature at timestep i
        T surface = T surface ambient + (T gas -
T_surface_ambient) * (1 - exp(alpha_conc * t_surface * h^2 /
k conc^2) * erfc(sqrt(alpha conc * t surface) * h / k conc));
        % Increase time to timestep i+1
        t surface = t surface + delta t surface;
        % Calculate surface temperature at timestep i+1
        T surface new = T surface ambient + (T gas -
T_surface_ambient) * (1 - exp(alpha_conc * t_surface * h^2 /
k_conc^2) * erfc(sqrt(alpha_conc * t_surface) * h / k_conc));
        % Check if surface temperature has converged
        surfaceError = abs(T surface / T surface new - 1);
     end
    Q_losses = h * A_surface_CV * (T_gas - T_surface_new);
```

Appendix A11: spec_ent_air

```
function h=spec_ent_air(T)
    load('thermo_data_air.mat');
    T_values = thermodataair(:,1);
    h_values = thermodataair(:,2);
    h = interpl(T_values, h_values, T);
end
```

Appendix A12: spec_ent_steam

```
function h=spec_ent_steam(T)
    load('thermo_data_steam.mat');
    T_values = thermodatawaterandsteam(:,1);
    h_values = thermodatawaterandsteam(:,2);
    h = interpl(T_values, h_values, T);
end
```

Appendix A13: DownstreamCVs

```
% Downstream energy balances
L CV = L downstream / N CV;
A_surface_CV = 2 * L_CV * (H_tunnel + W_tunnel); %(m^2)
N cars downstream = L downstream * K traffic;
eb ConvergenceError = 0.0001;
eb Error = 1;
for iii = 1 : N CV
    p_previous_CV = p_downstream(iii, 1);
    T CV low = 273;
    T CV high = 2250;
    E_previous_CV = E_downstream(iii, 1);
    eb Error = 1;
    F1 CV low = 0;
    F2 CV low = 0;
    F1_CV_high = 0;
    F2 CV high = 0;
    F1 CV middle = 0;
    F2 CV middle = 0;
    while eb Error > eb ConvergenceError
        % Low
        if T_CV_low < 373
            T steam low = 373;
        else
            T steam low = T CV low;
        end
        p_CV_low = p_previous_CV + (F1_CV_low + F2 CV low) /
A tunnel;
        rho_CV_low = p_CV_low / R_mix / T_CV low;
        V CV low = m downstream / rho CV low / A tunnel;
        h CV low air = spec ent air(T CV low);
        h_CV_low_steam = spec_ent_steam(T_steam_low);
        h CV low = (m air + m fuel) / m downstream *
```

```
h CV low air + m steam / m downstream * h CV low steam;
        F1 CV low = -0.5 * rho CV low * C d car * V CV low<sup>2</sup> *
A car * N cars downstream / N CV;
        F2 CV low = -0.5 * f * rho CV low * L CV / D h *
A_tunnel * V_CV_low * abs(V CV low);
        Q loss low CV = heat losses(T CV low, V CV low, D h,
f, A surface CV);
        W_CV_low = V_CV_low * (F1_CV_low + F2_CV_low);
        E CV low = m downstream * (h CV low + V CV low<sup>2</sup>/2 +
9.81 * H tunnel/2);
        bal low = -Q loss low CV - W CV low + E previous CV -
E CV low;
        % High
        if T CV high < 373
            T steam high = 373;
        else
            T_steam_high = T_CV_high;
        end
        p_CV_high = p_previous_CV + (F1_CV_high + F2_CV_high)
/ A tunnel;
        rho CV high = p CV high / R mix / T CV high;
        V CV high = m downstream / rho CV_high / A_tunnel;
        h CV high air = spec ent air(T CV high);
        h CV high steam = spec ent steam(T steam high);
        h CV high = (m air + m fuel) / m downstream *
h CV high air + m steam / m downstream * h CV high steam;
        F1 CV high = -0.5 * rho CV high * C d car *
V_CV_high^2 * A_car * N_cars_downstream / N_CV;
        F2 CV high = -0.5 * f * rho CV high * L CV / D h *
A tunnel * V CV high^2;
        Q loss high CV = heat losses(T CV high, V CV high,
D h, f, A surface CV);
        W_CV_high = V_CV_high * (F1_CV_high + F2_CV_high);
        E CV high = m downstream * (h CV high + V CV high^{2/2}
+ 9.81 * H_tunnel/2);
        bal high = -Q loss high CV - W CV high +
E_previous_CV - E CV high;
```

```
% Middle
        T CV middle = (T CV high + T CV low) / 2;
        if T CV middle < 373
            T steam middle = 373;
        else
            T steam_middle = T_CV_middle;
        end
        p CV middle = p previous CV + (F1 CV middle +
F2_CV_middle) / A tunnel;
        rho CV middle = p CV middle / R mix / T CV middle;
        V CV middle = m downstream / rho CV middle / A tunnel;
        h CV middle air = spec ent air(T CV middle);
        h CV middle steam = spec ent steam(T steam middle);
        h CV middle = (m air + m fuel) / m downstream *
h CV middle air + (m steam) / m downstream *
h CV middle steam;
        F1_CV_middle = -0.5 * rho_CV_middle * C_d_car *
V_CV_middle^2 * A_car * N_cars_downstream / N_CV;
        F2 CV middle = -0.5 * f * rho CV middle * L CV / D h *
A tunnel * V CV middle * abs(V CV middle);
        E CV middle = m downstream * (h CV middle +
V CV middle<sup>2</sup>/2 + 9.81 * H tunnel/2);
        W CV middle = V CV middle * (F1 CV middle +
F2 CV middle);
        Q loss middle CV = heat losses(T CV middle,
V CV middle, D h, f, A surface CV);
        bal middle = -Q loss middle CV - W CV middle +
E previous CV - E CV middle;
        if bal low * bal middle < 0</pre>
            T CV high = T CV middle;
        elseif bal high * bal middle < 0</pre>
            T CV low = T CV middle;
        else
            'temperature convergence error'
            'downstream energy balances'
            break
        end
        eb_Error = abs((T_CV_high - T_CV low) / T CV low);
```

```
p_CV = p_CV_middle;
T_CV = T_CV_middle;
rho_CV = rho_CV_middle;
V_CV = V_CV_middle;
T_downstream(1 + iii, 1) = T_CV;
rho_downstream(1 + iii, 1) = rho_CV;
V_downstream(1 + iii, 1) = V_CV;
p_downstream(1 + iii, 1) = p_CV;
F_1_downstream(1 + iii, 1) = F1_CV_middle;
F_2_downstream(1 + iii, 1) = F2_CV_middle;
E_downstream(1 + iii, 1) = E_CV_middle;
Q_losses(1 + iii, 1) = Q_loss_middle_CV;
```

Appendix A14: ForceBalance

```
% Force balance
% Vehicle drag force
F1_downstream = sum(F_1_downstream(:, 1));
F1 = F1 inlet + F1 upstream + F1 downstream;
% Wall friction
F2_downstream = sum(F_2_downstream(:, 1));
F2 = F2 inlet + F2 upstream + F2 downstream;
% Force due to pressure difference between inlet and outlet
F3 = -0.5 * rho_inlet * V_inlet^2 * A_tunnel;
% Force balance
F balance = F1 + F2 + F3 + F4 + F5 + F6
if stopAddingFans == 0
    if F balance < 0
        N fans = N fans + 1;
    elseif F_balance > 0
        stopAddingFans = 1;
    end
end
if stopAddingFans == 1
    if F balance > 50
        V inlet = V inlet * 1.01;
    elseif F_balance >= 10 && F_balance <= 50</pre>
        V inlet = V inlet * 1.001;
    elseif F balance < 10</pre>
        V inlet = V inlet * 1.0001;
    elseif F balance < -10</pre>
        V inlet = V inlet * 0.999;
    elseif F_balance > -10 && F_balance <= 0</pre>
```

```
V_inlet = V_inlet * 0.9999;
```

```
if abs(F_balance) < 1</pre>
```

stopIterations = 1;

end

Appendix B: Example input values

Parameter	Value (unit)	Ref.
HRR in open, still, ambient conditions	User defined (W)	-
Tunnel height	User defined (m)	-
Tunnel width	User defined (m)	-
Tunnel cross sectional area	User defined (m ²)	-
Tunnel length	User defined (m)	-
Number of lanes	User defined; either 1 or 2 (-)	[19]
Forced ventilation enhancement factor	User defined; either 10% , 30% ,	[19]
percentile	50%, expected, 70%, or 90% (-)	[(]
Length of water mist system zone	30 (m)	[6]
Number of water mist system zones	User defined (-)	-
Water mist system mass flux density	$0.5 (1/min/m^3)$	[6]
HRR reduction due to WMS	0.35 (-)	[6]
Proportion of water evaporating	User defined (-)	-
Length of inlet section	User defined (m)	-
Length of tunnel upstream of fire zone	User defined (m)	-
Length of tunnel downstream of fire zone	User defined (m)	-
Number of CVs the downstream section is divided into	User defined (-)	-
Radiative fraction	0.35 (-)	[12]
Effective heat of combustion of fuel load in tunnel	20 × 10 ⁶ (J/kg)	[43]
HGV height	User defined (m)	-
HGV width	User defined (m)	-
HGV drag coefficient	0.4 (-)	[49]
Car height	1.5 (m)	-
Car width	2 (m)	-
Car drag coefficient	0.3 (-)	[49]
Traffic density	15 (vehicles/lane/km)	[49]
Tunnel wall friction factor	0.02 (-)	[28]
Fan diameter	0.84 (m)	[50]
Fan outlet velocity	26.2 (m/s)	[50]
Fan pressure rise coefficient	0.9 (-)	[28]
Tunnel flow inlet separation factor	0.6 (-)	[28]

Appendix C: Tabulated k values for HRR enhancement due to forced ventilation [19] Part 1: One lane. Expected, 10%, 30%

Expected		10%		30%	
velocity (m/s)	k	velocity (m/s)	k	velocity (m/s)	k
0	1	0	1	0	1
0.245	1.139	0.130	1.064	0.375	1.220
0.708	1.510	0.245	1.108	0.462	1.272
0.578	1.410	0.375	1.171	0.549	1.334
0.448	1.307	0.578	1.280	0.693	1.429
0.852	1.617	0.693	1.342	0.809	1.520
1.025	1.768	0.809	1.399	0.939	1.617
1.155	1.894	0.924	1.467	1.083	1.719
1.300	2.028	1.069	1.550	1.213	1.816
1.444	2.157	1.170	1.604	1.357	1.932
1.603	2.310	1.329	1.694	1.502	2.040
1.747	2.457	1.487	1.776	1.675	2.155
1.921	2.595	1.690	1.876	1.863	2.275
2.094	2.740	1.877	1.953	2.051	2.370
2.296	2.834	2.065	2.020	2.195	2.418
2.498	2.931	2.253	2.075	2.325	2.467
2.700	2.989	2.498	2.116	2.426	2.483
2.888	3.049	2.729	2.143	2.700	2.531
3.090	3.088	2.931	2.155	3.004	2.545
3.292	3.150	3.119	2.153	3.191	2.542
3.495	3.191	3.379	2.150	3.422	2.557
3.682	3.254	3.596	2.148	3.653	2.571
3.899	3.365	3.798	2.161	3.870	2.604
4.087	3.481	4.000	2.173	4.087	2.675
4.245	3.600	4.144	2.202	4.289	2.766
4.448	3.749	4.332	2.231	4.419	2.842
4.592	3.906	4.433	2.261	4.549	2.940
4.780	4.124	4.592	2.306	4.664	3.021
4.968	4.355	4.736	2.353	4.765	3.104
5.112	4.600	5.040	2.466	4.881	3.212
5.271	4.825	5.184	2.516	5.040	3.369
5.430	5.060	5.401	2.620	5.170	3.509
5.617	5.344	5.690	2.746	5.300	3.656
5.791	5.643	5.819	2.802	5.487	3.861
5.949	5.878	6.007	2.878	5.704	4.133
6.123	6.122	6.166	2.936	5.863	4.335
6.310	6.376	6.383	3.015	6.094	4.608
6.484	6.641	6.700	3.116	6.296	4.833
6.657	6.869	6.859	3.157	6.455	4.999
6.845	7.055	7.134	3.219	6.585	5.136
7.018	7.247	7.292	3.261	6.816	5.348
7.220	7.443	7.509	3.280	6.989	5.493

7.408	7.592	7.697	3.323	7.148	5.604
7.610	7.690	7.957	3.341	7.292	5.717
7.812	7.735	8.303	3.358	7.408	5.794
8.000	7.781	8.606	3.376	7.552	5.870
8.188	7.828	8.953	3.370	7.711	5.947
8.390	7.765	9.256	3.365	7.856	6.026
8.578	7.758	9.430	3.362	8.072	6.103
8.780	7.696	9.661	3.358	8.260	6.182
8.968	7.636	9.747	3.356	8.419	6.220
9.170	7.523			8.578	6.258
9.357	7.412			8.780	6.295
9.545	7.353			9.054	6.330
9.718	7.296			9.213	6.324
				9.401	6.362
				9.545	6.357
				9.747	6.351

Appendix C: Tabulated k values for HRR enhancement due to forced ventilation [19] Part 2: One lane. 50%, 70%, 90%

50%		70%		90%	
velocity (m/s)	k	velocity (m/s)	k	velocity (m/s)	k
0	1	0	1	0	1
0.549	1.391	0.159	1.101	0.159	1.116
0.347	1.237	0.361	1.263	0.289	1.221
0.116	1.071	0.505	1.420	0.448	1.372
0.245	1.147	0.664	1.553	0.563	1.490
0.708	1.531	0.838	1.734	0.736	1.675
0.823	1.640	1.083	1.989	0.953	1.936
0.982	1.781	1.227	2.160	1.083	2.102
1.126	1.907	1.430	2.410	1.242	2.314
1.256	2.043	1.603	2.599	1.444	2.601
1.415	2.188	1.776	2.802	1.588	2.805
1.560	2.327	1.978	3.000	1.819	3.152
1.661	2.408	2.137	3.147	1.978	3.376
1.762	2.491	2.325	3.277	2.065	3.493
1.906	2.613	2.614	3.435	2.181	3.639
2.036	2.703	2.440	3.344	2.325	3.817
2.181	2.777	2.801	3.504	2.498	4.031
2.412	2.892	3.004	3.574	2.643	4.170
2.556	2.930	3.235	3.645	2.830	4.373
2.758	2.968	3.451	3.717	3.032	4.554
2.917	2.986	3.625	3.792	3.264	4.775
3.090	3.004	3.769	3.868	3.495	5.006
3.264	3.022	3.928	3.974	3.740	5.285
3.422	3.041	4.072	4.111	4.014	5.696
3.639	3.080	4.188	4.224	4.245	6.097
3.884	3.162	4.390	4.491	4.404	6.440
4.087	3.270	4.505	4.679	4.621	6.942
4.289	3.429	4.650	4.942	4.823	7.536
4.477	3.596	4.780	5.184	5.011	8.069
4.664	3.824	4.924	5.514	5.285	9.003
4.823	4.039	5.126	5.986	5.415	9.444
4.982	4.265	5.314	6.499	5.617	10.181
5.170	4.599	5.502	7.007	5.762	10.680
5.343	4.924	5.690	7.554	5.935	11.278
5.458	5.130	5.877	8.089	6.209	12.155
5.617	5.418	6.051	8.542	6.455	12.831
5.776	5.762	6.253	9.083	6.614	13.181
5.935	6.086	6.440	9.525	6.888	13.626
6.051	6.297	6.643	9.919	7.177	13.989
6.238	6.649	6.874	10.328	7.350	14.073
6.455	7.021	7.076	10.607	7.552	14.156
6.657	7.311	7.249	10.820	7.711	14.144

6.787	7.511	7.422	10.961	7.913	14.130
6.917	7.664	7.552	11.030	8.101	14.018
7.134	7.925	7.971	11.160	8.404	13.708
7.307	8.085	7.740	11.096	8.693	13.406
7.523	8.245	8.245	11.067	9.083	12.834
7.711	8.352	8.477	10.977	9.386	12.377
8.029	8.514	8.693	10.814	9.661	11.938
8.404	8.556	8.910	10.653	9.762	11.768
8.664	8.545	9.170	10.420		
8.881	8.535	9.329	10.268		
9.155	8.464	9.545	10.045		
9.458	8.392	9.747	9.828		
9.675	8.325				

Appendix C: Tabulated k values for HRR enhancement due to forced ventilation [19] Part 3: Two lane. Expected, 10%, 30%

Expected		10%		30%	
velocity (m/s)	k	velocity (m/s)	k	velocity (m/s)	k
0.002	0.993	0.002	0.993	0.002	0.997
0.204	1.038	0.250	1.003	0.328	1.020
0.360	1.067	0.436	1.013	0.530	1.038
0.516	1.104	0.669	1.023	0.809	1.063
0.671	1.139	0.933	1.033	1.073	1.088
0.827	1.178	1.166	1.044	1.353	1.122
0.983	1.215	1.522	1.058	1.695	1.168
1.139	1.253	1.864	1.064	1.897	1.201
1.294	1.297	2.220	1.071	2.146	1.247
1.450	1.337	2.608	1.074	2.348	1.285
1.621	1.379	2.872	1.081	2.535	1.326
1.777	1.427	3.136	1.092	2.722	1.372
1.917	1.467	3.384	1.110	2.940	1.429
2.073	1.512	3.648	1.141	3.096	1.474
2.229	1.565	3.726	1.156	3.329	1.541
2.384	1.608	3.944	1.197	3.469	1.589
2.540	1.659	4.131	1.247	3.781	1.690
2.696	1.716	4.193	1.264	4.061	1.791
2.852	1.770	4.302	1.303	4.373	1.911
3.007	1.819	4.536	1.386	4.622	1.998
3.163	1.882	4.677	1.450	4.887	2.103
3.319	1.934	4.864	1.537	5.136	2.198
3.475	1.988	4.973	1.596	5.431	2.306
3.646	2.043	5.098	1.657	5.696	2.402
3.817	2.107	5.255	1.745	6.131	2.528
3.973	2.165	5.333	1.787	6.566	2.624
4.144	2.225	5.442	1.856	7.032	2.695
4.315	2.287	5.536	1.907	7.389	2.721
4.486	2.358	5.676	1.987	7.761	2.738
4.642	2.415	5.848	2.085	8.117	2.727
4.828	2.473	6.019	2.180	8.582	2.706
4.999	2.533	6.097	2.218	8.970	2.677
5.201	2.594	6.315	2.326	9.434	2.629
5.388	2.647	6.502	2.399	9.682	2.592
5.574	2.702	6.891	2.525	9.976	2.555
5.776	2.748	7.248	2.594	10.100	2.537
5.947	2.785	7.527	2.619		
6.180	2.842	7.837	2.627		
6.366	2.871	8.147	2.608		
6.584	2.910	8.333	2.589		
6.785	2.929	8.519	2.561		
7.003	2.948	8.720	2.534		

7.204	2.967		8.983	2.473
7.422	2.987		9.246	2.421
7.639	2.996		9.664	2.313
7.856	3.005	:	10.004	2.226
8.073	3.004	-	10.097	2.202
8.290	3.003			
8.507	2.991			
8.724	2.990			
8.941	2.979			
9.158	2.957			
9.375	2.946			
9.592	2.935			
9.809	2.914			
10.010	2.902			

Appendix C: Tabulated k values for HRR enhancement due to forced ventilation [19] Part 4: Two lane. 50%, 70%, 90%

50%		70%		90%	
velocity (m/s)	k	velocity (m/s)	k	velocity (m/s)	k
0.002	0.997	0.002	0.993	0.002	0.993
0.359	1.042	0.111	1.017	0.095	1.024
0.546	1.070	0.220	1.042	0.158	1.046
0.841	1.119	0.329	1.063	0.251	1.075
1.122	1.170	0.469	1.097	0.361	1.116
1.339	1.214	0.609	1.131	0.564	1.191
1.573	1.270	0.734	1.162	0.720	1.254
1.744	1.314	0.921	1.211	0.861	1.311
1.994	1.388	1.077	1.258	1.017	1.385
2.181	1.451	1.264	1.315	1.157	1.448
2.399	1.533	1.373	1.351	1.314	1.525
2.555	1.602	1.576	1.423	1.423	1.584
2.789	1.704	1.716	1.477	1.532	1.639
2.976	1.794	1.841	1.523	1.626	1.696
3.148	1.882	1.903	1.555	1.751	1.767
3.257	1.935	1.981	1.587	1.892	1.854
3.397	2.009	2.090	1.643	2.032	1.952
3.600	2.115	2.200	1.694	2.142	2.027
3.756	2.189	2.293	1.747	2.220	2.084
3.912	2.265	2.356	1.778	2.485	2.286
4.083	2.343	2.465	1.840	2.595	2.374
4.238	2.408	2.559	1.897	2.673	2.440
4.456	2.492	2.699	1.984	2.767	2.517
4.736	2.578	2.809	2.053	3.001	2.714
4.969	2.640	2.949	2.147	3.141	2.828
5.280	2.703	3.105	2.245	3.282	2.947
5.482	2.740	3.230	2.331	3.391	3.049
5.777	2.786	3.324	2.396	3.563	3.177
6.040	2.814	3.448	2.471	3.687	3.265
6.211	2.832	3.589	2.565	3.781	3.333
6.397	2.841	3.745	2.664	3.952	3.437
6.599	2.860	3.838	2.719	4.107	3.520
6.832	2.869	3.978	2.795	4.201	3.556
7.033	2.877	4.134	2.872	4.325	3.605
7.312	2.896	4.320	2.941	4.480	3.641
7.653	2.894	4.523	3.012	4.713	3.678
8.134	2.902	4.662	3.043	4.899	3.702
8.475	2.900	4.818	3.073	5.085	3.701
8.801	2.899	5.050	3.115	5.302	3.687
9.142	2.887	5.500	3.134	5.550	3.660
9.374	2.876	5.702	3.144	5.876	3.633
9.700	2.864	6.043	3.142	6.170	3.593

9.948	2.853	6.29	1 3.141	6.309	3.580
10.103	2.843	6.64	8 3.139	6.480	3.567
		6.80	3 3.138	6.635	3.566
		7.00	4 3.137	6.774	3.565
		7.17	5 3.147	6.929	3.552
		7.34	5 3.146	7.053	3.551
		7.68	6 3.155	7.146	3.550
		7.93	4 3.153	7.255	3.550
		8.15	2 3.163	7.441	3.561
		8.41	5 3.162	7.581	3.560
		8.64	8 3.171	7.705	3.559
		9.23	7 3.179	7.860	3.571
		9.70	2 3.187	7.999	3.570
		9.96	6 3.197	8.170	3.581
		5.28	3 3.124	8.294	3.580
				8.402	3.592
				8.682	3.603
				8.961	3.626
				9.209	3.649
				9.457	3.660
				9.721	3.697
				9.845	3.696
				9.985	3.708
				10.093	3.720

Appendix D: Flow data for air [45]

		viscosity				thermal diffusivity
Т (К)		(m^2/s)	Pr		k (W/mK)	(m^2/s)
	100	1.92E-06		0.768	0.009246	2.50E-06
	150	4.34E-06		0.756	0.013735	5.74E-06
	200	7.51E-06		0.739	0.01809	1.02E-05
	250	1.05E-05		0.722	0.02227	1.57E-05
	300	1.68E-05		0.708	0.02624	2.22E-05
	350	2.08E-05		0.697	0.03003	2.98E-05
	400	2.59E-05		0.689	0.03365	3.76E-05
	450	3.17E-05		0.683	0.03707	4.64E-05
	500	3.79E-05		0.68	0.04038	5.56E-05
	550	4.43E-05		0.68	0.0436	6.53E-05
	600	5.13E-05		0.682	0.04659	7.51E-05
	650	5.85E-05		0.682	0.04953	8.58E-05
	700	6.63E-05		0.684	0.0523	9.67E-05
	750	7.39E-05		0.686	0.05509	0.00010774
	800	8.23E-05		0.689	0.05779	0.00011951
	850	9.08E-05		0.692	0.06028	0.00013097
	900	9.93E-05		0.696	0.06279	0.00014271
	950	0.0001082		0.699	0.06525	0.0001551
	1000	0.0001178		0.702	0.06752	0.00016779
	1100	0.0001386		0.704	0.0732	0.0001969
	1200	0.0001591		0.707	0.0782	0.0002251
	1300	0.0001821		0.705	0.0837	0.0002583
	1400	0.0002055		0.705	0.0891	0.000292
	1500	0.0002291		0.705	0.0946	0.0003266
	1600	0.0002545		0.705	0.1	0.0003624
	1700	0.0002809		0.705	0.105	0.0003977
	1800	0.0003081		0.704	0.111	0.0004379
	1900	0.0003385		0.704	0.117	0.0004811
	2000	0.000369		0.702	0.124	0.000526
	2100	0.0003996		0.703	0.131	0.000568
	2200	0.0004326		0.707	0.139	0.0006115
	2300	0.000464		0.71	0.149	0.0006537
	2400	0.000504		0.718	0.161	0.0007016
	2500	0.000543		0.73	0.175	0.0007437

Appendix E: Specific enthalpy values of air [44]

Т (К)		h (J/kg)	550	554740	1120	1184280
	200	199970	560	565170	1140	1207570
	210	209970	570	575590	1160	1230920
	220	219970	580	586040	1180	1254340
	230	230020	590	596520	1200	1277790
	240	240020	600	607020	1220	1301310
	250	250050	610	617530	1240	1324930
	260	260090	620	628070	1260	1348550
	270	270110	630	638630	1280	1372240
	280	280130	640	649220	1300	1395970
	285	285140	650	659840	1320	1419760
	290	290160	660	670470	1340	1443600
	295	295170	670	681140	1360	1467490
	300	300190	680	691820	1380	1491440
	305	305220	690	702520	1400	1515420
	310	310240	700	713270	1420	1539440
	315	315270	710	724040	1440	1563510
	320	320290	720	734820	1460	1587630
	325	325310	730	745620	1480	1611790
	330	330340	740	756440	1500	1635970
	340	340420	750	767290	1520	1660230
	350	350490	760	778180	1540	1684510
	360	360580	770	789110	1560	1708820
	370	370670	780	800030	1580	1733170
	380	380770	790	810990	1600	1757570
	390	390880	800	821950	1620	1782000
	400	400980	820	843980	1640	1806460
	410	411120	840	866080	1660	1830960
	420	421260	860	888270	1680	1855500
	430	431430	880	910560	1700	1880100
	440	441610	900	932930	1750	1941600
	450	451800	920	955380	1800	2003300
	460	462020	940	977920	1850	2065300
	470	472240	960	1000550	1900	2127400
	480	482490	980	1023250	1950	2189700
	490	492740	1000	1046040	2000	2252100
	500	503020	1020	1068890	2050	2314600
	510	513320	1040	1091850	2100	2377400
	520	523630	1060	1114860	2150	2440300
	530	533980	1080	1137890	2200	2503200
	540	544350	1100	1161070	2250	2566400

Appendix F: Specific enthalpy values of steam [48]

T (K)		h (J/kg)	443	2816200	763	3476400
	273	60	448	2826100	773	3488700
	278	21120	453	2836000	793	3531600
	283	42120	458	2845800	813	3574700
	288	63080	463	2855700	833	3618000
	293	84010	468	2865600	853	3661700
	298	104920	473	2875500	873	3705600
	303	125820	483	2895200	893	3749800
	308	146720	493	2915000	913	3794300
	313	167620	503	2934800	933	3839000
	318	188510	513	2954600	953	3884000
	323	209420	523	2974500	973	3929400
	328	230330	533	2994400	993	3975000
	333	251250	543	3014400	1013	4020900
	338	272180	553	3034400	1033	4067000
	343	293120	563	3054400	1053	4113500
	348	314080	573	3074500	1073	4160200
	353	335050	583	3094700	1093	4207200
	358	356050	593	3114900	1113	4254500
	363	377060	603	3135100	1133	4302100
	368	398100	613	3155500	1153	4349900
	373	2675800	623	3175800	1173	4398000
	378	2686100	633	3196300	1193	4446400
	383	2696300	643	3216700	1213	4495000
	388	2706500	653	3237300	1233	4543900
	393	2716600	663	3257900	1253	4593100
	398	2726700	673	3278600	1273	4642600
	403	2736700	683	3299300	1373	4893500
	408	2746700	693	3320100	1473	5150600
	413	2756700	703	3340900	1573	5413200
	418	2766700	713	3361900	1673	5681200
	423	2776600	723	3382800	1773	5953900
	428	2786500	733	3403900	1873	6231000
	433	2796400	743	3425000	2073	6797200
	438	2806300	753	3446200	2273	7377000